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STOCHASTIC INTEGRATION WITH RESPECT TO A CYLINDRICAL SPECIAL SEMI-MARTINGALE

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ABSTRACT. In this research, we introduce the stochastic integration with respect to a cylindrical special semi-martingale, which is a specific case of general integration, with specific properties of special semi-martingales.

1. INTRODUCTION

Cylindrical semi-martingales play a key role in application, specially in stochastic partial differential equations. Among the wide class of cylindrical semi-martingales, cylindrical Brownian motions are used widely as models in stochastic analysis \mathbb{S} , $5, 8, 9, 11, 14, 18, 19$ $5, 8, 9, 11, 14, 18, 19$ $5, 8, 9, 11, 14, 18, 19$ $5, 8, 9, 11, 14, 18, 19$ $5, 8, 9, 11, 14, 18, 19$ $5, 8, 9, 11, 14, 18, 19$ $5, 8, 9, 11, 14, 18, 19$. Although Brownian motions work as good models, motivation of using other classes of cylindrical semi-martingales appears in recent research. Interesting examples of such a view can be found in $\left|\frac{1}{2}, 0, 12, 13, 15\right|$ $\left|\frac{1}{2}, 0, 12, 13, 15\right|$ $\left|\frac{1}{2}, 0, 12, 13, 15\right|$ $\left|\frac{1}{2}, 0, 12, 13, 15\right|$. In spite of the fact that most of the past articles have an applied view to extend the concepts and utilities the stochastic integration, none of these works considers stochastic integration with respect to cylindrical special semi-martingales.

In this work, our main objective is to introduce a theory of stochastic integration for cylindrical special semi-martingales, which are a particular family of semi-martingales with complex behavior in relation with the measure of the space, defined on. P is a special semi-martingale if P can be decomposed into $P = M + A$ where M is a local martingale and A a process with predictable finite variation, with $A_0 = 0$. Such a decomposition is then unique and is called canonical decomposition.

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On the other hand, for a Banach space \mathscr{X} , the cylindrical σ -algebra is defined to be the coarsest σ -algebra, i.e. the one with the fewest measurable sets, such that every continuous linear function on $\mathscr X$ is a measurable function. That is important to note that in general, the cylindrical σ -algebra is not the same as the Borel σ algebra on \mathscr{X} , which is the coarsest σ -algebra that contains all open subsets of $\mathscr X.$

In the following, we study the cylindrical special semi-martingale $M: \mathcal{X}^* \to$ S^{SP} from the dual of a separable Banach space $\mathscr X$ to the space of special semimartingales. Moreover, we define the integral of a progressive process with respect to a cylindrical special semi-martingale.

2. Preliminaries

Let \mathscr{X}, \mathscr{Y} be two Banach spaces. We will denote the space of all bilinear operators from $\mathscr{X} \times \mathscr{Y}$ to R as $\mathscr{B}(\mathscr{X}, \mathscr{Y})$. Note that for a continuous $b \in \mathscr{B}(\mathscr{X}, \mathscr{Y})$ there exists an operator $\mathcal{B} \in \mathcal{L}(\mathcal{X}, \mathcal{Y}^*)$ such that

$$
b(x, y) = \langle \mathcal{B}x, y \rangle = \mathcal{B}x(y), \quad x \in X, y \in Y. \tag{1}
$$

An operator $\mathcal{B}: \mathcal{X} \to \mathcal{X}^*$ is called self-adjoint, if for each $x, y \in \mathcal{X}$

$$
\langle \mathcal{B}x, y \rangle = \langle \mathcal{B}y, x \rangle.
$$

and is called positive, if B is self-adjoint and $\mathcal{B}_x(x) = \langle \mathcal{B}x, x \rangle \ge 0$ for all $x \in \mathcal{X}$.

Recall that if $\mathcal{B}: \mathscr{X} \to \mathscr{X}^*$ is a positive self-adjoint operator, then the Cauchy-Schwartz inequality holds for the bilinear form $\langle \mathcal{B}x, y \rangle$. In a natural way in functional analysis, the norm of β is defined as

$$
\|\mathcal{B}\| = \sup_{x \in \mathcal{X}, \|x\| = 1} |\langle \mathcal{B}x, x \rangle| \tag{2}
$$

Note that if $\mathscr X$ is a Hilbert space, then $[2]$ would be coincides with the induced norm of the inner product defined on \mathscr{X} .

Let $(\Omega, \mathscr{F}, \mu)$ be a measure space and \mathscr{X} a Banach space. A function $f : \Omega \to \mathscr{X}$ is called simple if there exist $x_1, x_2, \ldots x_n \in \mathscr{X}$ and $E_1, E_2, \ldots, E_n \in \mathscr{F}$ such that $f = \sum_{i=1}^n x_i \chi_{E_i}$, where $\chi_{E_i}(\omega) = 1$ if $\omega \in E_i$ and $\chi_{E_i}(\omega) = 0$ if $\omega \neq E_i$. A function $f : \Omega \to \mathscr{X}$ is called strong measurable if there exists a sequence of simple functions (f_n) with $\lim_n ||f_n-f|| = 0$, μ -almost everywhere. A function $f : \Omega \to \mathcal{X}$ is called scalar measurable if for each $x^* \in \mathcal{X}^*$ the numerical function $x^* f$ is strong measurable.

Further we will need the following lemma.

Lemma 1. [15], Proposition 32] Let (S, Σ) be a measurable space, H be a separable Hilbert space, $f : S \to \mathcal{L}(\mathcal{H})$ be a scalar measurable self-adjoint operator-valued function. Let $F : \mathbb{R} \to \mathbb{R}$ be locally bounded measurable. Then $F(f) : S \to \mathscr{L}(\mathscr{H})$ is a scalar measurable self-adjoint operator-valued function.

That is trivial to think about the square root of a positive operator. It would be appreciated if the square root drops us in to a Hilbert space, even in a special case.

Lemma 2. [19], Lemma 2.4] Let \mathcal{X} be a reflexive separable Banach space, \mathcal{B} : $\mathscr{X} \to \mathscr{X}^*$ be a positive operator. Then there exists a separable Hilbert space \mathscr{H} and an operator $\mathcal{B}^{1/2}$: $\mathscr{X} \to \mathscr{H}$ such that $\mathcal{B} = \mathcal{B}^{1/2*}\mathcal{B}^{1/2}$.

A scalar-valued process M is called a continuous local martingale if there exists a sequence of stopping times $(\tau_n)_{n>1}$ such that $\tau_n \uparrow \infty$ almost surely as $n \to \infty$ and $1_{\tau_n>0}M^{\tau_n}$ is a continuous martingale.

We denote by \mathscr{M} and \mathscr{M}^{loc} the class of continuous and continuous local martingales, respectively. It is well known that \mathscr{M}^{loc} is a vector space with respect to usual operations. Several topologies can be defined on \mathscr{M}^{loc} , for example UCP, which is based on convergence in probability, or Emery topology \mathcal{A} , $\overline{7}$. Although, we can define a norm on \mathscr{M}^{loc} as

$$
||M||_{\mathcal{M}^{\text{loc}}} = \sum_{n=1}^{\infty} 2^{-n} E[1 \wedge \sup_{t \in [0,n]} |M_t|]. \tag{3}
$$

It can be seen that the topology induced by the norm in \mathcal{B} in coincides with the UCP and Emery topology (because of the continuity property). That is proved in several articles that \mathscr{M}^{loc} equipped with the norm \mathcal{B} is a complete metric space.

Let $\mathscr X$ be a Banach space. In general, a cylindrical semi-martingale on $\mathscr X$ is a continuous linear mapping $\varphi: \mathscr{X}^* \to S^0$, where S^0 denotes the space of real semi-martingales with respect to a common stochastic basis $(\Omega, \mathscr{F}, (\mathscr{F}_t)_{0 \leq t \leq 1}, P)$, endowed with the Emery topology. The general case is studied before in literature. (see for example $[10]$). As a special case, a continuous linear mapping $M : \mathcal{X}^* \to$ \mathscr{M}^{loc} is called a cylindrical continuous local martingale.

In the following, we interested to study the continuous linear mapping $M : X^* \to Y$ $\mathcal S$ where $\mathcal S$ is the collection of locally integrable semi-martingales. Our motivation comes from the collection of particular type of martingales, called as Special Semimartingales S^{SP} , coincides with S.

A processes $P = M + A$ which can be decomposed, by Doob decomposition, into a local martingale M and a predictable cádlag locally finite variation process A is known as special semimartingales. On the space of special semimartingales, we can define p-norm for $p > 0$ as follows and denote the semimartingales with finite p norm by \mathbb{H}^p :

$$
||P||_{\mathbb{H}^p} = \left(E \left[[M, M]_{\infty}^{p/2} + (\int_0^{\infty} |dA|)^p \right] \right)^{1/p}.
$$

One of the most interesting properties of special semi-martingales is compatibility of integration with the canonical decomposition in the construction of the stochastic integrals. That is, for a special semi-martingale $P = M + A$ and a predictable process ξ we have

$$
\int \xi dP = \int \xi dM + \int \xi dA
$$

3. Cylindrical Special Martingales

In this section, we define the notion of a cylindrical special martingale and integration with respect to a cylindrical special martingale.

Definition 1. Let X be a Banach space. A continuous linear mapping $P: \mathcal{X}^* \to$ $\mathcal{S}^{\rm SP}$ is called a cylindrical continuous special martingale. In this way, $Px^* = Mx^* +$ Ax^* , where Mx^* is a local martingale and Ax^* is a finite variation process, for any $x^* \in X^*$

For a cylindrical continuous special martingale P and a stopping time τ , one can define $P^{\tau}: \mathscr{X}^* \to \mathcal{S}^{\text{SP}}$ by $P^{\tau} x^*(t) = Px^*(t \wedge \tau)$. Clearly P^{τ} is also a cylindrical continuous special martingale.

We expect that our definition of a cylindrical continuous special martingale be a generalization of a cylindrical continuous local martingale. A characteristic property of a local martingale is its quadratic variation. Thanks to the finite variation part of P , which has the zero quadratic variation, we can easily define the quadratic variation $[[P]]$ of P similar to the quadratic variation of mapping to its local martingale part M.

Recall that If M is a continuous local martingale with values in a Hilbert space, then it is well known that it has a classical quadratic variation $[M]$ in the sense that there exists an a.s. unique increasing continuous process $[M]$ starting at zero such that $||M||^2 - [M]$ is a continuous local martingale again.

Definition 2. Let $P : \mathcal{X}^* \to \mathcal{S}^{\text{SP}}$ be a linear mapping. The quadratic variation $[|P|]$ of P is defined as

$$
[[P]]_t = \sup_{N \in \mathbb{N}} \sum_{n=1}^N \sup_m ([M x_m^*]_{t_i+1} - [M x_m^*]_{t_i}), \quad t \ge 0,
$$

where Mx^* is the local martingale part of Px^* and the limit is taken over all rational partitions $0 = t_0 < \cdots < t_N = t$ and $(x_m^*)_{m \geq 1}$ is a dense subset of the unit ball in X^* .

Note that existence of $(x_m^*)_{m\geq 1}$ follows from the separability of \mathscr{X}^* . For a cylindrical special semi-martingale P on a Banach space $\mathscr X$, one can think about covariance $[Px^*, Py^*]_t$ for any $x^*, y^* \in X^*$. However, by the ineffectiveness of finite variation part A of P, we have $[Px^*, Py^*]_t = [Mx^*, My^*]_t$. Therefore, by the polar

decomposition, there exists a process $Q_P : \mathbb{R}_+ \times \Omega \to \mathscr{L}(\mathscr{X}^*, \mathscr{X}^{**})$ such that for almost surly $t > 0$

$$
[Px^*, Py^*]_t = \int_0^t Q_P x^*(y^*)d[[P]]_s, \quad x^*, y^* \in X^*.
$$

The process Q_P is self-adjoint and $||Q_P(t)|| = 1$.

Let \mathscr{X}, \mathscr{Y} be two Banach spaces. For any $x^* \in \mathscr{X}^*, y \in \mathscr{Y}$, we can define the linear operator $x^* \otimes y \in \mathscr{L}(\mathscr{X}, \mathscr{Y})$ such that $x^* \otimes y : x \mapsto x^*(x)y$. Using the defined operator, the process $\phi : \mathbb{R}_+ \times \Omega \to \mathscr{L}(\mathscr{H}, \mathscr{X})$ is called elementary progressive with respect to the filtration $\mathscr{F} = (\mathscr{F}_t)_{t \in R_+}$ if it is of the form

$$
\phi(t,\omega) = \sum_{n=1}^{N} \sum_{m=1}^{M} \mathbf{1}_{(t_{n-1},t_n] \times B_{mn}}(t,\omega) \sum_{k=1}^{K} v_k \circledast x_{kmn},
$$

where $0 \le t_0 < \cdot < t_n < \infty$, for each $n = 1, \cdot, N$ the sets $B_{1n}, \ldots, B_{Mn} \in \mathscr{F}_{t_{n-1}}$ and vectors v_1, \ldots, v_K are orthogonal.

For each elementary progressive ϕ we define the stochastic integral with respect to $\mathscr{X} \in \mathscr{M}_{\text{var}}^{\text{sp}}(\mathscr{H})$ as an element of $L_0(\Omega; C_b(\mathbb{R}_+; \mathscr{X}))$ as

$$
\int_0^t \phi(s) dP(s) = \sum_{n=1}^N \sum_{m=1}^M \mathbf{1}_{B_{mn}} \sum_{k=1}^K (M(t_n \wedge t)v_k - M(t_{n-1} \wedge t)v_k + V_n(A)v_k)x_{kmn},\tag{4}
$$

where $V_n(A)$ is the total variation of process A in the n-th interval, $[t_{n-1}, t_n]$, and C_b is the set of all continuous and bounded mappings. This is usual to use the notation $\phi \cdot P$ for the process $\int_0^{\cdot} \phi(s) dP(s)$.

Clearly, the definition in $\left(\frac{\mathbf{q}}{\mathbf{q}}\right)$ is a generalization of integration with respect to a cylindrical local martingale.

Lemma 3. For all progressively measurable processes $\phi : \mathbb{R}_+ \times \Omega \to \mathscr{L}(\mathscr{H}, \mathbb{R})$ with $\phi Q_P^{1/2} \in L^2(\mathbb{R}_+,\left[\left[P\right]\right];\mathscr{L}(\mathscr{H},\mathbb{R}))$ we have

$$
\left[\int_0^\cdot \phi \, dP\right]_t = \int_0^t \phi(s) Q_P(s) \phi^*(s) \, d[[P]]_s. \tag{5}
$$

Proof. Note that our definition of quadratic variation for cylindrical special semimartingales P is reduced to its local martingale part M . Therefore, the proof is similar to the proof of $\boxed{13}$, Theorem 14.7.4.

It is important to note that for any (t, ω) in $\mathbb{R}_+ \times \Omega$, the mapping $Q_P(t, \omega)$ is a positive mapping from \mathscr{X}^* to \mathscr{X}^{**} . Therefore, there exists a Hilbert space \mathscr{H} such that $Q_P^{1/2}$ $P^{1/2}(t,\omega)$ maps \mathscr{X}^* to \mathscr{H} and $Q_P(t,\omega) = Q_P^{1/2*}$ $P^{1/2*}(t,\omega)Q_P^{1/2}$ $_{P}^{1/2}(t,\omega)$. Moreover, $\phi(t,\omega)Q_P^{1/2}$ $\frac{1}{p}(t,\omega)$ is an operator and we may think about $(\phi(t,\omega)Q_P^{1/2})$ $_P^{1/2}(t,\omega))^* =$ $Q_P^{1/2}$ $P_P^{1/2}(t,\omega)^*\phi(t,\omega)^*$. On the other hand, $\phi(t,\omega)$ is an operator from H to R and

 $\phi(t,\omega)^*$ is well defined. Breaking the Q_P appears in [\(5\)](#page-8-1) to its roots and have an inner product scheme can make a transparent illustration of the idea behind the lemma.

Theorem 1. Let \mathcal{H} be a Hilbert space and $P \in \mathcal{M}_{var}^{sp}(\mathcal{H})$. Let $\phi : \mathbb{R}_+ \times \Omega \to$ $\mathscr{L}(\mathscr{H},\mathscr{X})$ be such that ϕ^*x^* is progressively measurable for each $x^* \in \mathscr{X}^*$, and assume $\phi(\omega)Q_P(\omega)\phi^*(\omega)x^*(x^*) \in L^1_{loc}(\mathbb{R}_+,[[P]](\omega)),$ for all $x^* \in \mathcal{X}^*, \omega \in \Omega$. Set $M := \phi \cdot P$ by

$$
Mx^*(t) := \int_0^t \phi^* x^* \, dP, \quad x^* \in \mathcal{X}^*.
$$
 (6)

If $\|\phi Q_P \phi^*\|_{\infty} < \infty$ then $M \in \mathscr{M}_{\text{var}}^{\text{sp}}(\mathscr{X})$.

Proof. It is clear that for each $x^* \in \mathcal{X}^*$, mapping Mx^* is a continuous local martingale. We need just to show theta the mapping $x^* \mapsto Mx^*$ is continuous in the UPC topology. Fix $T > 0$ and set Ω_0 be a subset of Ω such that for almost every $\omega \in \Omega_0$ we have

$$
t\mapsto \langle \phi(t,\omega)Q_N(t,\omega)^*\phi(t,\omega)^*x^*,x^*\rangle\in \mathscr{L}^1(0,T).
$$

Therefore, we have a bounded operator and there exists a constant C such that

$$
\|\langle \phi(\cdot,\omega)Q_N(\cdot,\omega)^*\phi(\cdot,\omega)^*x^*,y^*\rangle\|_{L^1(0,T,[[N]](\omega))}\leq C\|x^*\|\,\|y^*\|.
$$

Moreover, we have

$$
[Mx^*]_t = \int_0^t \langle \phi(s)Q_P \phi^*(s)x^*, x^* \rangle d[[P]], \text{ for all } x^* \in \mathcal{X}^*.
$$

Note that $\|\phi(s)Q_P^{-1/2}\|_{\infty} < \infty$ by definition of ϕ and Q_P . Now let (x_n^*) be a sequence in \mathscr{X}^* and $\lim_{n\to\infty}x_n=x$. We have

$$
\begin{aligned}\n\| [Mx_n^*]_t - [Mx_n]_t \| \\
&= \left\| \int_0^t \langle \phi(s) Q_P \phi^*(s) x_n^*, x_n^* \rangle \, d[[P]] - \int_0^t \langle \phi(s) Q_P \phi^*(s) x^*, x^* \rangle \, d[[P]] \right\|_1 \\
&= \left\| \int_0^t \langle \phi(s) Q_P \phi^*(s) x_n^*, x_n^* \rangle - \langle \phi(s) Q_P \phi^*(s) x^*, x^* \rangle \, d[[P]] \right\|_1 \\
&\leq \| \phi(s) Q_P \phi^*(s) \|_{\infty} \| x_n - x \| \to 0\n\end{aligned}
$$

Corollary 1. Let M be the cylindrical continuous local martingale defined in The- $orem 1.$ $orem 1.$ Then we have

$$
[[M]]_t = \int_0^t \|\phi(s)Q_P(s)\phi^*(s)\| \,d[[P]], \quad t \ge 0.
$$

Proof. To prove the equivalence it suffices to observe that

$$
[[M]]_t = \lim \sum_{j=1}^J \sup_{x^* \in \mathcal{X}^*, ||x^*|| = 1} ([Px^*]_{t_j} - [Px^*]_{t_{j-1}})
$$

=
$$
\lim \sum_{j=1}^J \sup_{x^* \in \mathcal{X}^*, ||x^*|| = 1} \int_{t_{j-1}}^{t_j} \langle \phi(s)Q_P(s)\phi^*(s)x^*, x^* \rangle d[[P]]_s
$$

=
$$
\int_0^t ||\phi(s)Q_P(s)\phi^*(s)|| d[[P]]_s.
$$

The limit takes when the partition of $0 = t_0 < t_1 < \cdots < t_n = t$ of $[0, t]$ becomes refined, when n tends to infinity. Note that the space \mathscr{X}^* is assumed to be a separable space which helps us to justify the last equation. \Box

Corollary 2. Let M be the cylindrical continuous local martingale defined in The- $orem 1.$ $orem 1.$ Then we have

$$
\phi(s)Q_P(s)\phi^*(s) = Q_M(s)\|\phi(s)Q_P(s)\phi^*(s)\|
$$

Proof. By the Corollary $\boxed{1}$, we have

$$
[[M]]_t = \int_0^t ||\phi(s)Q_P(s)\phi^*(s)|| \, d[[P]]
$$

\n
$$
\Rightarrow \frac{d}{d[[P]]}[[M]]_t = \frac{d}{d[[P]]} \left(\int_0^t ||\phi(s)Q_P(s)\phi^*(s)|| \, d[[P]] \right)
$$

\n
$$
\Rightarrow d[[M]]_s = ||\phi(s)Q_N(s)\phi^*(s)|| \, d[[P]]_s. \tag{7}
$$

In the other way,

$$
[Mx^*, My^*]_t = \int_0^t \langle Q_P(s)\phi^*(s)x^*, \phi^*(s)y^*\rangle d[[P]]_s
$$

=
$$
\int_0^t \langle \phi(s)Q_P(s)\phi^*(s)x^*, y^*\rangle d[[P]]_s.
$$
 (8)

Replacing $\boxed{7}$ in $\boxed{8}$ implies the statement. \Box

$$
\mathbb{Z}^{\mathbb{Z}}
$$

CONCLUSION

The stochastic integration with respect to a cylindrical Semi-martingale is studied before in general case. In this research, we specified the general case to special semi-martingales and used their specific properties to refine the definition. Since the case of semi-martingales would be studied in relation with the Banach space and some convergence theorems, our refined definition would affect the convergence accuracy.

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LOCAL T_0 AND T_1 QUANTALE-VALUED PREORDERED SPACES

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Abstract. In this paper, we characterize explicitly the separation properties T_0 and T_1 at a point p in the topological category of quantale-valued preordered spaces and investigate how these characterizations are related. Moreover, we prove that local T_0 and T_1 quantale-valued preordered spaces are hereditary and productive.

1. INTRODUCTION

Classical separation axioms of topology have been extended to topological category by several authors. Baran $\boxed{2}$, in 1991, introduced these axioms in a setbased topological category in terms of initial, final structures and discreteness. He defined separation properties first locally, i.e., at a point p $[4]$, then they are expanded to point-free concepts. Using local lower separation axioms, Baran $\mathbb{Z}\setminus\{3\}$ introduced the notion of (strongly) closedness in set-based topological categories that creates closure operators in sense of Dikranjan and Giuli [\[16\]](#page-22-0) in some wellknown topological categories Conv (the category of convergence spaces and filter convergence maps) $\boxed{11}$, Lim (the category of limit spaces and filter convergence maps) $\boxed{9}$, **Prord** (the category of preordered sets and monotone maps) $\boxed{12}$ and SUConv (the category of semiuniform convergence spaces and uniformly continuous maps) [\[14\]](#page-22-4). The other significant use of these concepts to define the notions of Hausdorffness $\overline{5}$, compactness, perfectness $\overline{9}$, connectedness $\overline{10}$, regular, completely regular, normal objects $\sqrt{7}$ $\sqrt{8}$ in set-based topological categories.

A topological space defines a preorder (reflexive and transitive) relation, and a topology can be obtained from a preorder relation on a set $[17, 20]$ $[17, 20]$. This indicates

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a connection between topology and order. Domain theory that was introduced by Dana Scott in the 1960s, is a branch of order theory which studies special kinds of partially ordered sets generally named as domains. In computer science, this field is used to establish denotational semantics, particularly for functional programming languages $\left|18\right|29$. Domain theory is closely related to topology and formalizes the intuitive principles of convergence and approximation in a general way.

With the advancement of lattice theory, distinct mathematical frameworks have been studied with lattice structures including lattice-valued topology [\[15\]](#page-22-11), quantalevalued approach space $[23, 24, 28]$ $[23, 24, 28]$ $[23, 24, 28]$, quantale-valued metric space $[25]$, lattice-valued convergence space [\[22\]](#page-22-16) and lattice-valued preordered space [\[15\]](#page-22-11). This motivates us to study local T_0 and T_1 separation axioms in quantale-valued preordered spaces.

The main purpose of this paper is to give an explicit characterization for the local T_0 and T_1 separation axioms in the category of quantale-valued preordered spaces as well as to examine the relationship between them and to investigate their some invariance properties.

2. Preliminaries

Recall [\[24\]](#page-22-13) that a partially ordered set (L, \leq) is called a complete lattice if all subsets of L have both infimum (\wedge) and supremum (\vee) . For any complete lattice, the bottom element and top element is denoted by \perp and \top , respectively.

Definition 1. Let (L, \leq) be a complete lattice. We identify

- (1) $\alpha \triangleleft \beta$ (the well-below relation) if $\forall X \subseteq L$ such that $\beta \leq \bigvee X$ there exists $\delta \in X$ such that $\alpha \leq \delta$.
- (2) $\alpha \prec \beta$ (the well-above relation) if $\forall X \subseteq L$ such that $\bigwedge X \leq \alpha$ there exists $\delta \in X$ such that $\delta \leq \beta$.

Definition 2. A complete lattice (L, \leq) is called a completely distributive iff for any $\alpha \in L$, $\alpha = \bigvee {\{\beta : \beta \lhd \alpha\}}$.

Definition 3. The triple $(L, \leq, *)$ is called a quantale if the following conditions are satisfied.

- (1) (L, \leq) is a complete lattice.
- (2) $(L, *)$ is a semi group.
- (3) $(\bigvee_{i \in I} \alpha_i) * \beta = \bigvee_{i \in I} (\alpha_i * \beta) \text{ and } \beta * (\bigvee_{i \in I} \alpha_i) = \bigvee_{i \in I} (\beta * \alpha_i) \text{ for all } \alpha_i, \beta \in L$

Note that if $(L, *)$ is a commutative semi group, then the quantale $(L, \leq, *)$ is named as commutative and if for all $\alpha \in L$, $\alpha * \top = \top * \alpha = \alpha$, then it is called integral.

We denote a quantale by $L = (L, \leq, *)$ if it is integral and commutative, where (L, \leq) is completely distributive.

A quantale $L = (L, \leq, *)$ is named as a value quantale if (L, \leq) is completely distributive lattice such that $\forall \alpha, \beta \triangleleft \top$, $\alpha \vee \beta \triangleleft \top$ [\[19\]](#page-22-17).

Definition 4. [\[25,](#page-22-15) [30\]](#page-23-1) Let $X \neq \emptyset$ be a set. A map $\mathsf{R}: X \times X \to \mathsf{L} = (L, \leq, *)$ is called an L-preorder relation on X and the pair (X, R) is called an L-preordered space if it satisfies

- (1) reflexivity, i.e., for all $x \in X$, $R(x, x) = \top$,
- (2) transitivity, i.e., for all $x, y, z \in X$, $R(x, y) * R(y, z) < R(x, z)$.

Note that an L-preordered space (X, R) is named as an L-equivalence space (X, R) if for all $x, y \in X$, $\mathsf{R}(x, y) = \mathsf{R}(y, x)$ (symmetry). Also, (X, R) is called separated L-preordered space if $x = y$ whenever $R(x, y) = T$.

A map $f : (X, R_X) \to (Y, R_Y)$ is called an L-order preserving map if for all $x_1, x_2 \in X$, $\mathsf{R}_X(x_1, x_2) \leq \mathsf{R}_Y(f(x_1), f(x_2)).$

Let L-Prord denotes the category whose objects are L-preordered spaces and morphisms are L-order preserving mappings.

Example 1. (i) For $L = 2 = (\{0, 1\}, \leq, \wedge)$, 2-Prord ≅ Prord, where Prord is the category of preordered sets and order preserving maps.

- (ii) For $\mathsf{L} = ([0,\infty], >,+)$ (Lawvere's quantale), $[0,\infty]$ -**Prord** $\cong \infty q$ **Met**, where ∞q **Met** is the category of extended quasi metric spaces and nonexpansive maps.
- (iii) For $\mathsf{L} = (\triangle^+, \leq, *)$ (distance distribution functions quantale defined in $[24]$, then \triangle ⁺-Prord ≅ ProbqMet, where ProbqMet is the category of probabilistic quasi metric spaces and non-expansive maps [\[19\]](#page-22-17).

Note that in some literature, L-preordered space is often called a continuity space if L is a value quantale (see $[19]$), an L-metric space (see $[25]$) and an L-category $(see $[21]$).$ $(see $[21]$).$ $(see $[21]$).$

Recall [\[1\]](#page-21-5), let E be a category, Set be the category of sets and functions and $U : E \to Set$ be a functor. U is called topological or E is called topological category on Set if

- (i) U is amnestic and faithful (i.e., concrete),
- (ii) U consists of small fibers,
- (iii) Every U-source has a unique initial lift.

In addition, a topological functor is said to be normalized if constant objects, i.e., subterminals, have a unique structure.

Note that the forgetful functor $U : L\text{-}\text{Prod} \to \text{Set}$ is topological (see [\[21\]](#page-22-18)) and it is also normalized.

Lemma 1. [\[21\]](#page-22-18) Let (X_i, R_i) be a collection of L-preordered spaces. A source $(f_i$: $(X, \mathsf{R}) \to (X_i, \mathsf{R}_i))_{i \in I}$ is initial in **L-Prord** iff $\forall a, b \in X$,

$$
\mathsf{R}(a,b) = \bigwedge_{i \in I} \mathsf{R}_i(f_i(a), f_i(b)).
$$

Lemma 2. [\[21\]](#page-22-18) Let X be a non-empty set and (X, R) be an L-preordered space. For all $a, b \in X$,

(i) The discrete L -preorder structure on X is stated by

$$
\mathsf{R}_{dis}(a,b) = \begin{cases} \top, & a = b, \\ \bot, & a \neq b. \end{cases}
$$

(ii) The indiscrete L -preorder structure on X is stated by

$$
\mathsf{R}_{ind}(a,b) = \top.
$$

3. LOCAL T_0 and T_1 Objects

Let X be a set, $p \in X$ be a point and $X \vee_p X$ be the wedge product of X at $p \nvert 2$, i.e., two separate copies of X identified at p.

In the wedge $X\vee_p X$, a point x is represented as x_k if it lies in the k-th component for $k = 1, 2$.

Definition 5. [\[2\]](#page-21-1) Let $X \vee_p X$ be the wedge product at p and X^2 be the cartesian product of X.

(1) $A_p: X \vee_p X \to X^2$ (the principal p-axis mapping) is given by

$$
A_p(x_1) = (x, p) \text{ and } A_p(x_2) = (p, x).
$$

(2) $\mathsf{S}_p : X \vee_p X \to X^2$ (the skewed p-axis mapping) is given by

$$
S_p(x_1) = (x, x) \text{ and } S_p(x_2) = (p, x).
$$

(3) $\nabla_p: X \vee_p X \to X$ (the fold mapping at p) is given by

$$
\nabla_p(x_1) = \nabla_p(x_2) = x.
$$

Definition 6. Let (X, τ) be topological space and $p \in X$. For each point $x \neq p$, there exists an open set A such that $p \in A$, $x \notin A$ or (resp. and) there exists an open set B such that $x \in B$, $p \notin B$, then (X, τ) is said to be T_0 (resp. T_1) at $p \not |2, 6$.

Theorem 1. Let (X, τ) be topological space and $p \in X$. Then (X, τ) is T_0 (resp. T_1) at p iff the initial topology induced by { A_p (resp. S_p) : $X \vee_p X \to (X^2, \tau_*)$ and $\nabla_p: X \vee_p X \to (X, P(X))\}$ is discrete, where τ_* is the product topology on X^2 .

Proof. The proofs are given in $\boxed{6}$. \Box

Definition 7. \mathbb{R} Let $U : E \to Set$ be topological functor, $X \in Ob(E)$ with $U(X) =$ B and $p \in B$.

- (i) X is $\overline{T_0}$ at p provided that the initial lift of the U-source $\{A_p : B \vee_p B \to$ $\mathsf{U}(X^2) = B^2$ and $\nabla_p : B \vee_p B \to \mathsf{UD}(B) = B$ is discrete, where D is the discrete functor that is a left adjoint to U.
- (ii) X is T_1 at p provided that the initial lift of the U-source $\{S_p : B \vee_p B \rightarrow$ $\mathsf{U}(X^2) = B^2$ and $\nabla_p : B \vee_p B \to \mathsf{UD}(B) = B$ is discrete.

- **Remark 1.** (1) Separation axioms \overline{T}_0 at p and T_1 at p are used to identify the notions of (strong) closedness in arbitrary set-based topological categories $[2, 3]$ $[2, 3]$.
	- (2) \overline{In} **Top** (the category of topological spaces and continuous mappings), by Theorem $\boxed{1}$, \overline{T}_0 at p and T_1 at p reduce to Definition $\boxed{6}$ [\[2\]](#page-21-1).
	- (3) A topological space X is T_i , $i = 0, 1$ if and only if X is T_i , $i = 0, 1$, at p for all points p in X ($\boxed{6}$, Theorem 1.5(5)).
	- (4) Let $\mathsf{U} : \mathsf{E} \to \mathbf{Set}$ be a topological functor, X an object in E and $p \in \mathsf{U}(X)$ be a retract of X, i.e., the initial lift $h : \overline{1} \to X$ of the U-source $p : 1 \to U(X)$ is a retract, where 1 is the terminal object in Set. Then if X is $\overline{T_0}$ (resp. T_1), then X is $\overline{T_0}$ at p (resp. T_1 at p) but the converse of implication is not true, in general ($\sqrt{4}$, Theorem 2.6).
	- (5) Specially, if $\mathcal{U}: \mathcal{E} \to \mathbf{Set}$ is normalized, then $\overline{T_0}$ and T_1 imply $\overline{T_0}$ at p and T_1 at p, respectively. (\overline{A} , Corollary 2.7).

Theorem 2. An L-preordered space (X, R) is $\overline{T_0}$ at p iff $R(x, p) \wedge R(p, x) = \perp$ for all $x \in X$ distinct from p.

Proof. Assume (X, R) is $\overline{T_0}$ at p and $x \in X$ with $x \neq p$. Let R_{dis} be the discrete **L**-preorder relation on X and for $i = 1, 2, \pi_i : X^2 \to X$ be the projection maps. For $x_1, x_2 \in X \vee_p X$,

$$
R(\pi_1 A_p(x_1), \pi_1 A_p(x_2)) = R(\pi_1(x, p), \pi_1(p, x)) = R(x, p)
$$

\n
$$
R(\pi_2 A_p(x_1), \pi_2 A_p(x_2)) = R(\pi_2(x, p), \pi_2(p, x)) = R(p, x)
$$

\n
$$
R_{dis}(\nabla_p(x_1), \nabla_p(x_2)) = R_{dis}(x, x) = T
$$

Since (A, R) is $\overline{T_0}$ and $x_1 \neq x_2$, by Definition $\overline{7}$ and Lemmas $\overline{1}$, $\overline{2}$,

$$
\begin{array}{rcl} \bot & = & \bigwedge \ \{ \mathsf{R}(\pi_1\mathsf{A}_p(x_1), \pi_1\mathsf{A}_p(x_2)), \mathsf{R}(\pi_2\mathsf{A}_p(x_1), \pi_2\mathsf{A}_p(x_2)), \mathsf{R}_{dis}(\nabla_p(x_1), \nabla_p(x_2)) \} \\ & = & \bigwedge \ \{ \mathsf{R}(x, p), \mathsf{R}(p, x), \top \} \\ & = & \mathsf{R}(x, p) \land \mathsf{R}(p, x) \end{array}
$$

Hence, we have $R(x, p) \wedge R(p, x) = \bot$.

Conversely, let R' be the initial L-preorder relation on $X \vee_p X$ induced by A_p : $X \vee_p X \to \mathsf{U}(X^2, \mathsf{R}^2) = X^2$ and $\nabla_p : X \vee_p X \to \mathsf{U}(X, \mathsf{R}_{dis}) = X$, where R^2 is the product structure on X^2 induced by the projection maps π_1 and π_2 .

Assume that the condition holds, i.e., for all $x \in X$ distinct from p, $\mathsf{R}(x, p) \wedge$ $R(p, x) = \bot$. Let v and w be any points in the wedge.

(1) If $v = w$, then $\mathsf{R}'(v, w) = \top$.

(2) If
$$
v \neq w
$$
 and $\nabla_p v \neq \nabla_p w$, then $\mathsf{R}_{dis}(\nabla_p v, \nabla_p w) = \bot$. By Lemma 1.

$$
\begin{array}{lcl} \mathsf{R}^{\prime}(v,w) & = & \bigwedge \ \{ \mathsf{R}(\pi_1 \mathsf{A}_p v, \pi_1 \mathsf{A}_p w), \mathsf{R}(\pi_2 \mathsf{A}_p v, \pi_2 \mathsf{A}_p w), \mathsf{R}_{dis}(\nabla_p v, \nabla_p w) \} \\ & = & \bot \end{array}
$$

- (3) Suppose $v \neq w$ and $\nabla_p v = \nabla_p w$. It follows that $\nabla_p v = x = \nabla_p w$ for some points $x \in X$ with $x \neq p$. We must have $v = x_1$ and $w = x_2$ or $v = x_2$ and $w = x_1$ since $v \neq w$.
	- (a) If $v = x_1$ and $w = x_2$, then

$$
R(\pi_1 A_p v, \pi_1 A_p w) = R(x, p)
$$

\n
$$
R(\pi_2 A_p v, \pi_2 A_p w) = R(p, x)
$$

\n
$$
R_{dis}(\nabla_p v, \nabla_p w) = R_{dis}(x, x) = T
$$

and it follows that

$$
R'(v, w) = \bigwedge \{R(\pi_1 A_p v, \pi_1 A_p w), R(\pi_2 A_p v, \pi_2 A_p w), R_{dis}(\nabla_p v, \nabla_p w)\}
$$

= $\bigwedge \{R(x, p), R(p, x), \top\}$
= $R(x, p) \wedge R(p, x)$

By the assumption $R(x, p) \wedge R(p, x) = \bot$, we get $R'(v, w) = \bot$.

(b) Similarly, if $v = x_2$ and $w = x_1$, then $\mathsf{R}'(v, w) = \bot$.

Consequently, for all v, w in the wedge $X \vee_p X$, we obtain

$$
\mathsf{R}'(v, w) = \begin{cases} \top, & v = w \\ \bot, & v \neq w \end{cases}
$$

By Lemma $\boxed{2}$, R' is the discrete L-preorder relation on the wedge. Hence, by Definition $\overline{7}$, (X, R) is $\overline{T_0}$ at p.

Theorem 3. An L-preordered space (X, R) is T_1 at p iff $R(x, p) = \bot = R(p, x)$ for all $x \in X$ distinct from p.

Proof. Assume that (X, R) is T_1 at p and $x \in X$ with $x \neq p$. Let $v = x_1, w = x_2 \in Y$ $X \vee_p X$. Note that,

$$
R(\pi_1 S_p v, \pi_1 S_p w) = R(\pi_1(x, x), \pi_1(p, x)) = R(x, p)
$$

\n
$$
R(\pi_2 S_p v, \pi_2 S_p w) = R(\pi_2(x, x), \pi_2(p, x)) = R(x, x) = T
$$

\n
$$
R_{dis}(\nabla_p v, \nabla_p w) = R_{dis}(x, x) = T
$$

where R_{dis} is the discrete L-preorder relation on X and for each $i = 1, 2, \pi_i : X^2 \to Y$ X is the projection map. Since $v \neq w$ and (X, R) is T_1 at p, by Definition $\overline{7}$ and Lemmas $\boxed{1}$, $\boxed{2}$,

$$
\perp = \bigwedge \{ R(\pi_1 S_p v, \pi_1 S_p w), R(\pi_2 S_p v, \pi_2 S_p w), R_{dis}(\nabla_p v, \nabla_p w) \}
$$

= $\bigwedge \{ R(x, p), \top \}$
= $R(x, p)$

Similarly, if $v = x_2$, $w = x_1 \in X \vee_p X$, then by Lemma $\boxed{1}$, we have

$$
\perp = \bigwedge \{ \mathsf{R}(p,x), \top \} = \mathsf{R}(p,x)
$$

Conversely, let R' be the initial L-preorder relation on $X \vee_p X$ induced by S_p : $X \vee_p X \to U(X^2, \mathbb{R}^2) = X^2$ and $\nabla_p : X \vee_p X \to U(X, \mathbb{R}_{dis}) = X$, where \mathbb{R}^2 is the product structure on X^2 induced by the projection maps π_1 and π_2 .

Assume that for all $x \in X$ distinct from p, $R(x, p) = \bot = R(p, x)$. Let v and w be any points in the wedge.

- (1) If $v = w$, then $\mathsf{R}'(v, w) = \top$.
- (2) If $v \neq w$ and $\nabla_p v \neq \nabla_p w$, then $R_{dis}(\nabla_p v, \nabla_p w) = \perp$ since R_{dis} is the discrete structure. By Lemma $\overline{1}$,

$$
\begin{array}{lcl} \mathsf{R}^{\prime}(v,w) & = & \bigwedge \ \{ \mathsf{R}(\pi_1 \mathsf{S}_p v, \pi_1 \mathsf{S}_p w), \mathsf{R}(\pi_2 \mathsf{S}_p v, \pi_2 \mathsf{S}_p w), \mathsf{R}_{dis}(\nabla_p v, \nabla_p w) \} \\ & = & \bot \end{array}
$$

(3) Suppose $v \neq w$ and $\nabla_p v = \nabla_p w$. It follows that we must have $v = x_1$ and $w = x_2$ or $v = x_2$ and $w = x_1$.

If $v = x_1$ and $w = x_2$, then by Lemma [1,](#page-14-0)

$$
R'(v, w) = \bigwedge \{R(x, p), \top\} = R(x, p)
$$

By the assumption $R(x, p) = \bot = R(p, x)$, we get $R'(v, w) = \bot$.

Similarly, we obtain $\mathsf{R}'(v, w) = \bot$ for $v = x_2$ and $w = x_1$.

Hence, for all $v, w \in X \vee_p X$, we have

$$
\mathsf{R}'(v,w) = \begin{cases} \top, & v = w \\ \bot, & v \neq w \end{cases}
$$

By Lemma $\boxed{2}$, it follows that R' is the discrete L-preorder relation on the wedge. Consequently, by Definition $\overline{7}$, (X, R) is T_1 at p.

Example 2. Let $*$ be a binary operation identified as $\forall \alpha, \beta \in [0,1]$, $\alpha * \beta =$ $(\alpha - 1 + \beta) \vee 0$ and $\mathsf{L} = ([0, 1], \leq, *)$ be a triangular norm (Lukasiewicz t-norm) [\[26\]](#page-22-20), where the bottom and top elements are $\perp = 0$ and $\top = 1$, respectively. Let $X = \{a, b, c\}$ and an L-preorder relation $\mathsf{R}: X \times X \to \mathsf{L}$ defined by

$$
R(v, w) = \begin{cases} \top, & v = w \\ \frac{1}{2}, & (v, w) = (a, c) \\ \bot, & otherwise. \end{cases}
$$

Clearly, (X, R) is an L-preordered space. By Theorem [2,](#page-16-0) (X, R) is \overline{T}_0 at p for all $p \in X$, and by Theorem $\overline{\beta}$, (X, R) is T_1 at b but it is neither T_1 at a nor at c.

- **Remark [2](#page-16-0).** (1) By Theorems 2 and $\overline{3}$, if an L-preordered space (X, R) is T_1 at p, then it is \overline{T}_0 at p. But in general, the converse is not true (see previous Example).
	- (2) In an arbitrary set-based topological category, $\overline{T_0}$ at p and T_1 at p objects may be equivalent, for example, in $Prox$ (the category of proximity spaces and p-maps) $[27]$, CP (the category of pairs and pair preserving maps) $[3]$,

Born (the category of bornological spaces and bounded maps) **[\[3\]](#page-21-3)**, **SULim** (the category of semiuniform limit spaces and uniformly continuous maps) $[13]$, Remark 3.6.

4. Hereditary and Productive Properties

Definition 8. Let (X, R) be an L-preordered space and $A \subset X$. A subspace (A, R_A) is defined by $R_A(x, y) = R(x, y)$ for all $x, y \in A$, where R_A is the initial L-preorder structure on A induced by the inclusion map $i : A \rightarrow X$.

Theorem 4. Let (X, R) be an L-preordered space, $A \subset X$ and $p \in A$.

- (i) If (X, R) is \overline{T}_0 at p, then (A, R_A) is \overline{T}_0 at p.
- (ii) If (X, R) is T_1 at p, then (A, R_A) is T_1 at p.
- *Proof.* (i) Suppose that $p \in A$ and (X, R) is \overline{T}_0 at p. By Theorem $\overline{2}$, R $(x, p) \wedge$ $R(p, x) = \bot$ for $x \in A \subset X$ with $x \neq p$. By Definition \mathcal{B} , we have $\mathcal{R}_A(x, p) =$ $R(x, p)$ and $R_A(p, x) = R(p, x)$ for $x, p \in A \subset X$. It follows that $R_A(x, p) \wedge R_A(x, p)$ $R_A(p, x) = \perp$. Hence, by Theorem [2,](#page-16-0) the subspace (A, R_A) is also \overline{T}_0 at p.
	- (ii) Similarly, let $p \in A$ and (X, R) be T_1 at p. By Theorem $\boxed{3}$ and Definition \mathcal{R} we have $\mathcal{R}_A(x,p) = \mathcal{R}(x,p) = \bot = \mathcal{R}(p,x) = \mathcal{R}_A(p,x)$ for $x, p \in A \subset X$ with $x \neq p$. Hence, by Theorem $\overline{3}$, the subspace (A, R_A) is also T_1 at p. □

Theorem 5. Let (X_i, R_i) be an L-preordered space for each $i \in I$ and (X, R) be the product of the spaces $\{(X_i, R_i): i \in I\}$, where $X = \prod_{i \in I} X_i$ and for all $x, y \in X$, $R(x, y) = \bigwedge_{i \in I} R_i(\pi_i(x), \pi_i(y)).$ For all $i \in I$, the L-preordered space (X_i, R_i) is isomorphic to a subspace of the product space (X, R) .

Proof. Suppose that (X_i, R_i) is an L-preordered space for each $i \in I$ and (X, R) is the product space. Firstly, we choose a fixed point z_j in X_j for each $j \in I$ with j ≠ i. Let $A = \{z_1\} \times \{z_2\} \times ... \times \{z_{i-1}\} \times X_i \times \{z_{i+1}\} \times ... \subset X$. Then, (A, R_A) is a subspace of the product space (X, R) , where $\mathsf{R}_A(x, y) = \mathsf{R}(x, y)$ for all $x, y \in$ A. Clearly, *i*-th projection map $\pi_i : (A, \mathsf{R}_A) \to (X_i, \mathsf{R}_i)$ defined by for $a_i \in X_i$, $\pi_i(z_1, z_2, \ldots, z_{i-1}, a_i, z_{i+1}, \ldots) = a_i$ is bijective. For all $(z_1, z_2, \ldots, z_{i-1}, a_i, z_{i+1}, \ldots)$, $(z_1, z_2, ..., z_{i-1}, b_i, z_{i+1}, ...) \in A$, we have

$$
R_A((z_1, z_2, ..., z_{i-1}, a_i, z_{i+1}, ...,), (z_1, z_2, ..., z_{i-1}, b_i, z_{i+1}, ...))
$$

= $R((z_1, z_2, ..., z_{i-1}, a_i, z_{i+1}, ...,), (z_1, z_2, ..., z_{i-1}, b_i, z_{i+1}, ...))$
= $\bigwedge_{j \neq i} \{R_i(a_i, b_i), R_j(z_j, z_j) = \top\}$
 $\leq R_i(a_i, b_i)$
= $R_i(\pi_i(z_1, z_2, ..., z_{i-1}, a_i, z_{i+1}, ...), \pi_i(z_1, z_2, ..., z_{i-1}, b_i, z_{i+1}, ...))$

and it follows that π_i is an L-order preserving map.

On the other hand, let $f_i: (X_i, R_i) \to (A, R_A)$ be function defined by $f_i(a_i) =$ $(z_1, z_2, ..., z_{i-1}, a_i, z_{i+1}, ...)$ for $a_i \in X_i$. Then, we have

$$
(\pi_i \circ f_i)(a_i) = \pi_i(f_i(a_i))
$$

= $\pi_i(z_1, z_2, ..., z_{i-1}, a_i, z_{i+1}, ...)$
= a_i
= $1_{X_i}(a_i)$

and

$$
(f_i \circ \pi_i)(z_1, z_2, ..., z_{i-1}, a_i, z_{i+1}, ...) = f_i(\pi_i(z_1, z_2, ..., z_{i-1}, a_i, z_{i+1}, ...))
$$

= $f_i(a_i)$
= $(z_1, z_2, ..., z_{i-1}, a_i, z_{i+1}, ...)$
= $1_A(z_1, z_2, ..., z_{i-1}, a_i, z_{i+1}, ...)$

It follows that $f_i = (\pi_i)^{-1}$ since $\pi_i \circ f_i = 1_{X_i}$ and $f_i \circ \pi_i = 1_A$. For all $a_i, b_i \in X_i$, we obtain

$$
R_i(a_i, b_i) = \bigwedge_{j \neq i} \{R_i(a_i, b_i), R_j(z_j, z_j) = \top\}
$$

\n
$$
= R((z_1, z_2, ..., z_{i-1}, a_i, z_{i+1}, ...), (z_1, z_2, ..., z_{i-1}, b_i, z_{i+1}, ...))
$$

\n
$$
= R_A((z_1, z_2, ..., z_{i-1}, a_i, z_{i+1}, ...), (z_1, z_2, ..., z_{i-1}, b_i, z_{i+1}, ...))
$$

\n
$$
= R_A(f_i(a_i), f_i(b_i)) \le R_A(f_i(a_i), f_i(b_i))
$$

and it follows that f_i is an L-order preserving map.

Consequently, L-preordered space (X_i, R_i) and the subspace (A, R_A) are isomorphic. \Box

Theorem 6. Let $\{(X_i, \mathsf{R}_i): i \in I\}$ be a collection of L-preordered spaces and (X, R) be the product space, where $X = \prod_{i \in I} X_i$ and $\mathsf{R}(x, y) = \bigwedge_{i \in I} \mathsf{R}_i(\pi_i(x), \pi_i(y))$ for $x, y \in X$. Let $p = (p_i)_{i \in I}$ be a point in X.

- (i) (X, R) is T_0 at p iff (X_i, R_i) is T_0 at p_i for each $i \in I$.
- (ii) (X, R) is T_1 at p iff (X_i, R_i) is T_1 at p_i for each $i \in I$.

Proof. (i) Assume that the product space (X, R) is \overline{T}_0 at p. By Theorem $\overline{5}$ for each $i \in I$, (X_i, R_i) is isomorphic to a subspace of (X, R) and by Theorem [4,](#page-19-2) a subspace of a local \overline{T}_0 L-preordered space is \overline{T}_0 at p. Since (X, R) is \overline{T}_0 at p, it follows that (X_i, R_i) is \overline{T}_0 at p_i for each $i \in I$.

Conversely, suppose that (X_i, R_i) is T_0 at p_i for each $i \in I$. Let $x =$ $(x_i)_{i\in I}$ be a point in X with $x \neq p = (p_i)_{i\in I}$. Since $x \neq p$, there exists $i_0 \in I$ such that $x_{i_0} \neq p_{i_0}$. By the assumption L-preordered space $(X_{i_0}, \mathsf{R}_{i_0})$ is \overline{T}_0 at p and by Theorem [2,](#page-16-0) we have $R_{i_0}(x_{i_0}, p_{i_0}) \wedge R_{i_0}(p_{i_0}, x_{i_0}) = \perp$. It follows that

$$
R(x, p) = \bigwedge_{i \in I} \{R_i(x_i, p_i)\} \le R_{i_0}(x_{i_0}, p_{i_0})
$$

and

$$
R(p, x) = \bigwedge_{i \in I} \{ R_i(p_i, x_i) \} \le R_{i_0}(p_{i_0}, x_{i_0})
$$

Since $\mathsf{R}_{i_0}(x_{i_0},p_{i_0}) \wedge \mathsf{R}_{i_0}(p_{i_0},x_{i_0}) = \bot$, we get $\mathsf{R}(x,p) \wedge \mathsf{R}(p,x) = \bot$. Hence, by Theorem $\overline{2}$, the product space (X, R) is \overline{T}_0 at p.

(ii) Similarly, suppose that the product space (X, R) is T_1 at p. By the assump-tion and Theorems [4](#page-19-2) and [5,](#page-19-1) we have (X_i, R_i) is T_1 at p_i for each $i \in I$.

Conversely, assume that (X_i, R_i) is T_1 at p_i for each $i \in I$. Let $x \in$ X with $x \neq p$. Then, there exists $i_0 \in I$ such that $x_{i_0} \neq p_{i_0}$. By the assumption L-preordered space $(X_{i_0}, \mathsf{R}_{i_0})$ is T_1 at p and by Theorem $\overline{3}$, we have $R_{i_0}(x_{i_0}, p_{i_0}) = R_{i_0}(p_{i_0}, x_{i_0}) = \perp$. It follows that

$$
R(x, p) = \bigwedge \{R_1(x_1, p_1), R_2(x_2, p_2), ..., R_{i_0-1}(x_{i_0-1}, p_{i_0-1}),
$$

\n
$$
R_{i_0}(x_{i_0}, p_{i_0}) = \bot, R_{i_0+1}(x_{i_0+1}, p_{i_0+1}), ... \}
$$

\n
$$
= \bot
$$

and similarly,

$$
R(p, x) = \bigwedge_{i=1} \{R_1(p_1, x_1), ..., R_{i_0}(p_{i_0}, x_{i_0}) = \bot, ... \}
$$

= \bot

Consequently, by Theorem $\overline{3}$, we get the product space (X, R) is T_1 at p. □

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SHARP WEAK BOUNDS FOR p-ADIC HARDY OPERATORS ON p-ADIC LINEAR SPACES

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ABSTRACT. The current paper establishes the sharp weak bounds of p -adic fractional Hardy operator. Furthermore, optimal weak type estimates for padic Hardy operator on central Morrey space are also acquired.

1. INTRODUCTION

For every non-zero rational number x there is a unique $k = k(x) \in \mathbb{Z}$ such that $x = p^k s/t$, where $p \ge 2$ is a fixed prime number which is coprime to $s, t \in \mathbb{Z}$. We define a mapping $|.|_p : \mathbb{Q} \to \mathbb{R}_+$ as follows:

$$
|x|_p = \begin{cases} p^{-k} & \text{if } x \neq 0, \\ 0 & \text{if } x = 0. \end{cases}
$$
 (1)

The p-adic norm $|\cdot|_p$ undergoes many properties of the usual real norm $|\cdot|$ with an additional non-Archimedean property,

$$
|x + y|_p \le \max\{|x|_p, |y|_p\}.\tag{2}
$$

The field of p-adic numbers, denoted by \mathbb{Q}_p , is the completion of rational numbers with respect to the p-adic norm $|\cdot|_p$. A p-adic number $x \in \mathbb{Q}_p$ can be written in the formal power series as (see $[30]$):

$$
x = pk(\alpha_0 + \alpha_1 p + \alpha_2 p^2 + \dots)
$$
 (3)

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where $\alpha_i, k \in \mathbb{Z}, \alpha_0 \neq 0, \alpha_i \in \{0, 1, 2, ..., p-1\}, i = 1, 2, \cdots$. The *p*-adic norm ensures the convergence of series $\boxed{3}$ in \mathbb{Q}_p , because $|p^k \alpha_i p^i|_p \leq p^{-k-i}$.

The *n*-dimensional vector space \mathbb{Q}_p^n , $n \geq 1$, consists of tuples $\mathbf{x} = (x_1, x_2, \dots, x_n)$, where $x_j \in \mathbb{Q}_p$ and $j = 1, 2, \ldots, n$. The norm on this space is given by

$$
|\mathbf{x}|_p = \max_{1 \le j \le n} |x_j|_p.
$$

In non-Archimedean geometry, the ball and its boundary are defined, respectively, as:

$$
B_k(\mathbf{a}) = \{\mathbf{x} \in \mathbb{Q}_p^n : |\mathbf{x} - \mathbf{a}|_p \le p^k\}, S_k(\mathbf{a}) = \{\mathbf{x} \in \mathbb{Q}_p^n : |\mathbf{x} - \mathbf{a}|_p = p^k\}.
$$

For convenience we denote $B_k(\mathbf{0})$ and $S_k(\mathbf{0})$ by B_k and S_k , respectively.

The local compactness and commutativity of the group \mathbb{Q}_p^n under addition implies the existence of Haar measure $d\mathbf{x}$ on \mathbb{Q}_p^n , such that

$$
\int_{B_0} d\mathbf{x} = |B_0|_H = 1,
$$

where the notation $|B|_H$ refers to the Haar measure of a measurable subset B of \mathbb{Q}_p^n . Furthermore, it is not hard to see that $|B_k(\mathbf{a})|_H = p^{nk}, |S_k(\mathbf{a})|_H = p^{nk}(1-p^{-n}),$ for any $\mathbf{a} \in \mathbb{Q}_p^n$.

Let $w(\mathbf{x})$ be a nonnegative locally integrable function on \mathbb{Q}_p^n and $w(E)$ the weighted measure of measurable subset $E \subset \mathbb{Q}_p^n$, that is $w(E) = \int_E w(x) dx$ respectively. The space of all complex-valued functions f with norm conditions:

$$
||f||_{L^r(w;\mathbb{Q}_p^n)} = \bigg(\int_{\mathbb{Q}_p^n} |f(\mathbf{x})|^r w(\mathbf{x}) d\mathbf{x}\bigg)^{1/r} < \infty,
$$

is denoted by $L^r(w, \mathbb{Q}_p^n)$, $(0 < r < \infty)$, and is known as weighted Lebesgue space. Note that $L^r(1, \mathbb{Q}_p^n) = L^r(\mathbb{Q}_p^n)$.

In $[22]$, authors have defined the weighted p-adic weak Lebesgue space $L^{r,\infty}(w;\mathbb{Q}_p^n)$ by

$$
||f||_{L^{r,\infty}(w,\mathbb{Q}_p^n)} = \sup_{\mu>0} \mu w \bigg(\{\mathbf{x} \in \mathbb{Q}_p^n : |f(\mathbf{x})| > \mu \} \bigg)^{1/r} < \infty.
$$

When $w = 1$, we get the weak Lebesgue space $L^{r,\infty}(\mathbb{Q}_p^n)$ defined in $\boxed{32}$. Next, we give the relevant p-adic function spaces.

Definition 1. [\[34\]](#page-34-3) Suppose $1 < r < \infty$ and $\mu \in \mathbb{R}$. The p-adic space $\dot{B}^{r,\mu}(\mathbb{Q}_p^n)$ is the set of all measurable functions $f: \mathbb{Q}_p^n \to \mathbb{R}$ which satisfy

$$
||f||_{\dot{B}^{r,\mu}(\mathbb{Q}_p^n)} = \sup_{\gamma \in \mathbb{Z}} \left(\frac{1}{|B_{\gamma}|_H^{1+\mu r}} \int_{B_{\gamma}} |f(\mathbf{x})|^r d\mathbf{x} \right)^{1/r} < \infty.
$$

When $\mu = -1/r$, then

 $\dot{B}^{r,\mu}(\mathbb{Q}_p^n) = L^r(\mathbb{Q}_p^n)$. It is easy to see that $\dot{B}^{r,\mu}(\mathbb{Q}_p^n)$ is reduced to $\{0\}$ whenever $\mu < -1/r$.

Definition 2. [\[35\]](#page-34-4) Suppose $\mu \in \mathbb{R}$ and $1 < r < \infty$. The p-adic space $W\dot{B}^{r,\mu}(\mathbb{Q}_p^n)$ is defined as

$$
W\dot{B}^{r,\mu}(\mathbb{Q}_p^n) = \{f : ||f||_{W\dot{B}^{r,\mu}(\mathbb{Q}_p^n)} < \infty\},\
$$

where

$$
||f||_{W\dot{B}^{r,\mu}(\mathbb{Q}_p^n)} = \sup_{\gamma \in \mathbb{Z}} |B_{\gamma}|_{H}^{-\mu - 1/r} ||f||_{WL^r(B_{\gamma})},
$$

and $||f||_{WL^{r}(B_{\gamma})}$ is the local p-adic L^r-norm of $f(x)$ restricted to the ball B_{γ} , that is

$$
||f||_{WL^{r}(B_{\gamma})} = \sup_{\mu>0} |\{\mathbf{x} \in B_{\gamma} : |f(\mathbf{x})| > \mu\}|^{1/r}.
$$

Evidently, if $\mu = -1/r$, then $W\dot{B}^{r,\mu}(\mathbb{Q}_p^n) = L^{r,\infty}(\mathbb{Q}_p^n)$. Also, $\dot{B}^{r,\mu}(\mathbb{Q}_p^n) \subseteq W\dot{B}^{r,\mu}(\mathbb{Q}_p^n)$ for $-1/r < \mu < 0$ and $1 \leq r < \infty$.

In the last several decades, a growing interest to p -adic models have been seen because p-adic analysis is a natural base for development of various models of ultrametric diffusion energy landscape [\[4\]](#page-33-0). It also attracts great deal of interest towards quantum mechanics $\boxed{30}$, theoretical biology $\boxed{11}$, quantum gravity $\boxed{1, 7}$, string theory $\boxed{31}$, spin glass theory $\boxed{3}$, $\boxed{26}$. In $\boxed{4}$, it was shown that the *p*-adic analysis can be efficiently applied both to relaxation in complex speed systems and processes combined with the relaxation of a complex environment. Besides, the applications of p-adic analysis can be found in harmonic analysis and pseudodifferential equations, see for example $\sqrt{\frac{5}{9}}$, $\sqrt{21}$, $\sqrt{28}$, $\sqrt{29}$.

The one-dimensional Hardy operator

$$
Hf(x) = \frac{1}{x} \int_0^x f(t)dt, \quad x > 0,
$$
\n(4)

has been introduced by Hardy in [\[18\]](#page-33-8) for measurable functions $f: \mathbb{R}^+ \to \mathbb{R}^+$. This operator satisfies the inequality

$$
||Hf||_{L^r(\mathbb{R}^+)} \le \frac{r}{r-1} ||f||_{L^r(\mathbb{R}^+)}, \quad 1 < r < \infty,\tag{5}
$$

where the constant $r/(r-1)$ is sharp.

In $[12]$, Faris has proposed an extension of the Hardy operator H on higher dimensional Euclidean space \mathbb{R}^n by

$$
Hf(\mathbf{x}) = \frac{1}{|\mathbf{x}|^n} \int_{|\mathbf{t}| \le |\mathbf{x}|} f(\mathbf{t}) d\mathbf{t}.
$$
 (6)

where $|\mathbf{x}| = (\sum_{i=1}^n x_i^2)^{1/2}$ for $\mathbf{x} = (x_1, \dots, x_n)$. In addition, Christ and Grafakos $\boxed{8}$ have obtained the exact value of the norm of $[6]$. For more details related to Hardy type operators and, in particular, to boundedness of these operators, we refer to publications $\left| \frac{6}{13}, \frac{19}{23}, \frac{24}{24}, \frac{27}{36}, \frac{39}{39} \right|$.

On the other hand, the fractional Hardy operator is obtained by merely writing $|\cdot|^{n-\alpha}$ $(0 \leq \alpha < n)$ instead of $|\cdot|^n$ with in $[6]$. The weak type estimates for the fractional Hardy type operators has also spotlighted many researchers in the past, see for example $\sqrt{2, 13, 15, 16, 20, 37, 38}$ $\sqrt{2, 13, 15, 16, 20, 37, 38}$ $\sqrt{2, 13, 15, 16, 20, 37, 38}$ $\sqrt{2, 13, 15, 16, 20, 37, 38}$ $\sqrt{2, 13, 15, 16, 20, 37, 38}$ $\sqrt{2, 13, 15, 16, 20, 37, 38}$ $\sqrt{2, 13, 15, 16, 20, 37, 38}$.

In what follows, the higher dimensional fractional Hardy operator in the p-adic field

$$
H_{\alpha}^p f(\mathbf{x}) = \frac{1}{|\mathbf{x}|_p^{n-\alpha}} \int_{|\mathbf{t}|_p \leq |\mathbf{x}|_p} f(\mathbf{t}) d\mathbf{t}, \qquad \mathbf{x} \in \mathbb{Q}_p^n \setminus \{\mathbf{0}\}.
$$

has been defined and studied for $0 \leq \alpha < n$ and $f \in L_{loc}(\mathbb{Q}_p^n)$ in [\[33\]](#page-34-16). When $\alpha = 0$, the operator H^p_α transfers to the *p*-adic Hardy operator (see [\[14\]](#page-33-17)). Fu et al. in [14] have acquired the optimal bounds of p-adic Hardy operator on Lebesgue spaces. For more details, we refer the publications $\frac{17}{22}$ $\frac{25}{34}$ and the references therein.

The purpose of the current paper is to study the sharp weak bounds for fractional Hardy operator in the p -adic field on p -adic Lebesgue space. Moreover, we also discuss the optimal weak type estimates for Hardy operator in the p-adic field on central Morrey spaces.

2. Sharp weak bounds for p-adic fractional Hardy Operator on Lebesgue spaces

Our main result for this section is as follows.

Theorem 1. Suppose $0 < \alpha < n$ and $n + \gamma > 0$. If $f \in L^1(\mathbb{Q}_p^n)$, then

$$
\|H_{\alpha}^p f\|_{L^{(n+\gamma)/(n-\alpha),\infty}(|\mathbf{x}|_p^{\gamma};\mathbb{Q}_p^n)} \leq C \|f\|_{L^1(\mathbb{Q}_p^n)},
$$

where the constant

$$
C = \left(\frac{1 - p^{-n}}{1 - p^{-(n+\gamma)}}\right)^{(n-\alpha)/(n+\gamma)}
$$

is optimal.

Proof. We have

$$
|H_{\alpha}^{p} f(\mathbf{x})| = \left| \frac{1}{|\mathbf{x}|_{p}^{n-\alpha}} \int_{|\mathbf{t}|_{p} \leq |\mathbf{x}|_{p}} f(\mathbf{t}) d\mathbf{t} \right|
$$

$$
\leq |\mathbf{x}|_{p}^{-(n-\alpha)} \|f\|_{L^{1}(\mathbb{Q}_{p}^{n})}.
$$
 (7)

Let $C_1 = ||f||_{L^1(\mathbb{Q}_p^n)}$, then

$$
\{\mathbf x\in\mathbb Q_p^n:|H_\alpha^pf(\mathbf x)|>\mu\}\subset\{\mathbf x\in\mathbb Q_p^n:|\mathbf x|_p<(C_1/\mu)^{1/(n-\alpha)}\}.
$$

Thus,

$$
||H_{\alpha}^{p}f||_{L^{(n+\gamma)/(n-\alpha),\infty}(|x|_{p}^{\gamma};\mathbb{Q}_{p}^{n})
$$

\n
$$
\leq \sup_{\mu>0} \mu \left(\int_{\mathbb{Q}_{p}^{n}} \chi_{\{\mathbf{x}\in\mathbb{Q}_{p}^{n}:|H_{\alpha}^{p}f(\mathbf{x})|>\mu\}}(\mathbf{x})|\mathbf{x}|_{p}^{\gamma}d\mathbf{x} \right)^{(n-\alpha)/(n+\gamma)}
$$

\n
$$
\leq \sup_{\mu>0} \mu \left(\int_{\mathbb{Q}_{p}^{n}} \chi_{\{\mathbf{x}\in\mathbb{Q}_{p}^{n}:|\mathbf{x}|_{p}<\left(C_{1}/\mu\right)^{1/(n-\alpha)}\}}(\mathbf{x})|\mathbf{x}|_{p}^{\gamma}d\mathbf{x} \right)^{(n-\alpha)/(n+\gamma)}
$$

\n
$$
= \sup_{\mu>0} \mu \left(\sum_{\mu>0}^{\log_{p}\left(C_{1}/\mu\right)^{1/(n-\alpha)}} \mathbf{x}|_{p}^{\gamma}d\mathbf{x} \right)^{(n-\alpha)/(n+\gamma)}
$$

\n
$$
= \sup_{\mu>0} \mu \left(\sum_{j=-\infty}^{\log_{p}\left(C_{1}/\mu\right)^{1/(n-\alpha)}} \int_{S_{j}} |\mathbf{x}|_{p}^{\gamma}d\mathbf{x} \right)^{(n-\alpha)/(n+\gamma)}
$$

\n
$$
= (1-p^{-n})^{(n-\alpha)/(n+\gamma)} \sup_{\mu>0} \mu \left(\sum_{j=-\infty}^{\log_{p}\left(C_{1}/\mu\right)^{1/(n-\alpha)}} p^{j(n+\gamma)}d\mathbf{x} \right)^{(n-\alpha)/(n+\gamma)}
$$

\n
$$
= \left(\frac{1-p^{-n}}{1-p^{-(n+\gamma)}} \right)^{(n-\alpha)/(n+\gamma)} \sup_{\mu>0} \mu \left(\frac{C_{1}}{\mu} \right)
$$

\n
$$
\leq \left(\frac{1-p^{-n}}{1-p^{-(n+\gamma)}} \right)^{(n-\alpha)/(n+\gamma)} ||f||_{L^{1}(|\mathbf{x}|_{p}^{\beta})} .
$$

\n(8)

To show that the constant

$$
\left(\frac{1-p^{-n}}{1-p^{-(n+\gamma)}}\right)^{(n-\alpha)/(n+\gamma)},
$$

appeared in [\(8\)](#page-28-0) is optimal, we proceed as, consider

$$
f_0(\mathbf{x}) = \chi_{\{\mathbf{x} \in \mathbb{Q}_p^n : |\mathbf{x}|_p \le 1\}}(\mathbf{x}),
$$

then

$$
||f_0||_{L^1(\mathbb{Q}_p^n)}=1.
$$

Also,

$$
H_{\alpha}^{p} f_{0}(\mathbf{x}) = \frac{1}{|\mathbf{x}|_{p}^{n-\alpha}} \int_{|\mathbf{t}|_{p} \leq |\mathbf{x}|_{p}} f_{0}(\mathbf{t}) d\mathbf{t}
$$

\n
$$
= \frac{1}{|\mathbf{x}|_{p}^{n-\alpha}} \int_{|\mathbf{t}|_{p} \leq |\mathbf{x}|_{p}} \chi_{\{\mathbf{x} \in \mathbb{Q}_{p}^{n} : |\mathbf{t}|_{p} \leq 1\}}(\mathbf{t}) d\mathbf{t}
$$

\n
$$
= \frac{1}{|\mathbf{x}|_{p}^{n-\alpha}} \begin{cases} \int_{|\mathbf{t}|_{p} \leq |\mathbf{x}|_{p}} d\mathbf{t}, & |\mathbf{x}|_{p} \leq 1; \\ \int_{|\mathbf{t}|_{p} \leq 1} d\mathbf{t}, & |\mathbf{x}|_{p} > 1. \end{cases}
$$

Since $|B_{\log_p} \times |E|_p |_H = |\mathbf{x}|_p^n |B_0|_H$, therefore,

$$
H_{\alpha}^{p} f_0(\mathbf{x}) = \begin{cases} |\mathbf{x}|_p^{\alpha}, & |\mathbf{x}|_p \leq 1; \\ |\mathbf{x}|_p^{\alpha-n}, & |\mathbf{x}|_p > 1. \end{cases}
$$

Now,

$$
\{\mathbf x \in \mathbb Q_p^n : |H_\alpha^p f_0(\mathbf x)| > \mu\} = \{|\mathbf x|_p \le 1 : |\mathbf x|_p^\alpha > \mu\} \cup \{|\mathbf x|_p > 1 : |\mathbf x|_p^{\alpha - n} > \mu\}.
$$

Since $0 < \alpha < n$, therefore, when $\mu > 1$, then

Since $0 < \alpha < n$, therefore, when $\mu \geq 1$, then

$$
\{{\bf x}\in\mathbb{Q}_p^n:|H_\alpha^pf_0({\bf x})|>\mu\}=\emptyset,
$$

and when $0 < \mu < 1,$ then

$$
\{\mathbf x\in\mathbb Q_p^n:|H_\alpha^pf_0(\mathbf x)|>\mu\}=\{\mathbf x\in\mathbb Q_p^n:(\mu)^{1/\alpha}<|\mathbf x|_p<(1/\mu)^{1/n-\alpha}\}.
$$

Ultimately we are down to:

$$
||H_{\alpha}^{p}f_{0}||_{L^{(n+\gamma)/(n-\alpha)}),\infty}(|\mathbf{x}|_{p}^{\gamma};\mathbb{Q}_{p}^{n})
$$
\n
$$
= \sup_{0<\mu<1} \mu \left(\int_{\mathbb{Q}_{p}^{n}} \chi_{\{\mathbf{x}\in\mathbb{Q}_{p}^{n}:(\mu)^{1/\alpha}<|\mathbf{x}|_{p}<(1/\mu)^{1/(n-\alpha)}\}} (\mathbf{x})|\mathbf{x}|_{p}^{\gamma}d\mathbf{x} \right)^{(n-\alpha)/(n+\gamma)}
$$
\n
$$
= \sup_{0<\mu<1} \mu \left(\int_{(\mu)^{1/\alpha}<|\mathbf{x}|_{p}<(1/\mu)^{1/(n-\alpha)}} |\mathbf{x}|_{p}^{\gamma}d\mathbf{x} \right)^{(n-\alpha)/(n+\gamma)}
$$
\n
$$
= (1-p^{-n})^{(n-\alpha)/(n+\gamma)} \sup_{0<\mu<1} \mu \left(\sum_{j=\log_{p}\mu^{1/(n+\gamma)}}^{(\log_{p}\mu^{1/(n-\alpha)})} p^{j(n+\gamma)} \right)^{(n-\alpha)/(n+\gamma)}
$$
\n
$$
= (1-p^{-n})^{(n-\alpha)/(n+\gamma)} \sup_{0<\mu<1} \mu \left(\frac{p^{(\log_{p}\mu^{1/\alpha}+1)(n+\gamma)} - p^{(\log_{p}\mu^{1/(n-\alpha)}+1)(n+\gamma)} - p^{(n-\alpha)/(n+\gamma)} - (1-p^{(n+\gamma)})}{1-p^{(n+\gamma)}} \right)^{(n-\alpha)/(n+\gamma)}
$$
\n
$$
= (1-p^{-n})^{(n-\alpha)/(n+\gamma)} \sup_{0<\mu<1} \mu \left(\frac{\mu^{(n+\gamma)/\alpha} - \mu^{(n+\gamma)/(n-\alpha)}}{p^{-(n+\gamma)} - 1} \right)^{(n-\alpha)/(n+\gamma)}
$$
\n
$$
= (1-p^{-n})^{(n-\alpha)/(n+\gamma)} \sup_{0<\mu<1} \left(\frac{1-\mu^{(n+\gamma)/\alpha}\mu^{(n+\gamma)/(n-\alpha)}}{1-p^{-(n+\gamma)}} \right)^{(n-\alpha)/(n+\gamma)}
$$
\n
$$
= \left(\frac{1-p^{-n}}{1-p^{-(n+\gamma)}} \right)^{(n-\alpha)/(n+\gamma)} \sup_{0<\mu<1} \left(1-\mu^{(n+\gamma)/\alpha}\mu^{(n+\gamma)/(n-\alpha)} \right)^{(n-\alpha)/(n+\gamma)}
$$

We thus conclude from (8) and (9) that

$$
||H_{\alpha}^{p}||_{L^{1}(\mathbb{Q}_{p}^{n})\rightarrow L^{(n+\gamma)/(n-\alpha),\infty}(|\mathbf{x}|_{p}^{\gamma};\mathbb{Q}_{p}^{n})} = \left(\frac{1-p^{-n}}{1-p^{-(n+\gamma)}}\right)^{1/q}.
$$

3. Optimal Weak Type Estimates for p-adic Hardy Operator on Weak Central Morrey Spaces

In the current section we investigate the boundedness of p -adic Hardy operator on p-adic weak central Morrey spaces. It is shown the constant obtained in this case is also optimal.

Theorem 2. Suppose $-1/r \leq \mu < 0, 1 \leq r < \infty$ and if $f \in \dot{B}^{r,\mu}(\mathbb{Q}_p^n)$, then

$$
\|H^pf\|_{W\dot{B}^{r,\mu}(\mathbb{Q}_p^n)}\leq \|f\|_{\dot{B}^{r,\mu}(\mathbb{Q}_p^n)},
$$

and the constant 1 is optimal.

Proof. Applying Hölder's inequality, we obtain

$$
\begin{aligned} |H^p f(\mathbf{x})| \leq & \frac{1}{|\mathbf{x}|_p^n} \bigg(\int_{B(0,|\mathbf{x}|_p)} |f(\mathbf{t})|^r d\mathbf{t} \bigg)^{1/r} \bigg(\int_{B(0,|\mathbf{x}|_p)} d\mathbf{t} \bigg)^{1/r'} \\ = & |\mathbf{x}|_p^{n\mu} \|f\|_{\dot{B}^{r,\mu}(\mathbb{Q}_p^n)} . \end{aligned}
$$

Let $C_2 = ||f||_{\dot{B}^{r,\mu}(\mathbb{Q}_p^n)}$. Since $\mu < 0$, we have

$$
||H^p f||_{W\dot{B}^{r,\mu}(\mathbb{Q}_p^n)} \leq \sup_{\gamma \in \mathbb{Z}} \sup_{y>0} y|B_{\gamma}|_{H}^{-\mu-1/r} |\{\mathbf{x} \in B_{\gamma} : C_2 |\mathbf{x}|_p^{n\mu} > y\}|^{1/r}
$$

=
$$
\sup_{\gamma \in \mathbb{Z}} \sup_{y>0} y|B_{\gamma}|_{H}^{-\mu-1/r} |\{|\mathbf{x}|_p \leq p^{\gamma} : |\mathbf{x}|_p < (y/C_2)^{1/n\mu}\}|^{1/r}.
$$

If $\gamma \leq \log_p(y/C_2)^{1/n\mu}$, then for $\mu < 0$, we obtain

$$
\sup_{y>0} \sup_{\gamma \leq \log_p(y/C_2)^{1/n\mu}} y|B_{\gamma}|_H^{-\mu-1/r} |\{|\mathbf{x}|_p \leq p^{\gamma} : |\mathbf{x}|_p < (y/C_2)^{1/n\mu}\}|^{1/r}
$$

$$
\leq \sup_{y>0} \sup_{\gamma \leq \log_p(y/C_2)^{1/n\mu}} t p^{-\gamma n\mu}
$$

= C_2

$$
\leq ||f||_{\dot{B}^{r,\mu}(\mathbb{Q}_p^n)}.
$$

If $\gamma > \log_p(y/C_2)^{1/n\mu}$, then for $\mu + 1/r > 0$, we get

$$
\sup_{y>0} \sup_{\gamma > \log_p(y/C_2)^{1/n\mu}} y|B_{\gamma}|_H^{-\mu-1/r} |\{|\mathbf{x}|_p \le p^{\gamma} : |\mathbf{x}|_p < (y/C_2)^{1/n\mu}\}|^{1/r}
$$

$$
\le \sup_{y>0} \sup_{\gamma > \log_p(y/C_2)^{1/n\mu}} yp^{-\gamma n(\mu+1/r)} (y/C_2)^{1/r\mu}
$$

$$
= C_2
$$

$$
\le ||f||_{\dot{B}^{r,\mu}(\mathbb{Q}_p^n)}.
$$

Therefore,

$$
||H^p f||_{W\dot{B}^{r,\mu}(\mathbb{Q}_p^n)} \le ||f||_{\dot{B}^{r,\mu}(\mathbb{Q}_p^n)}.
$$
\n(10)

Conversely, to prove that the constant 1 is optimal, consider

$$
f_0(\mathbf{x}) = \chi_{\{|\mathbf{x}|_p \le 1\}}(\mathbf{x}),
$$

then,

$$
||f_0||_{\dot{B}^{q,\mu}(\mathbb{Q}_p^n)} = \sup_{\gamma \in \mathbb{Z}} \left(\frac{1}{|B_{\gamma}|_H^{1+\mu r}} \int_{B_{\gamma}} \chi_{\{|\mathbf{x}|_p \le 1\}}(\mathbf{x}) d\mathbf{x} \right)^{1/r}.
$$

If $\gamma < 0$, then

$$
\sup_{\substack{\gamma \in \mathbb{Z} \\ \gamma < 0}} \left(\frac{1}{|B_{\gamma}|_H^{1+\mu r}} \int_{B_{\gamma}} d\mathbf{x} \right)^{1/r} = \sup_{\substack{\gamma \in \mathbb{Z} \\ \gamma < 0}} p^{-n \gamma \mu} = 1,
$$

since $\mu < 0$. If $\gamma \ge 0$, then using the condition that $\mu + 1/r > 0$, we have

$$
\sup_{\substack{\gamma \in \mathbb{Z} \\ \gamma \ge 0}} \left(\frac{1}{|B_{\gamma}|_H^{1+\mu r}} \int_{B_0} d\mathbf{x} \right)^{1/r} = \sup_{\substack{\gamma \in \mathbb{Z} \\ \gamma \ge 0}} p^{-n\gamma(\mu+1/r)} = 1.
$$

Therefore,

$$
||f_0||_{\dot{B}^{r,\mu}(\mathbb{Q}_p^n)}=1.
$$

Moreover,

$$
Hp f_0(\mathbf{x}) = \begin{cases} 1, & |\mathbf{x}|_p \le 1; \\ |\mathbf{x}|_p^{-n}, & |\mathbf{x}|_p > 1, \end{cases}
$$

which implies that $|H^pf_0(\mathbf{x})| \leq 1$. Next, in order to construct weak central Morrey norm we divide our analysis into following two cases: Case 1. When $\gamma \leq 0$, then

$$
||H^p f_0||_{WL^r(B_\gamma)} = \sup_{0 < y \le 1} y | \{ \mathbf{x} \in B_\gamma : 1 > y \} |^{1/r} = p^{n\gamma/r},
$$

and

$$
||H^p f_0||_{W\dot{B}^{r,\mu}(\mathbb{Q}_p^n)} = \sup_{\gamma \leq 0} |B_{\gamma}|_H^{-\mu-1/r} ||f||_{WL^r(B_{\gamma})} = \sup_{\gamma \leq 0} p^{-n\gamma\mu} = 1 = ||f_0||_{\dot{B}^{r,\mu}(\mathbb{Q}_p^n)}.
$$

Case 2. When $\gamma > 0$, we have

$$
||H^p f_0||_{WL^r(B_\gamma)} = \sup_{0 < y \le 1} y | \{ \mathbf{x} \in B_0 : 1 > y \} \cup \{ 1 < | \mathbf{x} |_p < p^\gamma : | \mathbf{x} |_p^{-n} > y \} |^{1/r}.
$$

For further analysis, this case is further divided into the following subcases: Case 2(a). If $1 < \gamma < \log_p y^{-1/n}$, then

$$
||H^{p} f_0||_{WL^{r}(B_{\gamma})} = \sup_{0 < y \leq 1} y\{1 + p^{n\gamma} - 1\}^{1/r} = \sup_{0 < t \leq 1} tp^{n\gamma/r}.
$$

Case 2(b). If $1 < \log_p y^{-1/n} < \gamma$, then:

$$
||H^p f_0||_{WL^r(B_\gamma)} = \sup_{0 < y \le 1} y(1 + y^{-1} - 1)^{1/r} = \sup_{0 < y \le 1} y^{1 - 1/r}.
$$

Now, for $1 \le r < \infty$ and $-1/r \le \mu < 0$, from case 2(a) and 2(b), we obtain

$$
||H^{p} f_{0}||_{W\dot{B}^{r,\mu}(\mathbb{Q}_{p}^{n})}
$$
\n
$$
= \max \left\{ \sup_{0 \le y \le 1} \sup_{1 \le \gamma \le \log_{p}(1/y)^{-1/n}} y p^{-n\gamma\mu}, \sup_{0 \le y \le 1} \sup_{1 \le \log_{p}(1/y)^{-1/n} \le \gamma} y^{1-1/r} p^{-n\gamma(\mu+1/r)} \right\}
$$
\n
$$
= \max \left\{ \sup_{0 \le y \le 1} t^{1+\mu}, \sup_{0 \le y \le 1} y^{1+\mu} \right\}
$$
\n
$$
= 1 = ||f_{0}||_{\dot{B}^{r,\mu}(\mathbb{Q}_{p}^{n})}.
$$
\n
$$
\lim_{n \to \infty} \sqrt{10^n} \text{ and } \sqrt{11} \text{ are series at } (11)
$$
\n(11)

Finally, using (10) and (11) , we arrive at:

$$
||H||_{\dot{B}^{r,\mu}(\mathbb{Q}_p^n)\to W\dot{B}^{r,\mu}(\mathbb{Q}_p^n)}=1.
$$

□

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DOMINATOR SEMI STRONG COLOR PARTITION IN GRAPHS

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ABSTRACT. Let $G = (V, E)$ be a simple graph. A subset S is said to be Semi-Strong if for every vertex v in V, $|N(v) \cap S| \leq 1$, or no two vertices of S have the same neighbour in V , that is, no two vertices of S are joined by a path of length two in V . The minimum cardinality of a semi-strong partition of G is called the semi-strong chromatic number of G and is denoted by $\chi_s G$. A proper colour partition is called a dominator colour partition if every vertex dominates some colour class, that is , every vertex is adjacent with every element of some colour class. In this paper, instead of proper colour partition, semi-strong colour partition is considered and every vertex is adjacent to some class of the semi-strong colour partition.Several interesting results are obtained.

1. INTRODUCTION

Let $G = (V, E)$ be a finite, undirected graph. We follow standard definitions of graph theory $\sqrt{2}$, $\sqrt{8}$. A proper vertex coloring of a graph is defined as coloring the vertices of a graph such that no two adjacent vertices are colored using same color. A subset S of a graph $G = (V, E)$ is said to be a dominting set if every vertex not in S is adjacent to at least one vertex of $V-S$. The domination number $\gamma(G)$ is the number of vertices in a smallest dominating set for G $[9, 10]$ $[9, 10]$. S.M. Hedetniemi $\boxed{11}$, $\boxed{12}$ introduced and discussed the concept of dominator coloring and dominator partition of graphs. S.Arumugam et.al. discussed further in dominator coloring in graphs [\[1\]](#page-47-7). The combination of the two most important fields in graph

Keywords. Dominator coloring, semi strong color partition, semi-strong coloring.

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theory namely, Coloring and domination have a lot of research results. A dominator coloring of a graph G is a proper coloring, such that every vertex of G dominates at least one color class (possibly its own class). Gera et. al. $\overline{6}$ defined dominator colouring in a graph G as a proper colour partition in which every vertex dominates some color class. The dominator chromatic number of G, denoted by $\chi_d(G)$, is the minimum number of colors among all dominator colorings of G. Gera researched further in [\[7\]](#page-47-1) on dominator coloring and safe clique partitions. Kazemi proposed the concept of total dominator coloring in graphs and studied its properties [\[15\]](#page-47-2). A proper coloring, such that each vertex of the graph is adjacent to every vertex of some (other) color class. For more results on the total dominator coloring, refer to [\[14,](#page-47-3)[16\]](#page-48-0). M. Chellali and F. Maffray discussed Dominator colorings in some classes of graphs $\left[4\right]$. In 2015, Merouane et al. $\left[17\right]$ proposed the dominated coloring which is defined as a proper coloring such that every color class is dominated by at least one vertex. The dominated chromatic number of G, denoted by $\chi_{dom}(G)$, is the minimum number of colors among all dominated colorings of G . For comprehensive results of coloring and domination in graphs, color class domination in graphs introudced and studied in detail. refer to $\sqrt{5}$, $\sqrt{20}$, $\sqrt{21}$. As a generalization of strong set introduced by Claude Berge $\boxed{3}$, E.Sampathkumar defined semi-strong sets $\boxed{18}$ in a graph. In a simple graph G , a subset S of the vertex set $V(G)$ of G is called a semi-strong set of G if $|N[v] \cap S| \leq 1$ for v in $V(G)$. E.Sampathkumar also introduced Chromatic partition of a graph [\[19\]](#page-48-5) and studied its properties. Also, G. Jothilakshmi et al studied (k,r) - Semi Strong Chromatic Number of a Graph [\[13\]](#page-47-7). Instead of proper color partition, semi-strong partition $[18]$ of $V(G)$ is considered and domination property that every vertex dominates semi-strong color class is added. The minimum cardinality of such a partition is found for some classes of graphsand bounds are obtained. Interesting results in this new concepts are derived.

Definition 1. A subset S of $V(G)$ is called a maximal semi-strong set of G if S is semi-strong and no proper super set of S is semi-strong. The maximum cardinality of a maximal semi-strong set of G is called semi-strong number of G and is denoted by $ss(G)$.

Definition 2. A dominator coloring of a graph G is a proper coloring in which each vertex of the graph dominates every vertex of some color class.

Definition 3. A semi-strong coloring of G is called a **dominator semi-strong** color partition of G if every vertex of G dominates an element of the partition. The minimum cardinality of such a partition is called the dominator semi-strong color partition number of G and is denoted by $\chi_s^d(G)$.

Since the trivial partition is a semi-strong coloring of G , the existence of dominator semi-strong color partition is guaranteed in any graph.

2. $\chi_s^d(G)$ for Some Well-Known Graphs

Observation 1. (i) $\chi_s^d(K_n) = \chi_d(K_n) = n$. (ia) $\chi_s^d(K_n - e) = n$ (since $K_n - e$ has a full degree vertex). (ii) $\chi_s^d(K_{1,n}) = n + 1$, $\chi_s(K_{1,n}) + \gamma(K_{1,n}) = n + 1$. (iii) $\chi_d(K_{m,n}) = 2 < \chi_s^d(K_{m,n})$ if $m \leq n$ and $n \geq 3$.

Remark 1. Let $\Pi = \{V_1, V_2, \ldots, V_k\}$ be a dominator semi-strong color partition of G. A vertex $u \in V$ can dominate V_i if and only if $|V_i| = 1$.

Theorem 1. For any Path P_n , $\chi_s^d(P_n) = \left\lceil \frac{n}{2} \right\rceil + 1$, $n \ge 2$.

Proof. Let P_n be a path on n vertices.

Case 1: $n = 4k, k \ge 1$

Let $\Pi = \{V_1, V_2, \ldots, V_{2k}, V_{2k+1}\}\$ where $V_1 = \{v_2, v_3, v_6, v_7, \ldots, v_{4k-2}, v_{4k-1}\}\$, $V_2 =$ ${v_1}, V_3 = {v_5}, \ldots, V_k = {v_{4k-3}}, V_{k+1} = {v_4}, V_{k+2} = {v_8}, \ldots, V_{2k+1} = {v_{4k}}.$ Then Π is a dominator semi-strong color partition of P_n . Therefore $\chi_s^d(P_n) \leq$ $2k+1 = \lceil \frac{n}{2} \rceil + 1.$

Let Π_1 be a χ_s^d -partition of P_n . The maximum cardinality of an element of Π_1 is at most $2k$. There are at least $2k$ singletons to dominate $4k$ elements, since no single element can dominate two elements of a set which are at a distance 2. Therefore $|\Pi| \ge 2k + 1$. Therefore $\chi_s^d(P_{4k}) = 2k + 1 = \lceil \frac{n}{2} \rceil + 1$.

Case 2: Let $n = 4k + 1, k \ge 1$

Let $\Pi = \{V_1, V_2, \ldots, V_{2k}, V_{2k+1}, V_{2k+2}\}\$ where $V_1 = \{v_2, v_3, v_6, v_7, \ldots, v_{4k-2}, v_{4k-1}\}\$ $V_2 = \{v_1\}, V_3 = \{v_5\}, \ldots, V_k = \{v_{4k-3}\}, V_{k+1} = \{v_{4k+1}\}, V_{k+2} = \{v_4\},$ $V_{k+3} = \{v_8\}, \ldots, V_{2k+2} = \{v_{4k}\}.$ Then Π is a dominator semi-strong color partition of P_n . Therefore $\chi_s^d(P_n) \leq 2k + 2 = \lceil \frac{n}{2} \rceil + 1$.

Arguing as in case 1, $\chi_s^d(P_{4k+1}) \geq 2k+2 = \lceil \frac{n}{2} \rceil + 1$. Therefore $\chi_s^d(P_n) = \lceil \frac{n}{2} \rceil + 1$, where $n = 4k + 1$.

Case 3: Let $n = 4k + 2, k \ge 0$

Let $\Pi = \{V_1, V_2, \ldots, V_{2k}, V_{2k+1}, V_{2k+2}\}\$ where

 $V_1 = \{v_2, v_3, v_6, v_7, \ldots, v_{4k-2}, v_{4k-1}, v_{4k+2}\}, V_2 = \{v_1\}, V_3 = \{v_5\}, \ldots, V_k =$ ${v_{4k-3}}$, $V_{k+1} = {v_{4k+1}}$, $V_{k+2} = {v_4}$, $V_{k+3} = {v_8}$, ..., $V_{2k+2} = {v_{4k}}$. Then Π is a dominator semi-strong color partition of P_n . Therefore $\chi_s^d(P_n) \leq 2k+2 = \left[\frac{n}{2}\right]+1$.

Arguing as in case 1, $\chi_s^d(P_{4k+2}) \geq 2k+2 = \lceil \frac{n}{2} \rceil + 1$. Therefore $\chi_s^d(P_n) = \lceil \frac{n}{2} \rceil + 1$, where $n = 4k + 2$.

Case 4: Let $n = 4k + 3$, $k > 0$

Let $\Pi = \{V_1, V_2, \ldots, V_{2k}, V_{2k+1}, V_{2k+2}, V_{2k+3}\}\$ where

 $V_1 = \{v_2, v_3, v_6, v_7, \ldots, v_{4k-2}, v_{4k-1}, v_{4k+2}\}, V_2 = \{v_1\}, V_3 = \{v_5\}, \ldots, V_k =$ ${v_{4k-3}}, V_{k+1} = {v_{4k-3}}, V_{k+2} = {v_{4k+1}}, V_{k+3} = {v_4}, V_{k+4} = {v_8}, \ldots, V_{2k+2} =$ $\{v_{4k}\}, V_{2k+3} = \{v_{4k+3}\}.$ Then Π is a dominator semi-strong color partition of P_n . Therefore $\chi_s^d(P_n) \leq 2k + 3 = \lceil \frac{n}{2} \rceil + 1$.

Arguing as in case 1, $\chi_s^d(P_{4k+3}) \geq 2k+3 = \lceil \frac{n}{2} \rceil + 1$. Therefore $\chi_s^d(P_n) = \lceil \frac{n}{2} \rceil + 1$, where $n = 4k + 3$.

Theorem 2. $\chi_s^d(C_n) = \left\lceil \frac{n}{2} \right\rceil + 1, n \geq 3.$

Proof. Let C_n be a cycle on n vertices.

Case 1: $n = 4k, k > 1$

Let $V(C_n) = \{v_1, v_2, \ldots, v_{4k}\}.$ Let $\Pi = \{V_1, V_2, \ldots, V_{2k}, V_{2k+1}\}\$ where $V_1 = \{v_2, v_3, v_6, v_7, \ldots, v_{4k-2}, v_{4k-1}\}, V_2 = \{v_1\}, V_3 = \{v_5\}, \ldots, V_{k+1} = \{v_{4k-3}\}, V_{k+2} =$ ${v_4}, V_{k+3} = {v_8}, \ldots, V_{2k+1} = {v_{4k}}.$ Then Π is a dominator semi-strong color partition of C_n . Therefore $\chi_s^d(C_n) \leq |\Pi| = 2k + 1 = \frac{4k}{2} + 1 = \lceil \frac{n}{2} \rceil + 1$.

There are at least $2k$ singletons and no single element can dominate a two element set which are at a distance 2. Therefore $\chi_s^d(C_{4k}) \geq \lceil \frac{n}{2} \rceil + 1$. Therefore $\chi_s^d(C_{4k}) = \left\lceil \frac{n}{2} \right\rceil + 1$.

Case 2: Let $n = 4k + 1, k \ge 1$

Let $\Pi = \{V_1, V_2, \ldots, V_{2k}, V_{2k+1}, V_{2k+2}\}$ where $V_1 = \{v_2, v_3, v_6, v_7, \ldots, v_{4k-2}, v_{4k-1}\},$ $V_2 = \{v_1\}, V_3 = \{v_5\}, \ldots, V_{k+1} = \{v_{4k-3}\}, V_{k+2} = \{v_{4k+1}\}, V_{k+3} = \{v_4\},$ $\ldots, V_{2k+2} = \{v_{4k}\}.$ Then Π is a dominator semi-strong color partition of P_n . Therefore $\chi_s^d(C_{4k+1}) \leq |\Pi| = 2k+2 = \left\lceil \frac{4k+1}{2} \right\rceil + 1 = \left\lceil \frac{n}{2} \right\rceil + 1.$

There are at least $2k$ singletons and no single element can dominate a two element set which are at a distance 2. Therefore $\chi_s^d(C_{4k+1}) \geq \lceil \frac{n}{2} \rceil + 1$ and hence $\chi_s^d(C_{4k+1}) =$ $\lceil \frac{n}{2} \rceil + 1.$

Case 3: Let $n = 4k + 2, k \ge 1$

Let $\Pi = \{V_1, V_2, \ldots, V_{2k}, V_{2k+1}, V_{2k+2}\}\$ where $V_1 = \{v_2, v_3, v_6, v_7, \ldots, v_{4k-2}, v_{4k-1}\}\$ $V_2 = \{v_{4k+1}, v_{4k+2}\}, V_3 = \{v_1\}, V_4 = \{v_5\}, \ldots, V_{k+2} = \{v_{4k-3}\}, V_{k+3} = \{v_4\}, V_{k+4} =$ $\{v_8\}, \ldots, V_{2k+2} = \{v_{4k}\}.$ Then Π is a dominator semi-strong color partition of C_n . Therefore $\chi_s^d(C_{4k+2}) \leq |\Pi| = 2k + 2 + 1 = \lceil \frac{n}{2} \rceil + 1.$

There are at least $2k$ singletons and no single element can dominate a two element set which are at a distance 2. Any doubleton must consist of consecutive vertices. Therefore $\chi_s^d(C_{4k+2}) \geq \lceil \frac{n}{2} \rceil + 1$. Therefore $\chi_s^d(C_{4k+2}) = \lceil \frac{n}{2} \rceil + 1$.

Case 4: Let $n = 4k + 3$, $k > 0$

Let $\Pi = \{V_1, V_2, \ldots, V_{2k}, V_{2k+1}, V_{2k+2}, V_{2k+3}\}$ where

 $V_1 = \{v_2, v_3, v_6, v_7, \ldots, v_{4k-1}, v_{4k+2}\}, V_2 = \{v_1\}, V_3 = \{v_5\}, \ldots, V_{k+1} = \{v_{4k-3}\},$ $V_{k+2} = \{v_{4k+1}\}\,$, $V_{k+3} = \{v_4\}, \ldots, V_{2k+2} = \{v_{4k}\}\,$, $V_{2k+3} = \{v_{4k+3}\}\$. Then Π is a dominator semi-strong color partition of C_n . Therefore $\chi_s^d(C_{4k+3}) \leq |\Pi| = 2k+3$ $\lceil \frac{4k+3}{2} \rceil + 1 = \lceil \frac{n}{2} \rceil + 1.$

There are at least $2k + 1$ singletons and no single element can dominate a two element set which are at a distance 2. Any doubleton set must consist of consecutive vertices. Therefore $\chi_s^d(C_{4k+3}) \geq \lceil \frac{n}{2} \rceil + 1$. Therefore $\chi_s^d(C_{4k+3}) = \lceil \frac{n}{2} \rceil + 1$.

Theorem 3. For Complete bi-partite graph $K_{m,n}$, $\chi_s^d(K_{m,n}) = max\{m, n\} + 1$.

Proof. Let V_1, V_2 be the partite sets of $K_{m,n}$.

Case 1: Let $m < n$.

Let $V_1 = \{u_1, u_2, \ldots, u_m\}$ and $V_2 = \{v_1, v_2, \ldots, v_n\}.$

Let $\Pi = \{\{u_1, v_1\}, \ldots, \{u_{m-1}, v_{m-1}\}, \{u_m\}, \{v_m\}, \ldots, \{v_n\}\}\.$ Then each of $v_1, v_2, \ldots,$ v_n dominates $\{u_m\}$, and each of $u_1, u_2, \ldots, u_{m-1}$ dominates $\{v_n\}$. Therefore Π is a dominator semi-strong color partition of $K_{m,n}$.

Therefore $\chi_s^d(K_{m,n}) \leq |\Pi| = m + n - (m - 1) = n + 1$.

No two elements of V_1 can belong to an element of Π . Also no two elements of V_2 can belong to an element of Π. Any element of V_1 dominates all elements of V_2 . So is the case with V_2 . Therefore Π must consist of at least one singleton from V_1 and one singletons from V_2 . Therefore $\chi_s^d(K_{m,n}) \geq m - 1 + 2 + (n - m) = n + 1$. Therefore $\chi_s^d(K_{m,n}) = n + 1 = max\{m, n\} + 1$. **Case 2:** Let $m = n$

Let $\Pi = \{\{u_1, v_1\}, \ldots, \{u_{m-1}, v_{m-1}\}, \{u_m\}, \ldots, \{v_n\}\}\.$ Proceeding as in case 1, $\chi_s^d(K_{m,n}) = m + 1 = max\{m, n\} + 1.$

Corollary 1. $\chi_s^d(K_{1,n}) = n + 1$.

Theorem 4. $\chi_s^d(K_m(a_1, a_2, \ldots, a_m)) = m + max\{a_1, a_2, \ldots, a_m\}.$

Proof. Let $a_1 \le a_2 \le ... \le a_m$. Let $V(K_m(a_1, a_2, ..., a_m)) = \{u_1, u_2, ..., u_m\}$ $v_{1,1}, v_{1,2}, \ldots, v_{1,a_1}, \ldots, v_{m,1}, \ldots, v_{m,a_m}$. Let $\Pi = \{\{u_1\}, \ldots, \{u_m\}, \{v_{1,1}, v_{2,1}, \ldots, v_{m,a_m}\}\$. $v_{m,1}\},\ldots,\{v_{1,a_1},v_{2,a_1},\ldots,v_{m,a_1}\},\{v_{2,a_2},v_{3,a_2},\ldots,v_{m,a_2}\},\ldots,\{v_{m,a_m}\}\}.$ Then $|\Pi| = m + a_m = m + max\{a_1, a_2, \ldots, a_m\}.$

Therefore $\chi_s^d(K_m(a_1, a_2, \ldots, a_m)) \leq m + max\{a_1, a_2, \ldots, a_m\}$. Any χ_s^d -partition must contain u_1, u_2, \ldots, u_m as singletons for dominating the pendent vertices. Further no two pendent vertices at any u_i , $1 \leq i \leq m$ can belong to an element of the partition. Therefore $\chi_s^d(K_m(a_1, a_2, \ldots, a_m)) \geq m + max\{a_1, a_2, \ldots, a_m\}$. Therefore $\chi_s^d(K_m(a_1, a_2, \ldots, a_m)) = m + max\{a_1, a_2, \ldots, a_m\}.$

Let G be the graph shown in Figure 1

FIGURE 1. $G = K_4(1, 2, 3, 3)$ with $\chi_s^d(G) = 7$

Let $\Pi = \{\{v_1\}, \{v_2\}, \{v_3\}, \{v_4\}, \{v_5, v_6, v_8, v_{11}\}, \{v_7, v_9, v_{12}\}, \{v_{10}, v_{13}\}\}\.$ Then II is a χ_s^d -partition of G. Therefore $\chi_s^d(G) = |\Pi| = 4 + 3 = 7$.

Theorem 5. $\chi_s^d(K_{a_1, a_2,..., a_m}) = a_1 + a_2 + ... + a_m$ if $m \geq 3$.

Proof. Let $m \geq 3$. Then any vertex of $K_{a_1, a_2, ..., a_m}$ is a common vertex of two vertices. Hence no two vertices can be included in an element of a χ_s^d -partition. Hence $\chi_s^d(K_{a_1, a_2, ..., a_m}) = a_1 + a_2 + ... + a_m$ if $m \ge 3$.

Theorem 6. $\chi_s^d(P) = 7$ where P is the Petersen graph.

Proof. Consider the graph in Figure 2. Let $V(P) = \{v_1, v_2, \ldots, v_{10}\}.$ Let $\Pi = \{\{v_1, v_2\}, \{v_3, v_4\}, \{v_5\}, \{v_6, v_9\}, \{v_7\}, \{v_8\}, \{v_{10}\}\}\.$ Then Π is a dominator semi-strong color partition of P. Therefore $\chi_s^d(P) \leq 7$.

Figure 2. Petersen Graph

In any χ_s^d -partition of P, no three-element set can appear. Since for any three element set of P, there exists a vertex which is adjacent to two of the element of that set. Any three 2 element sets must have three singletons for domination. Hence the remaining one element must appear as a singleton. Therefore $\chi_s^d(P) \geq 7$. Therefore $\chi_s^d(P) = 7$.

Remark 2. (i) $1 \leq \chi_s^d(G) \leq n$. (*ii*) $\chi_s^d(G) = 1$ *if and only if* $G = K_1$.

Observation 2. Let G be a graph with full degree vertex. Then $\chi_s^d(G) = |V(G)|$.

Proof. Let Π be a χ_s^d -partition of G. Let $V_1 \in \Pi$. If $|V_1| \geq 2$, then any two points of V_1 are adjacent with full degree vertex, a contradiction. Therefore $|V_1| = 1$. Therefore χ_s^d $(G) = |V(G)|.$

Corollary 2. $\chi_s^d(W_n) = n$.

Corollary 3. $\chi_s^d(F_n) = n$.

3. Main Results

Theorem 7. $max\{\chi_s(G), \gamma(G)\} \leq \chi_s^d(G) \leq \chi_s(G) + \gamma(G)$.

Proof. Since any χ_s^d -partition of G is a χ_s -partition of G, $\chi_s(G) \leq \chi_s^d(G)$. Let $\Pi =$ $\{V_1, V_2, \ldots, V_k\}$ where $k = \chi_s^d(G)$ be a χ_s^d -partition of G. Let $x_i \in V_i$, $1 \leq i \leq k$. Let $S = \{x_1, x_2, \ldots, x_k\}$. Let $v \in V - S$. Then v dominates some color class, say V_i . Therefore v is adjacent with x_i . Therefore $\{x_1, x_2, \ldots, x_k\}$ dominates G. That is, S is a dominating set of G. That is, $\gamma(G) \leq |S| = k = \chi_s^d(G)$. Therefore $max\{\chi_s(G), \gamma(G)\} \leq \chi_s^d(G).$

Let $\Pi = \{V_1, V_2, \ldots, V_k\}$ be a χ_s -coloring of G. Assign colors $\chi_s(G) + 1, \ldots, \chi_s(G)$ $+ \gamma(G)$ to the vertices of a minimum dominating set of G, leaving the rest of the vertices colored as before. Then the resulting partition is a dominator semi-strong color partition of G. Therefore, $\chi_s^d(G) \leq |\Pi| + \gamma(G) = \chi_s(G) + \gamma(G)$.

Remark 3. The set S need not be a minimum dominating set. For example, when $G = P_6$, $\chi_s^d(G) = 4$. But $\gamma(P_6) = 2$.

Theorem 8. Let a, b be positive integers with $a \leq b$. Then there exists a graph G such that $\chi_d(G) = a$ and $\chi_s^d(G) = b$.

Proof. When $a = b$, $\chi_d(K_a) = \chi_s^d(K_a) = a$. Let $a < b$. Let $G = K_{a_1, a_2, ..., a_k}$ where $k = a$. Then $\chi_d(G) = a$. Choose a_1, a_2, \ldots, a_k such that $a_1 + a_2 + \ldots + a_k =$ b. Then χ_s^d $(G) = b.$

Theorem 9. $\chi_s^d(G) = 2$ if and only if $G = K_2$.

Proof. If $G = K_2$, then $\chi_s^d(G) = 2$. Suppose $\chi_s^d(G) = 2$. Let $\Pi = \{V_1, V_2\}$ be a χ_s^d -partition of G. Suppose $|V_1| \geq 2$. Then any vertex of V_2 dominates V_1 unless $|V_2| = 1$. If $|V_2| > 1$, then it is a contradiction. Therefore $|V_2| = 1$. Similarly, $|V_1| = 1$. Therefore $G = K_2$.

Corollary 4. Suppose T is a tree of order $n \ge 2$. Then $\chi(T) = 2$. $\chi_s^d(T) = \chi(T)$ if and only if $\chi_s^d(T) = 2$. That is if and only if $G = K_2$.

Theorem 10. Let G be a connected unicyclic graph. Then $\chi_s^d(G) = \chi(G)$ if and only if $G = C_3$.

Proof. If G is a cycle, then $\chi_s^d(G) = \chi(G)$ if and only if $G = C_3$. Suppose G contains C_{2n} . Then $\chi(G) = 2$, but $\chi_s^d(G) \geq 3$, a contradiction. Therefore G contains an odd cycle C_{2n+1} . Then $\chi(G) = 3$. If there exists a path attached with a vertex of C_{2n+1} , then $\chi_s^d(G) \geq 4$, a contradiction. Therefore G is a cycle. Since $\chi_s^d(G) = \chi(G), G = C_3.$

Theorem 11. Let G be a connected graph. Then $\chi_s^d(G) = n$ if and only if either G has a full degree vertex or $N(G) = K_n$.

Proof. Let $\chi_s^d(G) = n$. Let $V(G) = \{u_1, u_2, \ldots, u_n\}$. Then $\Pi = \{\{u_1\}, \{u_2\}, \ldots, \{u_n\}\}\$ is a χ_s^d -partition of G. Let $diam(G) = k \geq 3$. Let u and v be the end vertices of a diametrical path. Let $u = u_1, u_2, \ldots, u_{k+1} = v$. Then u and v have no common adjacent vertex. Therefore $\Pi_1 = \{\{u, v\}, \ldots, \{u_n\}\}\.$ Then u dominates $\{u_2\}$ and v dominates $\{u_k\}$. Also $\{u, v\}$ is dominated by a single vertex. Therefore Π_1 is a dominator semi-strong color partition of *G*. Therefore $χ_s^d(G) ≤ n - 1$, a contradiction. Therefore $diam(G) \leq 2$.

Suppose u_1 and u_2 are adjacent and u_1u_2 is not the edge of a triangle. Then $\{u_1, u_2\}$ can be taken as an element of a dominator semi-strong color partition of G with all other vertices as singletons. If u_1 is adjacent with some $u_i, i \geq 3$ and u_2 is adjacent with some $u_j, j \neq \{1, 2\}$, then $\chi_s^d(G) \leq n-1$, a contradiction. Therefore if $|V(G)| \geq 4$ and $diam(G) \leq 2$ and u_1u_2 is an edge such that u_1 and u_2 have separate adjacent vertices, then u_1u_2 is the edge of a triangle. In such case, $N(G) = K_n$. Suppose u_1 is adjacent with some vertex u_3 and u_2 is not adjacent with any vertex of G other than u_1 . Suppose u_3 is adjacent with some vertex u_4 . If u_1 is not adjacent with u_4 , then $\Pi_2 = \{\{u_1, u_3\}, \{u_2\}, \{u_4\}, \ldots, \{u_n\}\}\$ is a dominator semi-strong color partition of G , a contradiction. If u_3 is adjacent with u_1 , then u_4 is also adjacent with u_1 . Therefore G is a connected graph with a full degree vertex.

Suppose G has no full degree vertex. Then the case that only one of u_1, u_2 which are adjacent, has some other adjacent vertex does not hold. Therefore both u_1 and u_2 have different adjacent vertices. Therefore u_1u_2 is the edge of a triangle. Therefore $diam(G) \leq 2$ and when u_1u_2 is an edge, then u_1u_2 is the edge of a triangle. Therefore $N(G) = K_n$. The converse is obvious. \Box

Remark 4. Let G be the graph given in Figure 3.

Then $G = N(G)$, $N(G)$ is not complete and G has no full degree vertex. Therefore $\chi_s^d(G) = 4$ and $\chi_s(G) = 3$.

Remark 5. Let G be the graph shown in Figure 4.

Then $N(G) = K_5 - \{e\}$. G has a full degree vertex and hence $\chi_s^d(G) = 5$ eventhough $N(G)$ is not complete. Hence $\chi_s(G) = 4$ and $\chi_s^d(G) = 5$.

Remark 6. Let G be a complete multipartite graph $K_{a_1, a_2,...,a_n}$, $n \geq 3$. Then G has no full degree vertex. $\chi_s^d(G) = n$ and hence $N(G) = K_n$.

Observation 3. Let G be a cycle C_n with pendent vertex attached with exactly one vertex of C_n . Then $\chi_s^d(G) = \begin{cases} \chi_s^d(C_n) + 1 & \text{if } n \not\equiv 1 \pmod{4} \\ \chi_s^d(C_n) & \text{otherwise} \end{cases}$ $\chi_s^d(C_n)$ otherwise

Proof. Let $V(C_n) = \{u_1, u_2, \ldots, u_n\}$. Let u_{n+1} be a pendent vertex attached with u_1 . **Case 1:** Let $n = 4k$.

FIGURE 3. $G = N(G) = C_5$

FIGURE 4. G and $N(G)$

Let $\Pi = \{\{u_{4k+1}, u_3, u_4, u_7, u_8, \ldots, u_{4k-5}, u_{4k-4}, u_{4k-1}\}, \{u_1\}, \{u_2\}, \{u_5\}, \{u_6\}, \}$ $\ldots, \{u_{4k-3}\}, \{u_{4k-2}\}, \{u_{4k}\}\}.$ Then Π is a dominator semi-strong color partition of G. Therefore $\chi_s^d(G) \leq 1 + 2k + 1 = 2k + 2 = \lceil \frac{n}{2} \rceil + 2 = \chi_s^d(C_n) + 1$.

There are at least 2k singletons and no single element can dominate a 2 element set whose elements are at distance 2. Also for the pendent vertex either it appears as a singleton or its support appears as a singleton. Therefore $\chi_s^d(G) \geq \lceil \frac{n}{2} \rceil + 2$. Therefore $\chi_s^d(G) = \left\lceil \frac{n}{2} \right\rceil + 2 = \chi_s^d(C_n) + 1.$

Case 2: Let $n = 4k + 1$.

Let $\Pi = \{\{u_{4k+2}, u_3, u_4, u_7, u_8, \ldots, u_{4k-1}, u_{4k}\}, \{u_1\}, \{u_2\}, \{u_5\}, \{u_6\}, \ldots, \{u_{4k+1}\}\}.$ Then Π is a dominator semi-strong color partition of G. Therefore $\chi_s^d(G) \leq 1 + k + \frac{1}{2}$ $1 + k = 2k + 2 = \frac{n}{2} + 1 = \chi_s^d(C_n).$

If $\chi_s^d(G) < \left\lceil \frac{n}{2} \right\rceil + 1$, then removing the pendent vertex we get that $\chi_s^d(C_n)$ $\lceil \frac{n}{2} \rceil + 1$, a contradiction. Therefore $\chi_s^d(G) = \lceil \frac{n}{2} \rceil + 1 = \chi_s^d(C_n)$.

Case 3: Let $n = 4k + 2$.

Let $\Pi = \{\{u_{4k+3}, u_3, u_4, u_7, u_8, \ldots, u_{4k-5}, u_{4k-4}, u_{4k-1}, u_{4k}\}, \{u_1\}, \{u_2\}, \{u_5\}, \{u_6\}, \}$ $..., \{u_{4k-3}\}, \{u_{4k-2}\}, \{u_{4k+1}\}, \{u_{4k+2}\}\}.$ Then Π is a dominator semi-strong color partition of G. Therefore $\chi_s^d(G) \leq \left\lceil \frac{n}{2} \right\rceil + 2 = \chi_s^d(C_n) + 1$.

Arguing as in case 1, we get that
$$
\chi_s^d(G) \geq \lceil \frac{n}{2} \rceil + 2
$$
.
Therefore $\chi_s^d(G) = \lceil \frac{n}{2} \rceil + 2 = \chi_s^d(C_n) + 1$.

Case 4: Let $n = 4k + 3$.

Let $\Pi = \{\{u_{4k+4}, u_3, u_4, u_7, u_8, \ldots, u_{4k-1}, u_{4k}\}, \{u_1\}, \{u_2\}, \{u_5\}, \{u_6\}, \ldots, \{u_{4k+1}\},\$ ${u_{4k+2}, \{u_{4k+3}\}\}.$ Then Π is a dominator semi-strong color partition of G. Therefore $\chi_s^d(G) \leq |\Pi| = 1 + k + 1 + k + 2 = 2k + 4 = \lceil \frac{n}{2} \rceil + 2 = \chi_s^d(C_n) + 1.$

Arguing as in case 1, we get that $\chi_s^d(G) \geq \lceil \frac{n}{2} \rceil + 2$. Therefore $\chi_s^d(G) = \left\lceil \frac{n}{2} \right\rceil + 2 = \chi_s^d(C_n) + 1.$

Proposition 1. If $diam(G) \leq 2$, then $\chi_s^d(G) \geq \lceil \frac{n}{2} \rceil$, where $|V(G)| = n$.

Proof. Let G be a connected graph and $diam(G) \leq 2$. If $diam(G) = 1$, then $G = K_n$ and $\chi_s^d(G) = n \geq \lceil \frac{n}{2} \rceil$. Suppose $diam(G) = 2$. Then $\chi_s^d(G) \geq \chi_s(G) \geq$ $\lceil \frac{n}{2} \rceil$ [?]. □

Remark 7. The converse of the above proposition need not be true. For: $\chi_s^d(C_n) = \lceil \frac{n}{2} \rceil + 1 > \lceil \frac{n}{2} \rceil$ for all $n \geq 3$. When $n \geq 6$, $diam(C_n) \geq 3$.

Definition 4. $C_m(a_1, a_2, \ldots, a_m)$ is the graph obtained from the cycle C_m by attaching $a_i \geq 1$) pendent vertices at the vertex u_i of C_m , $1 \leq i \leq m$.

Proposition 2. $\chi_s^d(C_m(a_1, a_2, \ldots, a_m)) = m + max\{a_1, a_2, \ldots, a_m\}.$

Proof. The proof follows on the same line as the proof of the Theorem \overline{A} . □

Theorem 12. Let G be a connected graph. Then $\chi_s^d(G) = n-1$, where $|V(G)| = n$ if and only if $n \geq 4$. When $n = 4$, $G = P_4$ or C_4 . When $n = 5$, G is one of the ten graphs P_5 , C_5 , $D_{1,2}$ or G_i , $(1 \leq i \leq 7)$ given in Figure 5. When $n \geq 6$, there exist two vertices say u_1 , u_2 such that u_1 and u_2 may be either adjacent or independent and there exist u_i , $(3 \leq i \leq n)$ adjacent with u_1 and not with u_2 , there exist u_j , $(j \neq i)$, $(3 \leq k \leq n)$ such that u_r and u_s are adjacent with u_k and u_1 may

FIGURE 5. A set of graphs $G_1, G_2, G_3, G_4, G_5, G_5, G_7$ with $n = 5$ and $\chi_s^d(G) = n - 1$

be adjacent with any u_k , $(k \neq j)$, u_2 may be adjacent with any u_k , $(k \neq i)$ but u_1 and u_2 are not together adjacent with any u_k .

Proof. Let G be a connected graph. Let $\chi_s^d(G) = n - 1$. Let $\Pi = \{\{u_1, u_2\}, \{u_3\},\$ ${u_4}, \ldots, {u_n}$ be a χ_s^d -partition of G.

Case 1: u_1 and u_2 are adjacent.

Let $u_i, 3 \leq i \leq n$, be such that u_i is not adjacent with both u_1 and u_2 . That is, either u_i is adjacent with u_1 and not with u_2 or u_i is adjacent with u_2 and not with u_1 or u_i is not adjacent with both u_1 and u_2 . Since Π is a χ_s^d -partition, there exist some $u_i, 3 \le i \le n$ adjacent with u_1 and some $u_j, j \ne i, 3 \le j \le n$, adjacent with u_2 . Then u_i , u_2 have a common vertex u_1 and u_j , u_1 have a common vertex u_2 . Any two of the vertices u_3, \ldots, u_n have a common vertex that is, $d(u_r, u_s) \leq 2$. Let $n \geq 6$. Suppose u_r and u_s are adjacent, $r \neq s$, $r, s \notin \{1,2\}$, $3 \leq r, s \leq n$. Then there exist $u_k, 3 \leq k \leq n, k \neq \{r, s\}$ such that u_i, u_j, u_k form a triangle. If u_r and u_s are independent, then there exist u_k , $3 \leq k \leq n$, $k \neq \{i, j\}$ such that u_r, u_s, u_k form a path of length 2. If $n = 5$, then only one vertex is left other than u_1, u_2, u_i, u_j , and the graph is either P_5 or $D_{1,2}$ or C_5 , a contradiction. Subcase 1: $n = 3$

Then $G = P_3$ or K_3 . Then $\chi_s^d(G) = 3$, a contradiction. Therefore $n \geq 4$. Subcase 2: $n = 4$

Then $G = P_4$, C_4 , K_4 , $K_{1,3}$, $K_4 - \{e\}$. When $G = K_4$, $K_{1,3}$, $K_4 - \{e\}$, G has a full degree vertex. Therefore $\chi_s^d(G) = 4$, a contradiction. Hence $G = P_4$ or C_4 . Subcase 3: $n = 5$

Then $G = P_5, C_5, K_5, K_{1,4}, K_5 - \{e\}, K_5 - \{e_1, e_2\}$ or one of the following graphs shown in Figure 6:

Therefore $\chi_s^d(G) = 4$ if $G = P_5, C_5, D_{1,2}$ or one of the following graphs shown in Figure 7:

FIGURE 6. A set of graphs $G_1, G_2, G_3, G_4, G_5, G_5, G_7$ with $n = 5$

Case 2: u_i and u_j are independent.

Let $u_i, 3 \leq i \leq n$, be not adjacent with both u_1 and u_2 . That is, either u_i is adjacent with u_1 and not with u_2 or u_i is adjacent with u_2 and not with u_1 or u_i is not adjacent with both u_1 and u_2 . Then Π is a χ_s^d -partition, there exist some u_i , $3 \leq i \leq n$ adjacent with u_1 and some u_j , $j \neq i$, $3 \leq j \leq n$, adjacent with u_2 . Then u_i , u_2 have a common vertex u_1 and u_j , u_1 have a common vertex u_2 . Any two of the vertices u_3, \ldots, u_n have a common vertex that is, $d(u_r, u_s) \leq 2$. Let $n \geq 6$. Suppose u_r and u_s are adjacent, $r \neq s$, $r, s \notin \{1, 2\}$, $3 \leq r, s \leq n$. Then there exist $u_k, 3 \leq k \leq n, k \neq \{r, s\}$ such that u_r, u_s, u_k form a triangle. If u_r and u_s are independent, then there exist u_k , $3 \leq k \leq n$, $k \neq \{r, s\}$ such that u_r, u_s, u_k form a path of length 2. If $n = 5$, then only one vertex is left other than u_1, u_2, u_i, u_j , and the graph is either P_5 or a contraction.

□

4. CONCLUSION

In this paper, a study of dominator semi-strong partition and the parameter $\chi_s^d(G)$ is initiated. Further study can be made on the complexity of the parameter and Nordhaus-Gaddum type results for $\chi_s^d(G)$.

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FIGURE 7. A set of graphs $G_1, G_2, G_3, G_4, G_5, G_5, G_7$ with $n = 5$

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IDEAL CONVERGENCE OF A SEQUENCE OF CHEBYSHEV RADII OF SETS

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Abstract. In this paper, we investigate the diameters, Chebyshev radii, Chebyshev self-radii and inner radii of a sequence of sets in the normed spaces. We prove that if a sequence of sets is I -Hausdorff convergent to a set, the sequence of Chebyshev radii of that sequence is \mathcal{I} -convergent. Similar relations are showed for the sequence of diameters, Chebyshev self-radii and inner radii of that sequence.

1. INTRODUCTION

The concept of statistical convergence, which is a generalization of the ordinary convergence of sequences, was first introduced by Fast $\boxed{3}$ and Stainhaus $\boxed{13}$, independently. Fridy $\boxed{4}$ $\boxed{5}$ contributed greatly to the development of the theory of statistical convergence. In 2000, Kostyrko et al [\[7\]](#page-57-4) introduced ideal convergence, which is a generalization of statistical convergence. Recently the ideal convergence theory continues to be popularly studied (see $[9,10]$ $[9,10]$). On the other hand, Hausdorff convergence of a sequence of sets, which is defined by the Hausdorff distance, corresponds to the uniform convergence of the sequence of distance (see $[2, 6, 8]$ $[2, 6, 8]$ $[2, 6, 8]$). The theory of statistical convergence and the theory of ideal convergence were combined with the theory of convergence of sequences of sets by Nuray and Rhoades [\[11\]](#page-57-10) and by Talo and Sever $[14]$, respectively.

In [\[12\]](#page-58-2), Papini and Wu examined Kuratowski convergence and Hausdorff convergence of sequences of sets in Banach spaces. They showed that if a sequence of sets is Hausdorff convergent then the sequences of diameters, Chebyshev radii, Chebyshev self-radii, and inner radii, respectively, of this sequence are convergent.

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In this study, by generalizing some of the results in $[12]$, we show that if a sequence $(A_n)_{n\in\mathbb{N}}$ of sets is *I*-Hausdorff convergent to a set A then the sequence of Chebyshev radii of A_n 's is $\mathcal I$ -convergent to the Chebyshev radius of A. We give similar relations for diameter, relative Chebyshev radius, Chebyshev self-radius and inner radius.

2. Preliminaries

Let $(X, \|\cdot\|)$ be normed space. We denote the family of all nonempty closed subsets, the family of all nonempty closed and bounded subsets and the family of all nonempty closed, convex and bounded subsets of X by $Cl(X)$, $\mathcal{B}(X)$ and $\mathcal{C}(X)$, respectively.

The distance $d(x, A)$ from a point $x \in X$ to a subset A of X is defined to be

$$
d(x,A)=\inf_{a\in A}\|x-a\|\,.
$$

The set A is said to be *bounded* if $\text{diam}(A) < \infty$, where *diameter* $\text{diam}(A)$ of a nonempty set A in a normed space $(X, \|\cdot\|)$ is defined by

$$
diam(A) = \sup_{a_1, a_2 \in A} ||a_1 - a_2||.
$$

The open ball with centre $x \in X$ and radius $\delta > 0$ is the set

$$
S(x, \delta) = \{ y \in X : ||x - y|| < \delta \}.
$$

Hausdorff distance of sets $A, B \subseteq X$ is defined as

$$
H(A, B) = \max \{ h(A, B), h(B, A) \}
$$

where $h(A, B) = \sup_{a \in A} d(a, B)$, or equivalently

$$
H(A, B) = \inf \{ \varepsilon > 0 : A \subseteq B^{\varepsilon} \text{ and } B \subseteq A^{\varepsilon} \}
$$

where $A^{\varepsilon} = \bigcup_{a \in A} \{x \in X : ||x - a|| < \varepsilon\} = \{x \in X : d(x, A) < \varepsilon\}$ is the ε -enlargement of A.

Briefly, we recall some of basic notations in the theory of I−convergence and we refer readers to $\boxed{7, 8}$ for more details. A family $\mathcal{I} \subseteq 2^{\mathbb{N}}$ of subsets of $\boxed{\mathbb{N}}$ is said to be an ideal in N if $\emptyset \in \mathcal{I}$, and $A \cup B \in \mathcal{I}$ for each $A, B \in \mathcal{I}$, and $B \in \mathcal{I}$ for each $A \in \mathcal{I}$ such that $B \subseteq A$ (see \mathcal{B}). An ideal is called *proper* if $\mathbb{N} \notin \mathcal{I}$, and a proper ideal is called *admissible* if ${n \in \mathcal{I}$ for each $n \in \mathbb{N}$. Obviously, an admissible ideal includes all finite subset of \mathbb{N} (see [\[7\]](#page-57-4)).

The definition of ideal convergence for real numbers is as follows: Let $(x_n)_{n\in\mathbb{N}}$ be a sequence in $\mathbb R$ and $x_0 \in \mathbb R$. Let $\mathcal I$ be any ideal on $\mathbb N$. If for every $\varepsilon > 0$

$$
\{n \in \mathbb{N} : |x_n - x_0| \ge \varepsilon\} \in \mathcal{I}
$$

then (x_n) is said to be ideal convergent (briefly, *I*-convergent) to x_0 . Then we write $\mathcal{I} - \lim x_n = x_0$ (see [\[7\]](#page-57-4)).

Define $\mathcal{I}_f = \{A \subset \mathbb{N} : \text{the set } A \text{ has finite number of elements}\}\.$ Then \mathcal{I}_f -convergence and classical convergence is equivalent to each other. Similarly, if we denote $\mathcal{I}_d = \{A \subset \mathbb{N} : \text{the set } A \text{ has natural density zero}\},\$ then \mathcal{I}_d -convergence and statistical convergence is equivalent to each other. We note that the ideals \mathcal{I}_f and \mathcal{I}_d are admissible.

Let $\mathcal{I} \subset 2^{\mathbb{N}}$ be a proper ideal in \mathbb{N} . We say that the sequence (A_n) is \mathcal{I} -Hausdorff convergent to the set A if

$$
\left\{ n \in \mathbb{N} : \sup_{x \in X} |d(x, A_n) - d(x, A)| \ge \varepsilon \right\} \in \mathcal{I}
$$

for every $\varepsilon > 0$, or if $\mathcal{I} - \lim H(A_n, A) = 0$, i.e., for every $\varepsilon > 0$

 ${n \in \mathbb{N} : H(A_n, A) \geq \varepsilon} \in \mathcal{I}$

or equivalently

$$
\{n \in \mathbb{N} : h(A_n, A) \ge \varepsilon \text{ or } h(A, A_n) \ge \varepsilon\} \in \mathcal{I}.
$$

In this case, we write $A_n \stackrel{\mathcal{I} \text{-} H}{\longrightarrow} A$ (see [\[14\]](#page-58-1)).

Now, we list some definitions of radii and centers associated with these radii (see $[1, 12, 15]$ $[1, 12, 15]$ $[1, 12, 15]$). Let A be a bounded subset of X and $Y \subseteq X$.

Example 1. Consider the normed space $(\mathbb{R}^2, \|\cdot\|_1)$ where $\|\cdot\|_1$ is the ℓ_1 norm (aka the taxicab norm). Let A be a square whose vertices are on the points $(-1, -1)$, $(-1, 1), (1, -1)$ and $(1, 1),$ and let $Y = \{(x, y) \in \mathbb{R}^2 : x = 3\}$. We have the following results:

$$
R(A) = 2 \nR_A(A) = 3 \nR'(A) = 0 \nR'(A) = 5 \nR'(A) = 5 \nX'(A) = 5 \nX'(A) = { (0,0) } \nZ'(A) = { (0,0) }, (1,0) , (0,-1) , (0,1) }
$$
\n
$$
R'(A) = 5 \nZ'(A) = { (3,0) }
$$

Lemma 1. Let $A \in \mathcal{B}(X)$, $Y \subseteq X$ and $\varepsilon > 0$. Then the following is provided:

- (i) diam $(A^{\varepsilon}) \leq \text{diam}(A) + 2\varepsilon$
- (ii) $R(x, A^{\varepsilon}) \leq R(x, A) + \varepsilon$ for every $x \in X$
- (iii) $R_Y(A^{\varepsilon}) \le R_Y(A) + \varepsilon$
- (iv) $R(A^{\varepsilon}) \leq R(A) + \varepsilon$

$$
(v) R_{A^{\varepsilon}}(A^{\varepsilon}) \le R_A(A) + \varepsilon
$$

Proof. (i)

 $\alpha_1, \alpha_2 \in A^{\varepsilon} \Longrightarrow \exists a_1, a_2 \in A$ such that $\|\alpha_1 - a_1\| < \varepsilon$ and $\|\alpha_2 - a_2\| < \varepsilon$

Then, for every $\alpha_1, \alpha_2 \in A^{\varepsilon}$ we have

$$
\|\alpha_1 - \alpha_2\| \le \|\alpha_1 - a_1\| + \|a_1 - a_2\| + \|\alpha_2 - a_2\|
$$

<
$$
< \|a_1 - a_2\| + 2\varepsilon
$$

$$
\le \sup_{a_1, a_2 \in A} \|a_1 - a_2\| + 2\varepsilon
$$

$$
= \text{diam}(A) + 2\varepsilon
$$

and so

$$
\text{diam}\left(A^{\varepsilon}\right) = \sup_{\alpha_1,\alpha_2 \in A^{\varepsilon}} \|\alpha_1 - \alpha_2\| \le \text{diam}\left(A\right) + 2\varepsilon.
$$

(ii)

$$
\alpha \in A^{\varepsilon} \Longrightarrow \exists a \in A \text{ such that } ||\alpha - a|| < \varepsilon
$$

Let $x \in X$. For every $\alpha \in A^{\varepsilon}$ we have

$$
\|\alpha - x\| \le \|\alpha - a\| + \|a - x\|
$$

$$
< \|\alpha - x\| + \varepsilon
$$

$$
\le \sup_{a \in A} \|a - x\| + \varepsilon
$$

$$
= R(x, A) + \varepsilon
$$

and so

$$
R(x, A^{\varepsilon}) = \sup_{\alpha \in A^{\varepsilon}} \|\alpha - x\| \le R(x, A) + \varepsilon.
$$

(iii) From (ii), we have $R(y, A^{\varepsilon}) \leq R(y, A) + \varepsilon$ for every $y \in Y$. Then we get

$$
\inf_{y \in Y} R(y, A^{\varepsilon}) \leq \inf_{y \in Y} R(y, A) + \varepsilon
$$

$$
R_Y(A^{\varepsilon}) \leq R_Y(A) + \varepsilon.
$$

- (iv) It is easily obtained by taking $Y = X$ in (iii).
- (v) From (ii), we have

$$
R(a, A^{\varepsilon}) \le R(a, A) + \varepsilon
$$

for every $a \in A$, and so

$$
\inf_{a\in A} R(a, A^{\varepsilon}) \leq \inf_{a\in A} R(a, A) + \varepsilon.
$$

From the fact that

$$
\inf_{\alpha \in A^{\varepsilon}} R(\alpha, A^{\varepsilon}) \leq \inf_{a \in A} R(a, A^{\varepsilon}),
$$

we get

$$
\inf_{\alpha \in A^{\varepsilon}} R(\alpha, A^{\varepsilon}) \leq \inf_{a \in A} R(a, A) + \varepsilon
$$

$$
R_{A^{\varepsilon}}(A^{\varepsilon}) \leq R_A(A) + \varepsilon.
$$

□

We cannot give similar results above for the inner radius, i.e., the inequality $R'(A^{\varepsilon}) \leq R'(A) + \varepsilon$ may not be satisfied. Such as, if we take $\varepsilon = \frac{3}{2}$ in Example $\boxed{1}$, we get

$$
R'(A^{\varepsilon}) = \frac{5}{2} \nleq R'(A) + \varepsilon = 0 + \frac{3}{2}.
$$

Also, we cannot say a general upper bound for the difference $R'(A^{\varepsilon}) - R'(A)$. For example, in the Euclidean space \mathbb{R}^2 , let the set A be a spiral with $r = \theta$ $(0 \le \theta \le 2n\pi, n \in \mathbb{N})$ polar equation. Let's take $\varepsilon > \pi$. Then we have $R'(A) = 0$ and $R'(A^{\varepsilon}) \geq (2n-1)\pi$. Thus the difference $R'(A^{\varepsilon}) - R'(A)$ depends not only on ε but also on n.

3. Main Results

For a sequence of closed and bounded sets, we show that $\mathcal{I}\text{-Hausdorff}$ convergence implies I-convergence of the sequence of Chebyshev radii (diameters, relative Chebyshev radii and Chebyshev self-radii, respectively) of this sequence. If the sets are convex as an additional condition, this proposition is also true for the sequence of inner radii.

Proposition 1. Let $A, A_n \in \mathcal{B}(X)$ ($n \in \mathbb{N}$) and $Y \subseteq X$. If $A_n \stackrel{\mathcal{T}-H}{\longrightarrow} A$ then the following hold:

- (i) \mathcal{I} lim diam (A_n) = diam (A) (ii) $\mathcal{I} - \lim R_Y (A_n) = R_Y (A)$ (iii) $\mathcal{I} - \lim R(A_n) = R(A)$
- (iv) $\mathcal{I} \lim R_{A_n} (A_n) = R_A (A)$

Proof. (i) Let
$$
\varepsilon > 0
$$
. From $A_n \xrightarrow{\mathcal{I} - H} A$ we have

$$
L(\varepsilon) := \left\{ n \in \mathbb{N} : H(A_n, A) \ge \frac{\varepsilon}{3} \right\} \in \mathcal{I}.
$$

For every $n \in \mathbb{N} \setminus L(\varepsilon)$ we have

$$
A \subseteq A_n^{\varepsilon/3} \text{ and } A_n \subseteq A^{\varepsilon/3}.
$$

Then

$$
A \subseteq A_n^{\varepsilon/3} \implies \text{diam}(A) \le \text{diam}\left(A_n^{\varepsilon/3}\right) \le \text{diam}(A_n) + \frac{2\varepsilon}{3}
$$

$$
\implies \text{diam}(A) - \text{diam}(A_n) \le \frac{2\varepsilon}{3}
$$

$$
A_n \subseteq A^{\varepsilon/3} \implies \text{diam}(A_n) \le \text{diam}(A^{\varepsilon/3}) \le \text{diam}(A) + \frac{2\varepsilon}{3}
$$

$$
\implies \text{diam}(A_n) - \text{diam}(A) \le \frac{2\varepsilon}{3}
$$

for every $n \in \mathbb{N} \setminus L(\varepsilon)$. Hence we get

$$
\{n \in \mathbb{N} : |\text{diam}(A_n) - \text{diam}(A)| \ge \varepsilon\} \subseteq L(\varepsilon) \in \mathcal{I}
$$

$$
\{n \in \mathbb{N} : |\text{diam}(A_n) - \text{diam}(A)| \ge \varepsilon\} \in \mathcal{I}
$$

for every $\varepsilon > 0$. Consequently, we obtain $\mathcal{I} - \lim \text{diam}(A_n) = \text{diam}(A)$.

(ii) Let Y be any subset of X . From the triangle inequality, we have

$$
||a_n - y|| - ||a - y|| \le ||a_n - a||
$$

\n
$$
||a - y|| - ||a_n - y|| \le ||a_n - a||
$$
\n(1)
\n
$$
||a - y|| - ||a_n - y|| \le ||a_n - a||
$$
\n(2)

where $y \in Y$, $a_n \in A_n$ and $a \in A$. Then, from \Box

$$
\inf_{a \in A} (||a_n - y|| - ||a - y||) \leq \inf_{a \in A} ||a_n - a||
$$

\n
$$
||a_n - y|| - \sup_{a \in A} ||a - y|| \leq \inf_{a \in A} ||a_n - a||
$$

\n
$$
\sup_{a_n \in A_n} ||a_n - y|| - \sup_{a \in A} ||a - y|| \leq \sup_{a_n \in A_n} \inf_{a \in A} ||a_n - a||
$$

\n
$$
R_Y(A_n) - R_Y(A) = \inf_{\substack{y \in Y \ a_n \in A_n}} ||a_n - y|| - \inf_{y \in Y} \sup_{a \in A} ||a - y||
$$

\n
$$
\leq \sup_{a_n \in A_n} \inf_{a \in A} ||a_n - a|| = h(A_n, A)
$$
\n(3)

and similarly, from (2)

$$
R_Y(A) - R_Y(A_n) = \inf_{y \in Y} \sup_{a \in A} \|a - y\| - \inf_{y \in Y} \sup_{a_n \in A_n} \|a_n - y\|
$$

\$\leq\$ sup inf $\inf_{a \in A} a_n \in A_n$ $||a_n - a|| = h(A, A_n).$ (4)

Take $\varepsilon > 0$. From $A_n \stackrel{\mathcal{T}-H}{\longrightarrow} A$, we have

$$
L(\varepsilon) := \{ n \in \mathbb{N} : h(A_n, A) \ge \varepsilon \text{ or } h(A, A_n) \ge \varepsilon \} \in \mathcal{I}.
$$

From (3) and (4) , we get

$$
R_Y(A_n) - R_Y(A) \le h(A_n, A) < \varepsilon,
$$
\n
$$
R_Y(A) - R_Y(A_n) \le h(A, A_n) < \varepsilon
$$

and so

$$
|R_Y(A_n) - R_Y(A)| < \varepsilon
$$

for every $n \in \mathbb{N} \setminus L(\varepsilon)$. Hence we get

$$
\{n \in \mathbb{N} : |R_Y(A_n) - R_Y(A)| \ge \varepsilon\} \subseteq L(\varepsilon) \in \mathcal{I}
$$

$$
\{n \in \mathbb{N} : |R_Y(A_n) - R_Y(A)| \ge \varepsilon\} \in \mathcal{I}
$$

for every $\varepsilon > 0$. This means that $\mathcal{I} - \lim R_Y (A_n) = R_Y (A)$.

- (iii) It is the special case of (ii), with $Y = X$.
- (iv) Let $\varepsilon > 0$. From $A_n \stackrel{\mathcal{I}-H}{\longrightarrow} A$ we have

$$
L(\varepsilon) := \left\{ n \in \mathbb{N} : h(A_n, A) \ge \frac{\varepsilon}{2} \text{ or } h(A, A_n) \ge \frac{\varepsilon}{2} \right\} \in \mathcal{I}.
$$

If $a_0 \in Z_A(A)$ then $a_0 \in A$ and

$$
R(a_0, A) = \sup_{a \in A} ||a - a_0|| = R_A(A).
$$
 (5)

Take $n \in \mathbb{N} \setminus L(\varepsilon)$. From $h(A, A_n) < \frac{\varepsilon}{2}$ we have

$$
\sup_{a \in A} d(a, A_n) < \frac{\varepsilon}{2}.\tag{6}
$$

From the closeness of A_n there exists an $a_n^{(1)} \in A_n$ such that

$$
\left\|a_0 - a_n^{(1)}\right\| < \frac{\varepsilon}{2}.\tag{7}
$$

Also, there exists an $a_n^{(2)} \in A_n$ such that

$$
\sup_{a_n \in A_n} \|a_n - a_n^{(1)}\| = \|a_n^{(2)} - a_n^{(1)}\|.
$$
\n(8)

From $h(A_n, A) < \frac{\varepsilon}{2}$ we get

$$
d\left(a_n^{(2)}, A\right) \le \sup_{a_n \in A_n} d\left(a_n, A\right) < \frac{\varepsilon}{2} \tag{9}
$$

and so

$$
\left\|a_0 - a_n^{(2)}\right\| < R_A\left(A\right) + \frac{\varepsilon}{2}.\tag{10}
$$

From $\boxed{7}$ and $\boxed{10}$ we obtain

$$
R_{A_n}(A_n) \leq \|a_n^{(1)} - a_n^{(2)}\| \leq \|a_n^{(1)} - a_0\| + \|a_0 - a_n^{(2)}\| < R_A(A) + \varepsilon
$$

for every $n \in \mathbb{N} \setminus L(\varepsilon)$. Similarly, it can be shown that

$$
R_{A}\left(A\right) < R_{A_{n}}\left(A_{n}\right) + \varepsilon
$$

for every $n \in \mathbb{N} \setminus L(\varepsilon)$. Consequently, we get

$$
\{n \in \mathbb{N} : |R_{A_n}(A_n) - R_A(A)| \geq \varepsilon\} \subseteq L(\varepsilon) \in \mathcal{I}
$$

$$
\{n \in \mathbb{N} : |R_{A_n}(A_n) - R_A(A)| \geq \varepsilon\} \in \mathcal{I}
$$

for every $\varepsilon > 0$, and so $\mathcal{I} - \lim R_{A_n} (A_n) = R_A (A)$.

□

Lemma 2. (see $\sqrt{12}$, Lemma 1) Let $A, B \in \mathcal{C}(X)$. If $R'(A) > 0$ and $H(A, B) <$ $R'(A)$ $\frac{1}{2}$ then

$$
R'(B) \ge R'(A) - H(A, B) > 0.
$$

As a result of the above lemma we can give the following corollary.

Corollary 1. Let $A \in \mathcal{C}(X)$ and $\varepsilon > 0$. If $R'(A^{\varepsilon}) > 2\varepsilon$ then $R'(A) \ge R'(A^{\varepsilon}) - \varepsilon$

(That is, $R'(A^{\varepsilon}) \leq R'(A) + \varepsilon$). Of course, for the condition here to be satisfied, $R'(A) > \varepsilon$ must be.

Proposition 2. Let $A, A_n \in \mathcal{C}(X)$ ($n \in \mathbb{N}$). If $A_n \stackrel{\mathcal{I}-H}{\longrightarrow} A$ then $\mathcal{I} - \lim R'(A_n) = R'(A)$.

Proof. First let's assume that $R'(A) = 0$. Suppose that $\mathcal{I} - \lim R'(A_n) \neq 0$. Then there is an $\varepsilon_0 > 0$ such that

$$
K(\varepsilon_0) := \{ n \in \mathbb{N} : R'(A_n) \ge \varepsilon_0 \} \notin \mathcal{I}.
$$

From $A_n \stackrel{\mathcal{I}-H}{\longrightarrow} A$ we have

$$
L(\varepsilon_0) := \left\{ n \in \mathbb{N} : H(A_n, A) \ge \frac{\varepsilon_0}{2} \right\} \in \mathcal{I}.
$$

Then $(\mathbb{N} \setminus L(\varepsilon_0)) \cap K(\varepsilon_0) \neq \emptyset$ and so we have

$$
H(A_n, A) < \frac{\varepsilon_0}{2} \le \frac{1}{2} R'(A_n)
$$

for every $n \in (\mathbb{N} \setminus L(\varepsilon_0)) \cap K(\varepsilon_0)$. From Lemma [2,](#page-56-0) we get

$$
R'(A) > 0
$$

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and this is a contradiction. Therefore, $\mathcal{I} - \lim R'(A_n) = 0 = R'(A)$ holds. Now let's assume that $R'(A) > 0$. Let $0 < \varepsilon < \frac{R'(A)}{2}$ $\frac{(A)}{3}$. From $A_n \stackrel{\mathcal{T}-H}{\longrightarrow} A$ we have

 $L(\varepsilon) := \{ n \in \mathbb{N} : H(A_n, A) \geq \varepsilon \} \in \mathcal{I}.$

Then we have

$$
H(A_n, A) < \varepsilon < \frac{R'(A)}{3} < \frac{R'(A)}{2}
$$

for every $n \in \mathbb{N} \setminus L(\varepsilon)$. From Lemma $\boxed{2}$, we get

$$
R'(A_n) \ge R'(A) - H(A_n, A) > R'(A) - \varepsilon
$$
\n(11)

for every $n \in \mathbb{N} \setminus L(\varepsilon)$. We also have

$$
H(A_n, A) < \varepsilon < \frac{1}{2} \left(R'(A) - \varepsilon \right) < \frac{R'(A_n)}{2}
$$

for every $n \in \mathbb{N} \setminus L(\varepsilon)$. Again from Lemma $\boxed{2}$, we get

$$
R'(A) \ge R'(A_n) - H(A, A_n) > R'(A_n) - \varepsilon \tag{12}
$$

for every $n \in \mathbb{N} \setminus L(\varepsilon)$. From (11) and (12) we obtain

$$
\{n \in \mathbb{N} : |R'(A_n) - R'(A)| \ge \varepsilon\} \subseteq L(\varepsilon) \in \mathcal{I}
$$

$$
\{n \in \mathbb{N} : |R'(A_n) - R'(A)| \ge \varepsilon\} \in \mathcal{I}
$$

and so $\mathcal{I} - \lim R'(A_n) = R'(A).$

for every $\varepsilon > 0$, and so $\mathcal{I} - \lim R'(A_n) = R'$

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PARAMETER UNIFORM SECOND-ORDER NUMERICAL APPROXIMATION FOR THE INTEGRO-DIFFERENTIAL EQUATIONS INVOLVING BOUNDARY LAYERS

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ABSTRACT. The work handles a Fredholm integro-differential equation involving boundary layers. A fitted second-order difference scheme has been created on a uniform mesh utilizing interpolating quadrature rules and exponential basis functions. The stability and convergence of the proposed discretization technique are analyzed and one example is solved to display the advantages of the presented technique.

1. INTRODUCTION

In the study, we deal with singularly perturbed Fredholm integro-differential equation (SPFIDE) in the form:

$$
Lv := L_1 v + \lambda \int_0^l M(x, \zeta) v(\zeta) d\zeta = f(x), \qquad 0 < x < 1,\tag{1}
$$

$$
v(0) = A, \quad v(l) = B,\tag{2}
$$

where $L_1v = -\varepsilon v''(x) + a(x)v(x), 0 < \varepsilon \ll 1$ is a singular perturbation parameter, λ is a given parameter. The functions $a(x) \geq \alpha > 0$, $f(x)$ and $M(x, \zeta)$ are considered to be sufficiently smooth and satisfy certain regularity criteria. The solution $v(x)$ of (1) - (2) has in general boundary layers near $x = 0$ and $x = l$.

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Singularly perturbed problems (SPPs) are defined by a small parameter ε multiplying the highest order derivative term. The solution to them generally involves boundary or initial layers. To quote a few, the exact solutions of SPPs and their applications may be found in $\boxed{15}$, $\boxed{18}$, $\boxed{21}$. SPPs have a wide range of applications in the fields of population dynamics, nanofluid, neurobiology, fluid dynamics, viscoelasticity, heat transfer problems, simultaneous control systems and mathematical biology etc. It is worth noting that when a small ε parameter is multiplied with the derivative, the great majority of classic numerical techniques on uniform meshes are ineffective at solving issues unless the step-size of discretization is drastically reduced. Thus, as the perturbation parameter ε goes smaller, the truncation error becomes boundless. To solve SPPs numerically, general approaches are done with the fitted finite difference method and are widely utilized $[9, 12, 19, 20]$ $[9, 12, 19, 20]$ $[9, 12, 19, 20]$ $[9, 12, 19, 20]$.

Most engineering applications and scientific disciplines have been expressed by Fredholm integro-differential equations (FIDEs). Plasma physics, biomechanics, financial mathematics, artificial neural networks, oceanopraphy, epidemic models, electromagnetic theory, fluid mechanics, biological and population dynamics processes are amongst these (see, e.g., $\sqrt{5}$, $\sqrt{7}$, $\sqrt{13}$). For this reason, several studies have been conducted on FIDEs. Solving problems of this type is quite difficult. Therefore, we require robust and consistent numerical methods $[6, 8, 14, 16, 23, 26]$ $[6, 8, 14, 16, 23, 26]$ $[6, 8, 14, 16, 23, 26]$ $[6, 8, 14, 16, 23, 26]$ $[6, 8, 14, 16, 23, 26]$ $[6, 8, 14, 16, 23, 26]$ (see, as well as the references therein).

These investigations in relation to FIDEs are just in relation to regular situations. Numerical examination of SPFIDEs has not been widespread till recently. Finite difference schemes for solving linear SPFIDEs are constructed in $\left[\frac{1}{2}\right]$. A second order numerical tecnique for solving FIDE with boundary layer is developed in $[10, 11]$ $[10, 11]$.

The goal of this work is to propose a uniform convergence numerical technique to solve linear second-order FIDEs with boundary layers. A numerical technique that uses suitable interpolating quadrature rules and exponential basis functions is proposed on a uniform mesh. Error estimates are acquired in the discrete maximum norm with regard to the perturbation parameter. To corroborate theoretical estimates, numerical experiments are conducted and the results are presented.

The rest of the contents is organized kind of following. In Section 2, some properties of solutions (1) – (3) are presented, as well as a finite difference scheme. In Section 3, the stability and convergence analysis of this scheme are shown. In Section 4, the numerical results of an example to verify the theoretical estimates are presented. Finally, the work ends with a summary of the conclusions in Section 5.

2. Discretization Techniques

We have mentioned certain analytical bounds here, which we will use later in our error analysis.

Lemma 1. Let $a, f \in C^2[0, l], \frac{\partial^m M}{\partial x^m} \in C[0, l]^2$, $(m = 0, 1, 2)$ and

$$
|\lambda| < \frac{\alpha}{\max\limits_{0\leq x\leq l} \int\limits_{0}^{l} |M(x,\zeta)|\,d\zeta}.
$$

Then the solution $u(x)$ of the problem $\overline{1}$. $\overline{2}$ satisfies the following estimates:

$$
||v||_{\infty} \leq C,\tag{3}
$$

.

$$
\left|v^{(k)}(x)\right| \le C\left\{1+\varepsilon^{-\frac{k}{2}}\left(e^{-\frac{\sqrt{\alpha}x}{\sqrt{\varepsilon}}}+e^{-\frac{\sqrt{\alpha}(1-x)}{\sqrt{\varepsilon}}}\right)\right\}, \quad (k=1,2), \quad 0 \le x \le 1. \tag{4}
$$

Proof. The proof of Lemma $\overline{1}$ is by like approach as in $\overline{2}$, $\overline{10}$, $\overline{17}$.

Let ω_N be an equidistant mesh on [0, l]:

$$
\omega_N = \{x_i = ih, i = 1, 2, ..., N - 1, h = lN^{-1}\}, \quad \bar{\omega}_N = \omega_N \cup \{x_0 = 0, x_N = l\}.
$$

We utilize the following difference approximations for any mesh function $q(x)$ defined on $\bar{\omega}_N$:

$$
q_{x,i} = \frac{q_{i+1} - q_i}{h}, \qquad q_{\bar{x},i} = \frac{q_i - q_{i-1}}{h}, \qquad q_{\bar{x}_x,i} = \frac{q_{x,i} - q_{\bar{x},i}}{h}.
$$

For the equation $\left(1\right)$, we begin with the following integral identity:

$$
\frac{1}{\chi_i h} \int_{x_{i-1}}^{x_{i+1}} Lv(x) \psi_i(x) dx = \frac{1}{\chi_i h} \int_{x_{i-1}}^{x_{i+1}} f(x) \psi_i(x) dx, \qquad 1 \le i \le N - 1,
$$
 (5)

with the basis functions

$$
\psi(x) = \begin{cases}\n\psi_i^{(1)}(x) \equiv \frac{\sinh\gamma_i(x - x_i)}{\sinh\gamma_i h}, & x \in (x_{i-1}, x_i), \\
\psi_i^{(2)}(x) \equiv \frac{\sinh\gamma_i(x_{i+1} - x)}{\sinh\gamma_i h}, & x \in (x_i, x_{i+1}), \\
0, & x \notin (x_{i-1}, x_{i+1}),\n\end{cases}
$$

where

$$
\gamma_i = \sqrt{\frac{a(x_i)}{\varepsilon}}, \qquad \chi_i = \frac{1}{h} \int_{x_{i-1}}^{x_{i+1}} \psi_i(x) dx = \frac{2 \tanh(\gamma_i h/2)}{\gamma_i h}
$$

We should remark that the functions $\psi_i^{(1)}$ and $\psi_i^{(2)}$ are the solutions to the following problems:

$$
-\varepsilon\psi'' + a_i\psi = 0, \quad x_{i-1} < x < x_i, \quad \psi(x_{i-1}) = 0, \quad \psi(x_i) = 1, \n-\varepsilon\psi'' + a_i\psi = 0, \quad x_i < x < x_{i+1} \quad \psi(x_i) = 1, \quad \psi(x_{i+1}) = 0.
$$

By using the technique of the exact difference approximations $[3, 4, 11, 24, 25]$ $[3, 4, 11, 24, 25]$ $[3, 4, 11, 24, 25]$ $[3, 4, 11, 24, 25]$ $[3, 4, 11, 24, 25]$ (see also $[22]$, pp. 207-214), it follows that

$$
-\frac{\varepsilon}{\chi_{i}h} \int_{x_{i-1}}^{x_{i+1}} \psi_{i}(x)v''(x)dx + \frac{a_{i}}{\chi_{i}h} a_{i} \int_{x_{i-1}}^{x_{i+1}} \psi_{i}(x)v(x)dx =
$$

$$
-\frac{\varepsilon}{\chi_{i}} \left\{ 1 + a_{i}\varepsilon^{-1} \int_{x_{i-1}}^{x_{i}} \psi_{i}^{(1)}(x)(x - x_{i})dx \right\} v_{\bar{x}x,i} + \frac{a_{i}}{\chi_{i}} \left\{ h^{-1} \int_{x_{i-1}}^{x_{i}} \psi_{i}^{(1)}dx + h^{-1} \int_{x_{i}}^{x_{i+1}} \psi_{i}^{(2)}dx \right\} v_{i} = -\varepsilon \theta_{i}v_{\bar{x}x,i} + a_{i}v_{i}
$$

where

$$
\theta_i = \frac{a_i \rho^2}{4 \sinh^2 \left(\sqrt{a_i \frac{\rho}{2}}\right)}, \quad \left(\rho = \frac{h}{\sqrt{\varepsilon}}\right). \tag{6}
$$

Thus

$$
\frac{1}{\chi_{i}h} \int_{x_{i-1}}^{x_{i+1}} \varepsilon v''(x) \psi_{i}(x) dx + \frac{1}{\chi_{i}h} \int_{x_{i-1}}^{x_{i+1}} a(x) v(x) \psi_{i}(x) dx = -\varepsilon \theta_{i} v_{\bar{x}x, i} + a_{i} v_{i} + R_{i}^{(1)},
$$
\n(7)

with remainder term

$$
R_i^{(1)} = \frac{1}{\chi_i h} \int\limits_{x_{i-1}}^{x_{i+1}} [a(x) - a(x_i)] \, v(x) \psi_i(x) dx.
$$
 (8)

Furthermore, for the right-side in $\left(\overline{\mathbf{5}}\right)$ we get

$$
\frac{1}{\chi_i h} \int_{x_{i-1}}^{x_{i+1}} f(x) \psi_i(x) dx = f_i + R_i^{(2)},
$$
\n(9)

with remainder term

$$
R_i^{(2)} = \frac{1}{\chi_i h} \int\limits_{x_{i-1}}^{x_{i+1}} [f(x) - f(x_i)] \psi_i(x) dx.
$$
 (10)

For integral term that include the kernel function, from $\left(\overline{\boldsymbol{5}}\right)$, we have

$$
\frac{\lambda}{\chi_i h} \int_{x_{i-1}}^{x_{i+1}} dx \psi_i(x) \int_0^l M(x, \zeta) v(\zeta) d\zeta = \lambda \int_0^l M(x_i, \zeta) v(\zeta) d\zeta
$$

$$
+\frac{\lambda}{\chi_i h} \int\limits_{x_{i-1}}^{x_{i+1}} dx \psi_i(x) \int\limits_{x_{i-1}}^{x_{i+1}} \left(\int\limits_0^l \frac{\partial^2 M\left(\xi, \zeta\right)}{\partial \xi^2} v\left(\zeta\right) d\zeta \right) M_1\left(x, \xi\right) d\xi,
$$

where

$$
M_1(x,\xi) = T_1(x-\xi) - T_1(x_i-\xi) + (2h)^{-1}(x_{i+1} - \xi)(x_i - x),
$$

\n
$$
T_1(\lambda) = \lambda, \quad \lambda \ge 0; \qquad T_1(\lambda) = 0 \quad \lambda < 0.
$$

We computed by using composite trapezoidal integration with the remainder term in integral form for the second integral term in the left side of the identity of (5):

$$
\int_{0}^{l} M(x_{i}, \zeta)v(\zeta)d\zeta = \sum_{j=0}^{N} \hbar_{j}M_{ij}v_{j} + \frac{1}{2} \sum_{j=1}^{N} \int_{x_{j-1}}^{x_{j}} (x_{j} - \xi) (x_{j-1} - \xi) (M(x_{i}, \xi)v(\xi))'' d\xi,
$$

where

$$
\hbar_0 = \hbar_N = \frac{h}{2}, \quad \hbar_i = h, \quad 1 \le i \le N - 1.
$$

Thus we get

$$
\frac{\lambda}{\chi_i h} \int_{x_{i-1}}^{x_{i+1}} dx \psi_i(x) \int_0^l M(x, \zeta) v(\zeta) d\zeta = \lambda \sum_{j=0}^N \hbar_j M_{ij} v_j + R_i^{(3)},
$$
\n(11)

with remainder term

$$
R_i^{(3)} = \frac{\lambda}{\chi_i h} \int_{x_{i-1}}^{x_{i+1}} dx \psi_i(x) \int_{x_{i-1}}^{x_{i+1}} \left(\int_0^l \frac{\partial^2 M(\xi, \zeta)}{\partial \xi^2} v(\zeta) d\zeta \right) M_1(x, \xi) d\xi
$$

+
$$
\frac{1}{2} \lambda \sum_{j=1}^N \int_{x_{j-1}}^{x_j} (x_j - \xi) (x_{j-1} - \xi) (M(x_i, \xi) v(\xi))'' d\xi.
$$
 (12)

Combining $\boxed{7}$, $\boxed{9}$ and $\boxed{11}$ in $\boxed{5}$ we obtain the following difference scheme:

$$
L_N v_i := -\varepsilon \theta_i v_{\bar{x}x, i} + a_i v_i + \lambda \sum_{j=0}^N \hbar_j M_{ij} v_j + R_i = f_i, \qquad 1 \le i \le N - 1, \tag{13}
$$

with remainder term

$$
R_i = R_i^{(1)} + R_i^{(2)} + R_i^{(3)},\tag{14}
$$

where the remainder terms $R_i^{(1)}, R_i^{(2)}$ and $R_i^{(3)}$ are defined by $\langle 8 \rangle$, $\langle 10 \rangle$ and $\langle 12 \rangle$ respectively.

Based on [\(13\)](#page-63-2) we achieve the following difference approximate for approximating $(1) - (2)$ $(1) - (2)$ $(1) - (2)$:

$$
L_N y_i := -\varepsilon \theta_i y_{\bar{x}x, i} + a_i y_i + \lambda \sum_{j=0}^N \hbar_j M_{ij} y_j = f_i, \qquad 1 \le i \le N - 1,
$$
 (15)

$$
y_0 = A, \quad y_N = B,\tag{16}
$$

where θ_i is defined by $[6]$.

3. Error Analysis

For the error function $z_i = y_i - v_i$ $(i = 0, 1, ..., N)$ considering [\(13\)](#page-63-2) and [\(15\)](#page-63-3), we get

$$
L_N z_i := R_i, \qquad 1 \le i \le N - 1,\tag{17}
$$

$$
z_0 = 0, \quad z_N = 0,\tag{18}
$$

where the remainder term R_i is defined by (14) .

Theorem 1. Let $\frac{\partial^m M}{\partial x^m} \in C^2[0, l]^2$, $(m = 0, 1, 2)$, $M(x, 0) = M(x, l) = 0$; $a, f \in$ $C^2[0, l], a'(0) = a'(l) = 0, and$

$$
|\lambda| < \frac{\alpha}{\max\limits_{1 \leq i \leq N} \sum\limits_{j=0}^{N} \hbar_j |M_{ij}|}.
$$

Then for the error of the scheme (15) - (16) , we have

$$
||y - v||_{\infty, \bar{\omega}_N} \le Ch^2.
$$

Proof. Applying the discrete maximum principle to discrete problem $(\overline{17})$ and $(\overline{18})$, we get

$$
||z||_{\infty,\bar{\omega}_N} \leq \alpha^{-1} \left\| R - \lambda \sum_{j=0}^N \hbar_j M_{ij} z_j \right\|_{\infty,\omega_N}
$$

$$
\leq \alpha^{-1} ||R||_{\infty,\omega_N} + \alpha^{-1} |\lambda| \max_{1 \leq i \leq N} \sum_{j=0}^N \hbar_j |M_{ij}| ||z||_{\infty,\bar{\omega}_N}.
$$

Hence

$$
||z||_{\infty,\bar{\omega}_N} \le \frac{\alpha^{-1} ||R||_{\infty,\omega_N}}{1 - \alpha^{-1} |\lambda| \max_{1 \le i \le N} \sum_{j=0}^N \hbar_j |M_{ij}|},
$$

which leads to

$$
||z||_{\infty,\overline{\omega}_N} \le C ||R||_{\infty,\omega_N}.
$$
\n(19)

Now we estimate the remainder terms $R_i^{(1)}$, $R_i^{(2)}$ and $R_i^{(3)}$ separately. First we will show that, for $R_i^{(1)}$ the estimate

$$
\left| R_i^{(1)} \right| \le C h^2,\tag{20}
$$

holds. Using relations

$$
v(x) = v(x_i) + (x - x_i) v'(\eta_i), \quad \eta_i \in (x_i, x),
$$

$$
a(x) = a(x_i) + (x - x_i) a'(x_i) + \frac{(x - x_i)^2}{2} a''(\xi_i), \quad \xi_i \in (x_i, x)
$$

and

$$
\int_{x_{i-1}}^{x_{i+1}} (x - x_i) \psi_i(x) dx = 0,
$$

we take

$$
R_{i}^{(1)} = \frac{1}{\chi_{i}h} \int_{x_{i-1}}^{x_{i+1}} [a(x) - a(x_{i})] v(x) \psi_{i}(x) dx = \frac{a'(x_{i}) v(x_{i})}{\chi_{i}h} \int_{x_{i-1}}^{x_{i+1}} (x - x_{i}) \psi_{i}(x) dx + \frac{a'(x_{i})}{\chi_{i}h} \int_{x_{i-1}}^{x_{i+1}} (x - x_{i})^{2} v'(\eta_{i}(x)) \psi_{i}(x) dx + \frac{1}{2\chi_{i}h} \int_{x_{i-1}}^{x_{i+1}} (x - x_{i})^{2} a''(\xi_{i}(x)) v(x) \psi_{i}(x) dx = \frac{a'(x_{i})}{\chi_{i}h} \int_{x_{i-1}}^{x_{i+1}} (x - x_{i})^{2} v'(\eta_{i}(x)) \psi_{i}(x) dx + \frac{1}{2\chi_{i}h} \int_{x_{i-1}}^{x_{i+1}} (x - x_{i})^{2} a''(\xi_{i}(x)) v(x) \psi_{i}(x) dx.
$$
 (21)

Since $a \in C^2[0, l], |v(x)| \leq C$ and $|x - x_i| \leq h$ for the second term in the right side of (21) , we have

$$
\frac{1}{2\chi_{i}h} \left| \int_{x_{i-1}}^{x_{i+1}} (x - x_{i})^{2} a''\left(\xi_{i}\left(x\right)\right) v\left(x\right) \psi_{i}\left(x\right) dx \right| \leq \frac{Ch^{2}}{\chi_{i}h} \int_{x_{i-1}}^{x_{i+1}} \psi_{i}\left(x\right) dx
$$
\n
$$
= O\left(h^{2}\right). \tag{22}
$$

Next, according to Lemma $\boxed{1}$, we take the following inequality

$$
\begin{aligned} |v'(\eta_i)| &\leq C\left\{1+\frac{1}{\sqrt{\varepsilon}}\left(e^{-\frac{\sqrt{\alpha}\eta_i}{\sqrt{\varepsilon}}}+e^{-\frac{\sqrt{\alpha}(l-\eta_i)}{\sqrt{\varepsilon}}}\right)\right\} \\ &\leq C\left\{1+\frac{1}{\sqrt{\varepsilon}}\left(e^{-\frac{\sqrt{\alpha}x_{i-1}}{\sqrt{\varepsilon}}}+e^{-\frac{\sqrt{\alpha}(l-x_{i+1})}{\sqrt{\varepsilon}}}\right)\right\}, \qquad 1 < i < N-1. \end{aligned}
$$

Hence, for the first term in the right side of (21) , we have

$$
\frac{1}{\chi_{i}h} \left| a'(x_{i}) \int_{x_{i-1}}^{x_{i+1}} (x - x_{i})^{2} v'(\eta_{i} (x)) \psi_{i} (x) dx \right| \leq \frac{C}{\chi_{i}h} |a'(x_{i})| \int_{x_{i-1}}^{x_{i+1}} (x - x_{i})^{2} \psi_{i} (x) dx
$$

$$
+ \frac{C}{\sqrt{\varepsilon} \chi_{i}h} |a'(x_{i})| \int_{x_{i-1}}^{x_{i+1}} (x - x_{i})^{2} \psi_{i} (x) e^{-\frac{\sqrt{\alpha} x_{i-1}}{\sqrt{\varepsilon}}} dx
$$

$$
+ \frac{C}{\sqrt{\varepsilon} \chi_{i}h} |a'(x_{i})| \int_{x_{i-1}}^{x_{i+1}} (x - x_{i})^{2} \psi_{i} (x) e^{-\frac{\sqrt{\alpha} x_{i+1}}{\sqrt{\varepsilon}}} dx.
$$
(23)

We can easily view that the first term in the right side of (23) is that $O(h^2)$. From $a'(0) = 0$ and $xe^{-x} \le e^{-\frac{x}{2}}$, $(x \ge 0)$ for the second term of (21) , we have

$$
\begin{split}\n&\left|\frac{C}{\sqrt{\varepsilon}\chi_{i}h}a'\left(x_{i}\right)\int\limits_{x_{i-1}}^{x_{i+1}}\left(x-x_{i}\right)^{2}\psi_{i}\left(x\right)e^{-\frac{\sqrt{\alpha}x_{i-1}}{\sqrt{\varepsilon}}}dx\right| \\
&\leq \frac{C}{\sqrt{\varepsilon}\chi_{i}h}\left|a''\left(\bar{\xi}_{i}\right)\right|e^{-\frac{\sqrt{\alpha}x_{i-1}}{\sqrt{\varepsilon}}}\int\limits_{x_{i-1}}^{x_{i+1}}\left(x-x_{i}\right)^{2}\psi_{i}\left(x\right)dx \\
&\leq Ch^{2}\frac{x_{i}}{\sqrt{\varepsilon}}e^{-\frac{\sqrt{\alpha}x_{i-1}}{\sqrt{\varepsilon}}}\n\\ &\leq Ch^{2}\frac{x_{i}}{x_{i-1}}\frac{x_{i-1}}{\sqrt{\varepsilon}}e^{-\frac{\sqrt{\alpha}x_{i-1}}{\sqrt{\varepsilon}}}\n\\ &\leq Ch^{2}(i-1)^{-1}e^{-\frac{\sqrt{\alpha}x_{i-1}}{2\sqrt{\varepsilon}}}\n\\ &\leq Ch^{2}, \qquad i>1.\n\end{split}
$$

The same evaluation is achieved for the third term in the right side of (23) from $a'(l) = 0$, for $i < N - 1$. Thus, identity (21) is proved for $i = 2, 3, ..., N - 2$. Also for $i = 1$, using relations

$$
a(x) = a(x_1) + (x - x_1) a'(x_1) + \frac{(x - x_1)^2}{2} a''(\xi_1), \quad \xi_1 \in (x_1, x)
$$

and

$$
v(x) = v(x_0) + \int_{x_0}^{x} v'(\xi) d\xi,
$$

we get

$$
R_1^{(1)} = \frac{1}{\chi_1 h} a'(x_1) \int_{x_0}^{x_2} (x - x_1) \left[\int_{x_0}^x v'(\xi) d\xi \right] \psi_1(x) dx
$$

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$$
+\frac{1}{2\chi_{1}h} \int_{x_{0}}^{x_{2}} (x-x_{1})^{2} a''(\xi_{1}(x)) v(x) \psi_{1}(x) dx.
$$
 (24)

From (22) , the second term in the right side of (24) will be $O(h^2)$. From a' (0) and Lemma $\boxed{1}$, we can evaluate the first as following

$$
\begin{split}\n&\left|\frac{a'(x_1)}{\chi_1 h} \int_{x_0}^{x_2} (x - x_1) \left[\int_{x_0}^x v'(\xi) \, d\xi \right] \psi_1(x) \, dx \right| \leq |a'(x_1)| \, h \int_{x_0}^{x_2} |v'(x)| \, dx \\
&\leq C x_1 h \, |a''(\bar{\eta}_1)| \int_{x_0}^{x_2} \left\{ 1 + \frac{1}{\sqrt{\varepsilon}} \left(e^{-\frac{\sqrt{\alpha} x}{\sqrt{\varepsilon}}} + e^{-\frac{\sqrt{\alpha}(1-x)}{\sqrt{\varepsilon}}} \right) \right\} dx \\
&\leq Ch^2 \left\{ h + \frac{1}{\sqrt{\varepsilon}} \int_{x_0}^{x_2} e^{-\frac{\sqrt{\alpha} x}{\sqrt{\varepsilon}}} dx \right\} \\
&\leq Ch^2 \left\{ h + \sqrt{\alpha}^{-1} \left(1 - e^{\frac{2\sqrt{\alpha} h}{\sqrt{\varepsilon}}} \right) \right\} = O\left(h^2\right).\n\end{split}
$$

Thus,

$$
\left| R_1^{(1)} \right| = O\left(h^2 \right)
$$

are proved. The proof of $|R_{N}^{(1)}|$ $\begin{vmatrix} 1 \\ N-1 \end{vmatrix} = O(h^2)$ is similar. So, the inequality $\begin{pmatrix} 20 \\ N \end{pmatrix}$ is proved.

Next, it is not difficult to see that, for $f \in C^2[0, l]$

$$
\left| R_i^{(2)} \right| = O\left(h^2 \right), \qquad 1 \le i \le N - 1. \tag{25}
$$

Finally, for $R_i^{(3)}$ we have

$$
\left| R_i^{(3)} \right| \le \left| \frac{\lambda}{\chi_i h} \int_{x_{i-1}}^{x_{i+1}} dx \psi_i(x) \int_{x_{i-1}}^{x_{i+1}} \left(\int_0^l \frac{\partial^2 M(\xi, \zeta)}{\partial \xi^2} v(\zeta) d\zeta \right) M_1(x, \xi) d\xi \right| + \left| \frac{1}{2} \lambda \sum_{j=1}^N \int_{x_{j-1}}^{x_j} (x_j - \xi) (\xi - x_{j-1}) \left(M(x_i, \xi) v(\xi) \right)'' d\xi \right|.
$$
 (26)

By virtue of boundedness of $\frac{\partial^2 M}{\partial x^2}$, $v(x)$ and $|M_1(x,\zeta)| \leq Ch$ the first term in the right side of (26) will be $O(h^2)$.

Rearranging the second term in the right side of (26) gives

$$
\frac{1}{2} |\lambda| \sum_{j=1}^{N} \int_{x_{j-1}}^{x_j} (x_j - \xi) (\xi - x_{j-1}) | \big(M(x_i, \xi) v(\xi) \big) |^{n} d\xi
$$

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$$
\leq \frac{1}{2} |\lambda| \sum_{j=1}^{N} \int_{x_{j-1}}^{x_j} (x_j - \xi) (\xi - x_{j-1}) |M''(x_i, \xi)| |v(\xi)| d\xi
$$

+
$$
|\lambda| \sum_{j=1}^{N} \int_{x_{j-1}}^{x_j} (x_j - \xi) (\xi - x_{j-1}) |M'(x_i, \xi)| |v'(\xi)| d\xi
$$

+
$$
\frac{1}{2} |\lambda| \sum_{j=1}^{N} \int_{x_{j-1}}^{x_j} (x_j - \xi) (\xi - x_{j-1}) |M(x_i, \xi)| |v''(\xi)| d\xi.
$$
 (27)

Hence, from $|v(x)| \leq C$ and $\frac{\partial^2 M}{\partial x^2} \in C^2[0, l]$ for the first term on the right side (27) will be $O(h^2)$.

For the second term in the right side (27) , we have the estimate

$$
|\lambda| \sum_{j=1}^{N} \int_{x_{j-1}}^{x_j} (x_j - \xi) (\xi - x_{j-1}) |M'(x_i, \xi)| |v'(\xi)| d\xi \le |\lambda| h^2 \int_{0}^{l} |M'(x_i, \xi)| |v'(\xi)| d\xi
$$

$$
\le |\lambda| h^2 \int_{0}^{l} \{ |M'(x_i, \xi)| |v(\xi)| + |M(x_i, \xi)| |v'(\xi)| \} d\xi.
$$

From here using Lemma $\boxed{1}$ it is obtained the estimate

$$
|\lambda| \sum_{j=1}^{N} \int_{x_{j-1}}^{x_j} (x_j - \xi) (\xi - x_{j-1}) |M'(x_i, \xi)| |v'(\xi)| d\xi
$$

\n
$$
\leq C |\lambda| h^2 \int_{0}^{l} \left(1 + 1/\sqrt{\varepsilon} \left(e^{-\frac{\sqrt{\alpha}\xi}{\sqrt{\varepsilon}}} + e^{-\frac{\sqrt{\alpha}(l-\xi)}{\sqrt{\varepsilon}}} \right) \right) d\xi
$$

\n
$$
\leq Ch^2.
$$
\n(28)

For the third term in the right side (27) , by virtue of (4) for $k = 2$, we have

$$
\frac{1}{2} |\lambda| \sum_{j=1}^{N} \int_{x_{j-1}}^{x_j} (x_j - \xi) (\xi - x_{j-1}) |M(x_i, \xi)| |v''(\xi)| d\xi
$$
\n
$$
\leq C \sum_{j=1}^{N} \int_{x_{j-1}}^{x_j} (x_j - \xi) (\xi - x_{j-1}) |M(x_i, \xi)| \left\{ 1 + \frac{1}{\varepsilon} e^{-\frac{\sqrt{\alpha}\xi}{\sqrt{\varepsilon}}} + \frac{1}{\varepsilon} e^{-\frac{\sqrt{\alpha}(1-\xi)}{\sqrt{\varepsilon}}} \right\} d\xi
$$
\n
$$
\leq Ch^2 \left\{ 1 + \sum_{j=1}^{N} \int_{x_{j-1}}^{x_j} |M(x_i, \xi)| \left(\frac{1}{\varepsilon} e^{-\frac{\sqrt{\alpha}\xi}{\sqrt{\varepsilon}}} + \frac{1}{\varepsilon} e^{-\frac{\sqrt{\alpha}(1-\xi)}{\sqrt{\varepsilon}}} \right) \right\}.
$$

Taking into account the relations (the partial derivatives are estimated at intermediate points, as required by the mean value theorem, as indicated by the bar.)

$$
M(x_i, \xi) = M(x, 0) + \frac{\partial \overline{M}}{\partial \xi} \xi, \qquad M(x, 0) = 0,
$$

we get

$$
Ch^2 \sum_{j=1}^N \int_{x_{j-1}}^{x_j} |M(x_i,\xi)| \frac{1}{\varepsilon} e^{-\frac{\sqrt{\alpha}\xi}{\sqrt{\varepsilon}}} d\xi = Ch^2 \sum_{j=1}^N \int_{x_{j-1}}^{x_j} \left| M(x_i,0) + \frac{\partial M}{\partial \xi} \xi \right| \frac{1}{\varepsilon} e^{-\frac{\sqrt{\alpha}\xi}{\sqrt{\varepsilon}}}
$$

$$
\leq Ch^2 \int_0^l \frac{\xi}{\varepsilon} e^{-\frac{\sqrt{\alpha}\xi}{\sqrt{\varepsilon}}} d\xi,
$$

from which after taking into consideration $xe^{-x} \leq e^{-\frac{x}{2}}$, we obtain

$$
Ch^2 \sum_{j=1}^N \int_{x_{j-1}}^{x_j} |M(x_i,\xi)| \frac{1}{\varepsilon} e^{-\frac{\sqrt{\alpha}\xi}{\sqrt{\varepsilon}}} d\xi \le Ch^2 \frac{1}{\sqrt{\alpha}} \int_0^l \frac{1}{\sqrt{\varepsilon}} e^{-\frac{\sqrt{\alpha}\xi}{2\sqrt{\varepsilon}}} d\xi
$$

=
$$
Ch^2 \frac{2}{\alpha} \left(1 - e^{-\frac{\sqrt{\alpha}l}{\sqrt{\varepsilon}}}\right) \le Ch^2.
$$

Analogously, after using the relation

$$
M(x_i, \xi) = M(x_i, l) + \frac{\partial \overline{M}}{\partial \xi} (\xi - l), \qquad M(x, l) = 0,
$$

it is not difficult to confirm that

$$
Ch^2\sum_{j=1}^N\int\limits_{x_{j-1}}^{x_j}\left|M(x_i,\xi)\right|\frac{1}{\varepsilon}e^{-\frac{\sqrt{\alpha}(l-\xi)}{\sqrt{\varepsilon}}}d\xi\leq Ch^2.
$$

Therefore, we obtain

$$
\frac{1}{2} |\lambda| \sum_{j=1}^{N} \int_{x_{j-1}}^{x_j} (x_j - \xi) (\xi - x_{j-1}) |M(x_i, \xi)| |v''(\xi)| d\xi \le Ch^2.
$$
 (29)

Thus, it can be easily seen that the first term in the right side of (26) is that $O(h^2)$. In addition, after taking into account (28) and (29) we obtain

$$
\left| R_i^{(3)} \right| \le C h^2. \tag{30}
$$

From (20) , (25) and (30) , we have

$$
|R_i| \le Ch^2. \tag{31}
$$

The bound (19) together with (31) finish the proof. \Box

4. Numerical Calculates

In this section, theoretical calculates are tested on one sample. Our particular example is

$$
Lv := -\varepsilon v''(x) + (2 - \cos^2(\pi x)) v(x) + \frac{1}{2} \int_{0}^{1} \left(e^{x \sin(\pi \zeta)} - 1 \right) v(\zeta) d\zeta = (1+x)^{-1},
$$

$$
(0 < x < 1),
$$

 $v(0) = 1, \quad v(1) = 0.$

The exact solution to this problem is unknown. For this reason, we estimate errors and calculate solutions using the double-mesh method, which compares the obtained solution to a solution computed on a mesh that is twice as fine. We introduce the maximum point-wise errors and the computed ε -uniform maximum point-wise errors as

$$
e^N_\varepsilon=\max_i|y^{\varepsilon,N}_i-\tilde{y}^{\varepsilon,2N}_{2i}|_{\infty,\overline{\omega}_N},\quad e^N=\max_\varepsilon e^N_\varepsilon,
$$

where $\tilde{y}_{2i}^{\varepsilon,2N}$ is the approximate solution of the related method on the mesh

$$
\tilde{\omega}_{2N} = \{x_{\frac{i}{2}} : i = 0, 1, ..., 2N\}, \quad x_{i+\frac{1}{2}} = \frac{x_i + x_{i+1}}{2} \quad \text{for} \quad 0 \le i \le N - 1.
$$

We also describe the computed ε -uniform the rates of convergence and the rates of convergence as follows

$$
p^N_\varepsilon=\frac{\ln\left(\frac{e^N_\varepsilon}{e^{2N}_\varepsilon}\right)}{\ln 2},\quad p^N=\frac{\ln\left(\frac{e^N}{e^{2N}}\right)}{\ln 2}.
$$

The rate of convergence of the difference approximation is significantly in agreement with the theoretical analysis, as shown in the Table 1.

5. Conclusion

In this paper, we described a new second-order difference scheme, which was constructed on the uniform mesh by using composite trapezoidal rule for integral term involving kernel function to solve linear FIDEs with singular perturbation. We tested the technique on one example with various values of ε and N to demonstrate the appropriateness of the method. Numerical investigations can be sustained for more sophisticated types such as partial integro-differential equations, nonlinear, delay form, higher dimensional, etc.

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ON F-COSMALL MORPHISMS

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ABSTRACT. In this paper, we first define the notion of F -cosmall quotients for an additive exact substructure $\mathcal F$ of an exact structure $\mathcal E$ in an additive category A . We show that every F -cosmall quotient is right minimal in some cases. We also give the definition of F-superfluous quotients and we relate it the approximation of modules. As an application, we investigate our results in a pure-exact substructure \mathcal{F} .

1. INTRODUCTION

In [\[12\]](#page-82-1), Ziegler introduced the partial morphisms by using model theory of modules. Then in $\overline{9}$, the partial morphisms was investigated by Monari Martinez in terms of systems of linear equations. But this algebraic definition of partial morphisms was not useful in the categorical studies of purity. Then in $[4]$ Cortés-Izurdiaga, Guil Asensio, Kaleboğaz and Srivastava studied partial morphisms by using category theory. In $\vert \mathbf{4} \vert$, the authors defined partial morphisms by using pushout with respect to an additive exact substructure $\mathcal F$ of an exact structure $\mathcal E$ in an additive category $\mathcal A$ and they call them $\mathcal F$ -partial morphisms. They showed that the definition of $\mathcal F$ -partial morphisms with the pure-exact substructure $\mathcal F$ in the category of right R-modules are coincide with the partial morphisms that defined by Ziegler in $\boxed{12}$. By using F-partial morphisms they also define F-small extension and gave an application of this definition to the pure-exact substructure $\mathcal F$ in the category of right modules over a ring and called it Ziegler small extension. As a dual notion of $\mathcal{F}\text{-partial}$ morphisms, in **6** $\mathcal{F}\text{-}{\text{copartial}}$ morphisms was defined by Kaleboğaz: a morphism $f: X \longrightarrow U$ is F-copartial morphism with respect to a quotient map $p: Y \longrightarrow U$ if and only if $\text{Ext}^1(f, -)$ transforms p in an F-deflation. She studied the properties of $\mathcal F$ -copartial morphisms and investigated the applications of $\mathcal F$ -copartial morphisms to some exact substructures of $\mathcal E$ in the category of right R-modules.

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In this paper, by using $\mathcal{F}\text{-}{\rm copartial}$ morphisms, we first introduce $\mathcal{F}\text{-}{\rm cosmall}$ quotients for any additive exact substructure F of an exact structure $\mathcal E$ in an additive category A (see Definition $\boxed{2}$). We also give a new characterization of F-cosmall quotients (see Proposition $\boxed{1}$). As an application to a pure-exact structure \mathcal{F} , we give the definition of pure-cosmall quotients and we say that pure-cosmall quotients are dual of Ziegler small extensions.

A morphism $p : M \longrightarrow N$ is called *right minimal* if any endomorphism $q : M \longrightarrow$ M with $pg = p$ is an isomorphism (see $[1]$, page 6]). In $[8]$, right minimal morphisms are studied by Keskin Tütüncü. In $\boxed{8}$ the author dualizes some results in $\boxed{3}$ and gets several useful results by investigating the relationship between $\text{End}_R(N)$ and End_R (M) when there is a right minimal epimorphism $p : M \longrightarrow N$. The author also proves that there is an isomorphism between two rings $END_R^M(N)/J(END_R^M(N))$ and $END_R^N(M)/J(END_R^N(M))$ if there exists a right minimal epimorphism p : $M \longrightarrow N$ in $\overline{8}$, Theorem 2.6 (1). As a consequence of this result the structure of the endomorphism ring of a quasi-projective module and an automorphismcoinvariant module are explained. One of the main purposes of this paper is to give an example of right minimal morphisms. In Theorem \prod , we prove that every F-cosmall quotient $f : P \longrightarrow M$ with P an F-projective object (projective objects with respect to $\mathcal{F}\text{-}deflations$ is right minimal. An application of this theorem to the pure-exact structure gives us the dual version of \mathbb{S} . Proposition 1.6. Moreover, we give the definitions of $\mathcal F$ -superfluous quotient and weakly $\mathcal F$ -superfluous quotient (see Definition $\overline{5}$). Then we investigate the relation between $\mathcal{F}\text{-}$ cosmall quotient and \mathcal{F} -superfluous quotient (see Proposition [2\)](#page-79-1). And finally we relate to the existence of approximations of modules. In Theorem $\overline{2}$, we show that a weakly *F*-superfluous quotient $p: Y \longrightarrow U$ with F-projective Y is an F-Proj-cover when F-Proj is the class of F-projective objects of A.

2. RESULTS

Let A be an additive category and (i, p) be a pair of composable morphisms in A:

$$
A \xrightarrow{i} B \xrightarrow{p} C
$$

If i is a kernel of p and p is a cokernel of i then (i, p) is called kernel-cokernel pair in A. Let $\mathcal E$ be the class of kernel-cokernel pairs on A. i is called an *admissible* monomorphism if there exists a morphism p such that $(i, p) \in \mathcal{E}$. Similarly, p is called an *admissible epimorphism* if there exists a morphism i such that $(i, p) \in \mathcal{E}$.

The class of kernel-cokernel pairs $\mathcal E$ is said to be an exact structure on $\mathcal A$ if it is closed under isomorphisms and satisfies the following conditions;

- [E0] For every object $A \in \mathcal{A}$, the identity morphism 1_A is an admissible monomorphism.
- [E0^{op}] For every object $A \in \mathcal{A}$, the identity morphism 1_A is an admissible epimorphism.
	- [E1] The classes of admissible monomorphisms are closed under compositions.

 $[E1^{op}]$ The classes of admissible epimorphisms are closed under compositions. [E2] The pushout of an admissible monomorphism along an arbitrary morphism exists and yields an admissible monomorphism, that is, for any admissible \mathbf{v} morphism f \cdot / is a

monomorphism
$$
i: A \longrightarrow B
$$
 and any morphism $f: A \longrightarrow B'$, there
pushout diagram;

$$
A \xrightarrow{i} B
$$

f

$$
B' \xrightarrow{i'} P
$$

with i' an admissible monomorphism.

 $[E2^{op}]$ The pullback of an admissible epimorphism along an arbitrary morphism exists and yields an admissible epimorphism, that is, for any admissible epimorphism $p : B \longrightarrow C$ and any morphism $g : B' \longrightarrow C$ there is a pullback diagram;

$$
Q \xrightarrow{p'} B'
$$

\n
$$
g' \downarrow g
$$

\n
$$
B \xrightarrow{p} C
$$

with p' an admissible epimorphism.

An exact category is a pair (A, \mathcal{E}) with an additive category A and an exact structure $\mathcal E$ on $\mathcal A$. Elements of $\mathcal E$ are called *short exact sequences*. Keller $\overline{7}$ uses conflation, inflation and deflation for what we call short exact sequence, admissible monomorphism and admissible epimorphism, respectively. Throughout the paper we also use this terminology. Let A be an object of A . An *admissible quotient* of A is a quotient object U of an object A such that one (and any) quotient map $p: A \longrightarrow U$ is a deflation.

An exact substructure $\mathcal F$ of $\mathcal E$ is just an exact structure on $\mathcal A$ such that each conflation in $\mathcal F$ (that we shall call $\mathcal F$ -conflation) is also a conflation in $\mathcal E$. Inflations, deflations and admissible quotient objects with respect to $\mathcal F$ will be called $\mathcal{F}\text{-}$ inflations, $\mathcal{F}\text{-}$ deflations and $\mathcal{F}\text{-}$ demissible quotient objects, respectively.

We shall start with giving the definition of $\mathcal{F}\text{-}{\rm copartial}$ morphisms (respectively, F-copartial isomorphisms) for an additive substructure $\mathcal F$ of an exact structure $\mathcal E$ in an additive category \mathcal{A} . F-copartial morphisms first introduced and investigated in $\overline{6}$ by Kaleboğaz as the dual notion of \mathcal{F} -partial morphism that are studied in $\overline{4}$.

For the rest of the paper, we fix an exact category of (A, \mathcal{E}) and an additive exact substructure $\mathcal F$ of $\mathcal E$.

Definition 1. Let X, Y be objects of A and U an admissible quotient of Y with the quotient map $p: Y \longrightarrow U$.

Let $f: X \longrightarrow U$ be a morphism and consider the pullback of f along the quotient map p:

Then:

- (1) f is called an F-copartial morphism from X to Y with codomain U if \bar{p} is an F-deflation.
- (2) f is called an F-copartial isomorphism from X to Y with codomain U if both \bar{p} and \bar{f} are *F*-deflations.

Now we recall two lemmas from $[6]$, without proofs, that we will use in the rest of the paper. The first lemma given below is a special case of the dual of Obscure Axiom in $\boxed{2}$, Proposition 2.16] (see $\boxed{6}$, Proposition 2.3]). The other one is one of the main properties of F-copartial morphisms (see $[6]$, Proposition 2.5(1)]).

Lemma 1. Let X, Y, Z be objects of A. If an F-deflation $f: Z \longrightarrow Y$ factors through an deflation $p : X \longrightarrow Y$ as follows;

then p is an F -deflation too.

Lemma 2. Let X , Y be objects of A and U , an admissible quotient of Y with the quotient morphism $p: Y \longrightarrow U$. Suppose that p is an F-deflation. A morphism $f: X \longrightarrow U$ is an F-deflation if and only if f is an F-copartial isomorphism from X to Y with codomain U.

As a consequence of this lemma, we can give the following corollary:

Corollary 1. Let Y be an object of A and $g: Z \longrightarrow Y$ be any morphism with any object Z in A. g is an $\mathcal{F}\text{-}definition$ if and only if g is an $\mathcal{F}\text{-}copartial$ isomorphism from Z to Y with codomain Y .

Proof. Let us take the pullback of g along 1_Y . Since 1_Y is an F-deflation, g is an F-deflation if and only if g is an F-copartial isomorphism from Z to Y with codomain Y by Lemma 2 . □

One of the aims of this paper is to give an example of right minimal morphisms. To attain our goal we shall first give the definition of $\mathcal{F}\text{-} \text{cosmall}$ quotient morphisms. These morphisms are dual of \mathcal{F} -small extensions that are defined in \mathbb{F}_4 , Definition 3.4].

Definition 2. Let the object Y of $\mathcal A$ be an admissible quotient of any object X with the quotient map $p': X \longrightarrow Y, U$ be an admissible quotient of X and $p: Y \longrightarrow U$ be a deflation.

(1) We shall say that Y is F-cosmall in U over X if for any F-copartial morphism $g: Z \longrightarrow Y$ from any object Z to X with codomain Y, the following holds:

pg is an $\mathcal F$ -copartial isomorphism from Z to X with codomain U implies that g is an $\mathcal F$ -copartial isomorphism from Z to X with codomain Y.

(2) We shall say that Y is F -cosmall in U if Y is F -cosmall in U over Y. Namely, the deflation p' is the identity morphism of Y.

With the notion of $\mathcal{F}\text{-cosmall}$ object which is defined above, now we can define F -cosmall quotient morphisms as in the following:

Definition 3. Let $p: Y \longrightarrow U$ be a deflation. If Y is F-cosmall in U then the deflation $p: Y \longrightarrow U$ is called an F-cosmall quotient.

Namely, if Y is $\mathcal F$ -cosmall in U over Y then p is an $\mathcal F$ -cosmall quotient.

Here we will give a characterization of $\mathcal{F}\text{-}$ cosmall quotient which will be used in the rest of the paper.

Proposition 1. Let $p: Y \longrightarrow U$ be a deflation. p is an F-cosmall quotient if and only if for any morphism $g: Z \longrightarrow Y$ for any object Z such that pg is an F -copartial isomorphism from Z to Y with codomain U, g is an F -deflation.

Proof. Let Z be an object of A and $g: Z \longrightarrow Y$ be a morphism such that pg is an $\mathcal F$ -copartial isomorphism from Z to Y with codomain U. We will show that g is an F-deflation. If we take pullback of g along 1_Y , then we get the following commutative diagram:

Since 1_Y is an F-deflation, h is an F-deflation. Therefore, g is an F-copartial morphism from Z to Y with codomain Y. As p is an $\mathcal F$ -cosmall quotient, q is also an F-copartial isomorphism from Z to Y with codomain Y. Then, by Corollary \prod q is an $\mathcal{F}\text{-definition.}$

For the converse, to show that p is an $\mathcal{F}\text{-}\mathrm{cosmall}$ quotient, let us take an $\mathcal{F}\text{-}$ copartial morphism $q: Z \longrightarrow Y$ from Z to Y with codomain Y such that pq is an F-copartial isomorphism from Z to Y with codomain U. By assumption, g is an *F*-deflation. By Corollary $\boxed{1}$, g is an *F*-copartial isomorphism from Z to Y with codomain Y. Therefore, p is an $\mathcal F$ -cosmall quotient. \Box

Let R be a ring, Y and Z be right R-modules and $f: Y \longrightarrow Z$ be an epimorphism. Recall that, f is called *pure epimorphism* if $\text{Hom}_R(M, f) : Hom_R(M, Y) \longrightarrow$

 $\text{Hom}_R(M, Z)$ is an epimorphism for all finitely presented right R-modules M. Let X be the kernel of f with the inclusion $u : X \longrightarrow Y$. Then by the theorem of Fieldhouse $\overline{5}$ and Warfield $\overline{10}$, f is pure epimorphism if and only if X is pure in Y (u is a pure monomorphism) in the sense that the natural homomorphism $X \otimes_R N \longrightarrow Y \otimes_R N$ derived from the inclusion map $u : X \longrightarrow Y$ is a monomorphism for all left R-modules N. Then, the conflation $X \longrightarrow Y \longrightarrow Z$ is said to be a *pure conflation* if f is a pure epimorphism (or u is a pure monomorphism). The class of all pure conflations is exact substructure of exact structure of the class of all conflations from $\boxed{2}$, Exercise 5.6. F-copartial morphisms (respectively, F-copartial isomorphisms) with respect to a pure-exact substructure $\mathcal F$ in the category of right R-modules are called copartial morphisms (respectively, copartial isomorphisms), (see $\left[\vec{6}\right]$). Here we will define pure-cosmall quotient morphisms as an application of F-cosmall quotient with respect to a pure-exact substructure $\mathcal F$ in the category of right R-modules.

Definition 4. Let Y and U be right R-modules. An epimorphism $p: Y \longrightarrow U$ is called a *pure-cosmall quotient* if Y is pure-cosmall in U , that means, for any right R-module Z, any copartial morphism $g: Z \longrightarrow Y$ from Z to Y with codomain Y, the following holds:

If pg is a copartial isomorphism from Z to Y with codomain U then g is a copartial isomorphism from Z to Y with codomain Y .

Corollary 2. Let Y and U be right R-modules, $p: Y \longrightarrow U$ be a deflation. p is a pure-cosmall quotient if and only if for any right R -module Z , any morphism $g: Z \longrightarrow Y$ such that pq is a copartial isomorphism from Z to Y with codomain U is a pure epimorphism.

Pure-cosmall quotients are the dual of Ziegler small extensions that are introduced in $\boxed{4}$ and are studied in $\boxed{3}$. In $\boxed{3}$, the authors proved that every Ziegler small extension $u : M \longrightarrow E$ with E being pure-injective is a left minimal morphism. Now we proceed to extend dual of this result to any exact substructure \mathcal{F} . We will show that $\mathcal{F}\text{-} \text{cosmall}$ quotient morphisms are right minimal under a condition. So the following theorem gives us an example of right minimal morphisms.

Let P be an object of A and $p: Y \longrightarrow Z$ be a deflation. Recall that, P is said to be p-projective (or projective with respect to p) if for each morphism $f: P \longrightarrow Z$ there exist a morphism $g : P \longrightarrow Y$ with $pg = f$. P is said to be a projective object in A if it is projective with respect to each deflation. Projective objects with respect to F -deflations will be called $\mathcal{F}\text{-projective objects.}$

Theorem 1. Every F-cosmall quotient $f : P \longrightarrow M$ with P being an F-projective object is right minimal.

Proof. Let $g: P \longrightarrow P$ be a morphism such that $fg = f$. Now we will show that g is an isomorphism. If we consider the pullback of f along fg we get the following commutative diagram;

Since $fg = f$, the identity map 1_P satisfies that $fg1_P = f1_P$. Then by the universal property of pullback, there exist $\alpha : P \longrightarrow Q$ such that $h_1 \alpha = 1_P$ and $h_2 \alpha = 1_P$. By Lemma \prod , h_1 and h_2 are both *F*-deflations. Therefore, fg is an *F*-copartial isomorphism from P to P with codomain M . Since f is an \mathcal{F} -cosmall quotient, g is an $\mathcal{F}\text{-deflation}$ by Proposition $\boxed{1}$. So it is an epimorphism.

Now, using that P is an F-projective, we get that there exists $h : P \longrightarrow P$ such that $gh = 1_P$. Then $f = f1_P = fgh = fh$. By using the previous argument we conclude that h is an epimorphism. Then as $hgh = h = 1_p h$, we get that $hg = 1_p$. Therefore, g is a monomorphism. So g is an isomorphism.

Corollary 3. Every pure-cosmall quotient $f : P \longrightarrow M$ with P being a pureprojective right R-module is right minimal.

Now we will give the definition of $\mathcal{F}\text{-superfluous}$ and weakly $\mathcal{F}\text{-superfluous}$ tients.

Definition 5. Let X and Y be objects of A .

(1) An F-superfluous quotient is an F-deflation $p: X \longrightarrow Y$ such that for any object of Z in A and any morphism $\alpha: Z \longrightarrow X$ the following holds:

 $p\alpha$ is an ${\mathcal F}\text{-deflation}$ implies that α is an ${\mathcal F}\text{-deflation}.$

(2) A weakly F-superfluous quotient is an F-deflation $p: X \longrightarrow Y$ such that for any object of Z in A and any morphism $\alpha: Z \longrightarrow X$ the following holds:

 $p\alpha$ is an F-deflation implies that α is a deflation.

- **Remark 1.** (1) If A is the category of right R-modules and \mathcal{E} is the abelian exact structure, then \mathcal{E} -superfluous quotient morphism is coincide with the small epimorphism that is recalled in $[8]$, Example 2.2(2).
	- (2) If A is the category of right R-modules and $\mathcal F$ is the pure-exact structure, then F -superfluous quotient morphism is coincide with the S-superfluous epimorphism for S being the class of finitely presented modules that is introduced in [\[11\]](#page-82-12).

Now we give the relation between F-cosmall quotient and F-superfluous quotient.

Proposition 2. Let $p: Y \longrightarrow U$ be a deflation. p is an F-superfluous quotient if and only if p is an F -deflation and F -cosmall quotient.

Proof. Suppose that p is an $\mathcal{F}\text{-superfluous quotient. So p is an $\mathcal{F}\text{-deflation. Now we}$$ will show that p is an $\mathcal F$ -cosmall quotient. Let us take an object Z and a morphism $g: Z \longrightarrow Y$ such that pg is an *F*-copartial isomorphism from Z to Y with codomain U. Now if we take the pullback of pq along p we get the following commutative diagram:

$$
Q \xrightarrow{\overline{p}} Z
$$
\n
$$
h \downarrow \qquad \qquad \downarrow p g
$$
\n
$$
Y \xrightarrow{p} U
$$

By Lemma $\overline{2}$, pq is an F-deflation. Then q is an F-deflation by the definition of $\mathcal F$ -superfluous quotient. Therefore, by Proposition $\prod_{i=1}^n p_i$ is an $\mathcal F$ -cosmall quotient.

For the converse, assume that p is an $\mathcal F$ -deflation and $\mathcal F$ -cosmall quotient. To show that p is an F-superfluous quotient let us take a morphism $\alpha: Z \longrightarrow Y$ such that $p\alpha$ is an F-deflation. Now take the pullback of $p\alpha$ along p we get the following commutative diagram:

By Lemma $\overline{2}$, $p\alpha$ is an *F*-copartial isomorphism from *Z* to *Y* with codomain U. Since p is an F-cosmall quotient, α is an F-deflation. Therefore p is an Fsuperfluous quotient. \Box

Let A be any category and X be a class of objects in A . Recall that, a morphism $\phi: X \longrightarrow Y$ in A is a X-precover of Y if $X \in \mathcal{X}$ and for any morphism $f: Z \longrightarrow Y$ with $Z \in \mathcal{X}$, there is a morphism $g: Z \longrightarrow X$ such that $\phi g = f$. A X-precover $\phi: X \longrightarrow Y$ is said to be a X-cover if every morphism $g: X \longrightarrow X$ such that $\phi q = \phi$ is an isomorphism. It is clear that, an X-cover is an X-precover which is a right minimal morphism.

In the next result we will show that, under certain circumstances, a weakly F-superfluous quotient $p: Y \longrightarrow U$ with Y being F-projective is actually an F-Proj-cover for $\mathcal{F}\text{-Proj}$ being the class of $\mathcal{F}\text{-projective}$ objects of \mathcal{A} .

Theorem 2. Let $p: Y \longrightarrow U$ be a deflation. Consider the following assertions:

- (1) p is an F -superfluous quotient and Y is an F -projective object.
- (2) p is an $\mathcal{F}\text{-}definition$, Y is an $\mathcal{F}\text{-}projective$ and p is an $\mathcal{F}\text{-}cosmall$ quotient.
- (3) p is an $\mathcal{F}\text{-}definition, Y$ is an $\mathcal{F}\text{-}projective$ and for any object X, each morphism $f: X \longrightarrow Y$ satisfying that pf is an F-deflation, is a split epimorphism.
- (4) p is an F -Proj-cover for F -Proj being the class of F -projective objects of A.
- (5) p is a weakly $\mathcal{F}\text{-superfluous quotient with }Y$ being $\mathcal{F}\text{-projective object.}$

We have $(1) \Leftrightarrow (2) \Leftrightarrow (3), (2) \Rightarrow (4), (1) \Rightarrow (5)$.

If there exists an F-deflation α : $P \longrightarrow U$ with P being an F-projective object then $(4) \Rightarrow (3)$.

If there exists an F-superfluous quotient α : $P \longrightarrow U$ with P being an Fprojective object then $(5) \Rightarrow (1)$.

Proof. (1) \Leftrightarrow (2) Obvious from Proposition [2.](#page-79-1)

 $(1) \Rightarrow (3)$ Let $f : X \longrightarrow Y$ be a morphism with pf being an F-deflation. Since p is an $\mathcal F$ -superfluous quotient, f is an $\mathcal F$ -deflation. As Y is an $\mathcal F$ -projective module, f is a split epimorphism.

 $(3) \Rightarrow (1)$ It is clear since split epimorphisms are *F*-deflations.

 $(2) \Rightarrow (4)$ Since p is an F-deflation, it is an F-Proj-precover for F-Proj being the class of F-projective objects of A. As p is an F-cosmall quotient, p is right minimal by Theorem $\boxed{1}$. Therefore, p is an F-Proj-cover for F-Proj being the class of $\mathcal F$ -projective objects of $\mathcal A$.

 $(1) \Rightarrow (5)$ It is clear, since every *F*-superfluous quotient is weakly *F*-superfluous quotient.

(4) \Rightarrow (3) Assume that there exists an F-deflation α : P \rightarrow U with P being an F-projective object. Since p is an F-Proj-precover for F-Proj being the class of F-projective objects of A, there exists $g : P \longrightarrow Y$ such that $pg = \alpha$. Since α is an F-deflation, then p is also an F-deflation by Lemma $\overline{\Pi}$. Now let $f : X \longrightarrow Y$ be a morphism such that pf is an $\mathcal F$ -deflation. Since Y is an $\mathcal F$ -projective object then there exists $h: Y \longrightarrow X$ such that $pfh = p$. As p is an F-Proj-cover then fh is an isomorphism. Therefore, f is split.

 $(5) \Rightarrow (1)$ There exists an *F*-superfluous quotient $\alpha : P \longrightarrow U$ with P is an F-projective object. Since Y is F-projective, there exists a morphism $w: Y \longrightarrow P$ such that $\alpha w = p$. Since p is an F-deflation and α is an F-superfluous then w is an F-deflation. And α is an F-deflation too by Lemma \Box . As P is an Fprojective object then there exists $h : P \longrightarrow Y$ such that $wh = 1_P$. So w is an epimorphism. We get $ph = \alpha uh = \alpha 1_P = \alpha$. Then h is an *F*-deflation as p is a weakly F-superfluous. Then $hwh = h1_P = 1ph$. Since h is epic $hw = 1_P$. So w is a monomorphism. Therefore, w is an isomorphism. By $\alpha w = p$ and α is an $\mathcal F$ -superfluous quotient then p is an $\mathcal F$ -superfluous quotient. \Box

Remark 2. Let $p: Y \longrightarrow U$ be a deflation with Y an *F*-projective object of A. From Theorem $2(4) \Rightarrow (2)$, we can say that if p is an F-Proj-cover of U for F-Proj being the class of $\mathcal F$ -projective objects of $\mathcal A$, then p is an $\mathcal F$ -cosmall quotient. But Theorem \prod shows that p can be an F-cosmall quotient map which is not an F-Projcover (since here p need not be an $\mathcal{F}\text{-deflation}$). But p is always right minimal.

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ON THE ZEROS OF R-BONACCI POLYNOMIALS AND THEIR DERIVATIVES

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ABSTRACT. The purpose of the present paper is to examine the zeros of R -Bonacci polynomials and their derivatives. We obtain new characterizations for the zeros of these polynomials. Our results generalize the ones obtained for the special case $r = 2$. Furthermore, we find explicit formulas of the roots of derivatives of R-Bonacci polynomials in some special cases. Our formulas are substantially simple and useful.

1. INTRODUCTION

The problem finding a convenient method to determine the zeros of a polynomial has a long history that dates back to the work of Cauchy [\[14\]](#page-96-1). Zeros of polynomials, which can be real or complex conjugate, have been perhaps among the most popular topics of study for centuries. When the historical development of polynomial studies have been examined, in 2000 BC, the ancient Babylon Tribe living in Mesopotamia stands out. This tribe knowing how to calculate positive roots is perhaps the best example. Some recent applications of the theory of polynomials with symmetric zeros can be found in $[21]$. This is a short review on the polynomials whose zeros are symmetric either to the real line or to the unit circle. These kind polynomials are very important in mathematics and physics (for more details see [\[21\]](#page-97-0) and the references therein). On the other hand, the open problem of determining the exact number of zeros of a given polynomial on the unit circle was studied in [\[22\]](#page-97-1). Several classes of polynomials with symmetric zeros are also discussed in detail.

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Fibonacci polynomials, a broad class of polynomials, were first described by Belgian mathematician Eugene Charles Catalan (1814-1894), German mathematician E. Jacobsthal and Lucas polynomials in 1970 by M. Bicknell. The starting point of this polynomial class is based on well-known Golden Ratio and Fibonacci numbers, which are still of great interest in the world of modern applied sciences and whose new applications are still found (see, for instance, $\boxed{1}$ - $\boxed{16}$ and $\boxed{18}$ - $\boxed{20}$). For any positive real number x , the Fibonacci polynomials are defined by

$$
F_{n+2}(x) = x F_{n+1}(x) + F_n(x),
$$

with initial values $F_0(x) = 0$, $F_1(x) = 1$. In [\[10\]](#page-96-3), V. E. Hoggat and M. Bicknell are found explicitly the zeros of these polynomials using hyperbolic trigonometric functions. The symmetric polynomials of the zeros of Fibonacci polynomials were found by M. X. He, D. Simon and P. E. Ricci in $\overline{7}$. Furthermore, in $\overline{8}$, the location and distribution of the zeros of the Fibonacci polynomials were determined. Fibonacci polynomials and their different properties have been examined (see, for example, $\begin{bmatrix} 3 \end{bmatrix}$, $\begin{bmatrix} 24 \end{bmatrix}$, $\begin{bmatrix} 25 \end{bmatrix}$, and the references therein).

In this paper our aim is to examine the zeros of R-Bonacci polynomials and their derivatives. R-Bonacci polynomials $R_n(x)$ are defined by the following recursive equation in $\boxed{9}$ for any integer n and $r \geq 2$:

$$
R_{n+r}(x) = x^{r-1} R_{n+r-1}(x) + x^{r-2} R_{n+r-2}(x) + \dots + R_n(x), \qquad (1)
$$

with the initial values $R_{-k}(x) = 0, k = 0, 1, \dots, r-2, R_1(x) = 1$. For $r = 2, 3$ in the recurrence relation (Π) , R-Bonacci polynomials become the so called Fibonacci and Tribonacci polynomials, respectively. Although, there are a large number of publications regarding to Fibonacci polynomials and their generalizations (see $\sqrt{7}$ -[\[9\]](#page-96-7), [\[11\]](#page-96-8) and [\[13\]](#page-96-9)), the open expressions have not been found for the zeros of Tribonacci polynomials and their derivatives yet. Instead, numerical studies have been done more intensively in recent years. Zero attractors of these polynomials were obtained by W. Goh, M. X. He and P. E. Ricci in $[6]$. In $[15]$, the number of the real roots of Tribonacci-coefficient polynomials were found. Recently, the smallest disc or annulus containing the zeros of Tribonacci polynomials have been examined by O. Oztunç Kaymak and an algorithm has given to use in other boundary problems in [\[12\]](#page-96-11).

In this study, in order to determine the distribution of the zeros of R-Bonacci polynomials, we examine some properties of R-bonacci polynomials, a more general class of Fibonacci and Tribonacci polynomials. In Section $\boxed{2}$, we consider some classes of R-Bonacci polynomials. We find the symmetric polynomials which are made up of the r^{th} order of the zeros of R-Bonacci polynomials. Using these symmetric polynomials, we determine the reference roots for the polynomials $R_{rn+p}(x)$ for $p = 0, 1$ and $n = 1$. So, we have generalized the results obtained for the special case $r = 2$ in [\[10\]](#page-96-3).

On the other hand, there are several papers on the derivatives of the Fibonacci polynomials (see $\left[4\right]$, $\left[5\right]$, $\left[17\right]$, $\left[23\right]$ and the references therein). In Section $\overline{3}$, we study the roots of the derivatives of R-Bonacci polynomials. We obtain the most general symmetric polynomials which are made up of the r^{th} order of the zeros of derivatives of R-Bonacci polynomials. Using these symmetric polynomials, we find some formulas for the zeros of derivatives of R-Bonacci polynomials for some special values of t.

2. Zeros of Some Classes of R-Bonacci Polynomials

The general representations for R-Bonacci polynomials was given in $\boxed{9}$ as

$$
R_n(x) = \sum_{j=0}^{\left[\frac{(r-1)(n-1)}{r}\right]} \binom{n-j-1}{j}_r x^{(r-1)(n-1)-rj}.
$$
 (2)

Here $r_{n,j} = \binom{n}{j}_r$ denotes the *r*-nomial coefficient and [.] denotes the greatest integer function. In this section, we obtain the symmetric polynomials including the zeros of R-Bonacci polynomials. Before finding symmetric polynomial of the zeros of R-Bonacci polynomials, the following observation based on $2:$

Observation 1. The zeros of $R_n(x)$ and $R_n(xe^{\frac{2\pi}{r}i})$ are identical.

To see the above observation, the following result is obtained by writing $xe^{\frac{2\pi}{r}i}$ instead of x in 2 :

$$
R_n(xe^{\frac{2\pi}{r}i}) = \sum_{j=0}^{\left[\frac{(r-1)(n-1)}{r}\right]} r_{n,j} \left(xe^{\frac{2\pi i}{r}}\right)^{(r-1)(n-1)-rj}.
$$
 (3)

Then, the desired result is easily seen by taking a parenthesis $\left(e^{\frac{2\pi i}{r}}\right)^{(r-1)(n-1)}$ and we have

$$
R_n(xe^{\frac{2\pi}{r}i}) = \left(e^{\frac{2\pi i}{r}}\right)^{(r-1)(n-1)} \left(\begin{array}{c}r_{n,0} \ x^{(r-1)(n-1)} + r_{n,1} \ x^{(r-1)(n-1)-r} \\ + \cdots + r_{n,\left[\frac{(r-1)(n-1)}{r}\right]} \ x\end{array}\right)
$$

= $\left(e^{\frac{2\pi i}{r}}\right)^{(r-1)(n-1)} R_n(x).$

By this observation, we can simply state that the zeros of R -Bonacci polynomials can be created by rotating the angle of $\frac{2\pi}{r}$ degrees in the complex plane. The zeros of $R_n(x)$ are same as $R_n(xe^{\frac{2\pi i}{r}})$, as they are with $R_n(xe^{-\frac{2\pi i}{r}})$. Thus, the zeros of $R_n(x)$ can be divided into r sets: $\{x_i\}$, $\{x_i e^{\frac{2\pi i}{r}}\}$, \cdots , $\{x_i e^{\frac{2\pi i}{r}}\}$. Here we refer to this set $\{x_i\}$ as a set of reference zeros. The zeros of the 20^{th} Tribonacci polynomial are seen in Figure $\boxed{1}$. Notice that the zeros of this polynomial can be generated at an angle of 120 degrees with reference to the set $\{x_i\}$.

FIGURE 1. The zeros of $T_{20}(x)$

Our theorems are coincide with the ones obtained in $\overline{7}$ for $R = 2, 3$. Actually, Theorem $\boxed{1}$ and Theorem $\boxed{2}$ are the most generalized versions of the results obtained for Tribonacci and Fibonacci polynomials. For the definition of a symmetric polynomial one can see [\[7\]](#page-96-4).

Theorem 1. The most general form of the jth symmetric polynomials consisting of over the r^{th} zeros of $R_{rn}(x)$ is as follows:

$$
\sigma_j \left(x_1^r, \cdots, x_{(r-1)n-1}^r \right) = (-1)^j {rn - j - 1 \choose j}.
$$
 (4)

.

Proof. It is known that the zeros of R-Bonacci polynomials lie in the argument $\frac{2\pi}{r}$ and hence the polynomial $R_{rn}(x)$ can be factorized as

$$
R_{rn}(x) = x \prod_{k=1}^{(r-1)n-1} (x - x_k) \left(x - x_k e^{\frac{2\pi i}{r}} \right) \cdots \left(x - x_k e^{-\frac{2\pi i}{r}} \right)
$$

If we rearrange this equation, we obtain

$$
R_{rn}(x) = x\{x^{r^{2}n-rn-r} - (r-1)n-1
$$

$$
x^{r^{2}n-rn-2r} \sum_{k=1}^{(r-1)n-1} x_{k}^{r} + x^{r^{2}n-rn-3r} \sum_{j \neq k} x_{j}^{r} x_{k}^{r}
$$

$$
-x^{r^{2}n-rn-4r} \sum_{j \neq k \neq l} x_{j}^{r} x_{k}^{r} x_{l}^{r} + \cdots - \prod_{k=1}^{(r-1)n-1} x_{k}^{r} \}
$$

=
$$
\left\{ \sum_{j=0}^{(r-1)n-1} (-1)^{j} x^{(r-1)(rn-1)-rj} \left\{ \sum_{1=l_{1} < l_{2} < \cdots < l_{j} i=1}^{j} x_{l_{i}}^{r} \right\} \right\}
$$

=
$$
\sum_{j=0}^{(r-1)n-1} (-1)^{j} \sigma_{j} \left(x_{1}^{r}, x_{2}^{r}, \cdots, x_{(r-1)n-1}^{r} \right) x^{(r-1)(rn-1)-rj}.
$$
 (5)

On the other hand by (2) we can write

$$
R_{rn}(x) = \sum_{j=0}^{(r-1)n-1} {rn - j - 1 \choose j} x^{(r-1)(rn-1) - rj}.
$$
 (6)

Since the equations $\boxed{5}$ and $\boxed{6}$ are equal, we obtain the desired result $\boxed{4}$. \Box **Corollary 1.** The following equations are satisfied by the zeros of $R_{rn}(x)$:

$$
\sum_{k=1}^{(r-1)n-1} x_k^r = -\binom{rn-2}{1}_r.
$$
 (7)

Proof. By setting $j = 1$ in the equation $\left| \mathbf{A} \right|$ desired result is obtained. \square **Theorem 2.** The most general form of the jth symmetric polynomials consisting of the r^{th} zeros of $R_{rn+1}(x)$ is as follows :

$$
\sigma_j\left(x_1^r, \cdots, x_{(r-1)n}^r\right) = (-1)^j \binom{rn-j}{j}_r.
$$
\n(8)

Proof. By a similar way used in the proof of Theorem $\boxed{1}$, we can write

$$
R_{rn+1}(x) = \prod_{k=1}^{(r-1)n} (x - x_k) \left(x - x_k e^{\frac{2\pi i}{r}} \right) \cdots \left(x - x_k e^{-\frac{2\pi i}{r}} \right).
$$

Then we get

$$
R_{rn+1}(x) = \left\{ x^{r^2n-rn} - x^{r^2n-rn-r} \sum_{k=1}^{(r-1)n} x_k^r + x^{r^2n-rn-2r} \sum_{j \neq k} x_j^r x_k^r \right\}
$$

$$
-x^{r^2n-rn-3r} \sum_{j \neq k \neq l} x_j^r x_k^r x_l^r + \cdots - \prod_{k=1}^{(r-1)n} x_k^r \}
$$

$$
= \left\{ \sum_{j=0}^{(r-1)n} (-1)^j x^{rn(r-1)-rj} \left\{ \sum_{1=l_1 < l_2 < \cdots < l_j i = 1} \prod_{j=1}^{j} x_{l_i}^r \right\} \right\}
$$

$$
= \sum_{j=0}^{(r-1)n} (-1)^j \sigma_j \left(x_1^r, x_2^r, \cdots, x_{(r-1)n}^r \right) x^{rn(r-1)-rj}.
$$
 (9)

By putting $rn + 1$ instead of n in (2) , we find

$$
R_{rn+1}(x) = \sum_{j=0}^{n(r-1)} \binom{rn-j}{j}_r x^{(r-1)rn-rj}.
$$
 (10)

It follows from the comparison (9) and (10) , it is possible to write the desired result (\mathbb{S}) .

Corollary 2. The following equations are satisfied by the zeros of $R_{rn+1}(x)$:

$$
\sum_{k=1}^{(r-1)n} x_k^r = -\binom{rn-1}{1}_r.
$$
\n(11)

Proof. If we set $j = 1$ in the equation $\boxed{\$\}$ then we get the equation $\boxed{\{11\}}$.

Now, using these symmetric polynomials, we obtain the reference roots of $R_{rn+p}(x)$ for $p = 0, 1$.

Theorem 3. For $p = 0, 1$ and $n = 1$, let $x_j (1 \le j \le r)$ be the reference zeros of $R_{rn+p}(x)$. Then we have

$$
x_j^r = -1.\t\t(12)
$$

Proof. Let $p = 0$ or $p = 1$ and let the set of the reference zeros of $R_{rn+p}(x)$ be ${x_1, \dots, x_r}$. The other zeros of the polynomial $R_{rn+p}(x)$ will be generated by the argument $\frac{2\pi}{r}$ except the root $x = 0$. For a fixed j, using the equations (11) and (7) , we have

$$
\sum_{k=1}^{r-1} x_k^r = x_1^r + x_2^r + \dots + x_{r-1}^r
$$

$$
= x_j^r + \left(x_j e^{\frac{2\pi i}{r}}\right)^r + \left(x_j e^{\frac{4\pi i}{r}}\right)^r + \dots + \left(x_j e^{\frac{2(r-2)\pi i}{r}}\right)^r = -(r-1)
$$

and

$$
\sum_{k=1}^{r-2} x_k^r = x_1^r + x_2^r + \dots + x_{r-1}^r
$$

= $x_j^r + \left(x_j e^{\frac{2\pi i}{r}}\right)^r + \left(x_j e^{\frac{4\pi i}{r}}\right)^r + \dots + \left(x_j e^{\frac{2(r-3)\pi i}{r}}\right)^r = -(r-2),$

respectively. Rearranging the above equations, it can be easily seen that the reference roots of $R_{rn+p}(x)$ as in the equation $\sqrt{12}$.

FIGURE 2. The zeros of $B_6(x)$

Example 1. Let us consider the following 5-Bonacci polynomial

$$
B_6(x) = (x^5 + 1)^4.
$$

Using ([12](#page-88-2)), if we solve the equation $x_j^5 = -1(1 \le j \le 5)$, the reference roots of the polynomial $B_6(x)$ are found as follows (see Figure $\boxed{2}$):

$$
x_1 = (-1), x_2 = (-1)^{\frac{1}{5}}, x_3 = -(-1)^{\frac{2}{5}}, x_4 = (-1)^{\frac{3}{5}}, x_5 = -(-1)^{\frac{4}{5}}.
$$

3. Zeros of Derivatives of R-Bonacci Polynomials

Before we find the symmetric polynomials which are made up of the r^{th} order of the zeros of the derivatives of R-Bonacci polynomials $R_n^{(t)}(x)$, we write the algebraic representations of them. For any fixed n, using the equation (2) , the algebraic representation of the derivative polynomial $R_n^{(t)}(x)$ is obtained as follows:

$$
R_n^{(t)}(x) = \left[\frac{\binom{r-1}{r-1}}{\binom{r}{j}}\right]_r (r-1)(n-1) - rj \dots ((r-1)(n-1) - rj - t + 1)x^{(r-1)(n-1) - rj - t}.
$$
\n(13)

Now, we determine the symmetric polynomials for $R_{rn+p}^{(t)}(x)$ for special values of t. We give the following theorem.

Theorem 4. Let
$$
k \in \mathbb{N}^+
$$
, $p \in \{0, 1, \dots, r-1\}$. If we consider
 $t = rk - (1-p)(r-1)$, (14)

$$
\mu = ((r-1)(rn+p-1))\cdots(rn(r-1)-t+(p-1)r+(2-p))
$$
 (15)

and

$$
\eta = (r-1)n - \left(\frac{t + (1-p)(r-1)}{r}\right),\tag{16}
$$

then the most general form of the symmetric polynomials consisting of the zeros of $R_{rn+p}^{(t)}(x)$ is as follows:

$$
\sigma(x_1^r, ..., x_n^r) = \frac{(-1)^j((r-1)(rn+p-1) - rj)...((r-1)(rn+p-1) - rj - t + 1)}{\mu} {rn + p - j - 1 \choose j}_r
$$
\n(17)

Proof. It can be easily seen that

$$
R_{rn+p}^{(t)}(x) = \mu \prod_{k=1}^{\eta} (x - x_k) \left(x - x_k e^{\frac{2\pi i}{r}} \right) \cdots \left(x - x_k e^{-\frac{2\pi i}{r}} \right),
$$

where μ is a constant. Then we have

$$
R_{rn+p}^{(t)}(x) = \mu \{ x^{r^2n-rn-(t+(1-p)(r-1))} - x^{r^2n-rn-(t+(1-p)(r-1))-r} \sum_{k=1}^{\eta} x_k^r + x^{r^2n-rn-(t+(1-p)(r-1))-2r} \sum_{j \neq k} x_j^r x_k^r
$$

$$
-x^{r^2n-rn-(t+(1-p)(r-1))-3r} \sum_{j \neq k \neq l} x_j^r x_k^r x_l^r + \cdots - \prod_{k=1}^{\eta} x_k^r \}
$$

$$
= \mu \left\{ \sum_{j=0}^{\eta} (-1)^j x^{r^2n-rn-(t+(1-p)(r-1))-rj} \left\{ \sum_{1=l_1 < l_2 < \cdots < l_j i=1} \prod_{j=1}^{\eta} x_{l_i}^r \right\} \right\}
$$

$$
= \mu \sum_{j=0}^{\eta} (-1)^j \sigma_j (x_1^r, x_2^r, \cdots, x_\eta^r) x^{(r-1)(rn+p-1)-rj-t} . \tag{18}
$$

By using the equation (13) and taking $rn + p$ instead of n we can write

$$
R_{rn+p}^{(t)}(x)=\displaystyle\sum_{j=0}^{\left[\frac{(r-1)(rn+p-1)}{rt}\right]} \binom{rn+p-j-1}{j} \times
$$

$$
((r-1)(rn+p-1)-rj)...((r-1)(rn+p-1)-rj-t+1)x^{(r-1)(rn+p-1)-rj-t}.
$$
 (19)

Since the equations (18) and (19) are equal, then the proof follows. \Box

Corollary 3. Let t and η be as in the equations $\begin{bmatrix} 14 \end{bmatrix}$ $\begin{bmatrix} 14 \end{bmatrix}$ $\begin{bmatrix} 14 \end{bmatrix}$ and $\begin{bmatrix} 16 \end{bmatrix}$ $\begin{bmatrix} 16 \end{bmatrix}$ $\begin{bmatrix} 16 \end{bmatrix}$, respectively. For $k \in \mathbb{N}^+$ and $p \in \{0, 1, \cdots, r-1\}$, the following equations are satisfied by the zeros

.

$$
of R_{rn+p}^{(t)}(x):
$$
\n
$$
(i) \prod_{k=1}^{n} x_k^r = \frac{(-1)^{\eta} t (t-1)...(1)}{((r-1)(rn+p-1))...(rn(r-1)-t+(p-1)r+(2-p))} {rn+p-\eta-1 \choose \eta}.
$$
\n
$$
(ii) \sum_{k=1}^{n} x_k^r = \frac{((r-1)(rn+p-1)-r)...((r-1)(rn+p-1)-r-t+1)}{((r-1)(rn+p-1))...(rn(r-1)-t+(p-1)r+(2-p))} {rn+p-2 \choose 1}.
$$
\n(21)

Proof. In the equation \mathbf{A} , if we put $j = \eta$ and $j = 1$ we obtain the desired results, respectively. \Box

$$
v_{\eta} = \frac{(-1)^{\eta}t(t-1)\cdots(1)}{((r-1)(rn+p-1))\cdots(rn(r-1)-t+(p-1)r+(2-p))}\binom{rn+p-\eta-1}{\eta}_r
$$
\nand

$$
\psi_{\eta} = (23)
$$
\n
$$
-\frac{((r-1)(rn+p-1)-r)\cdots((r-1)(rn+p-1)-r-t+1)}{((r-1)(rn+p-1))\cdots(rn(r-1)-t+(p-1)r+(2-p))}\binom{rn+p-2}{1}_r.
$$
\nThen we can give the following theorem.

Theorem 5. For
$$
t = r(r-1)n - 2r - (1-p)(r-1)
$$
, $R_{rn+p}^{(t)}(x)$ has $r\left((r-1)n - \left(\frac{t+(1-p)(r-1)}{r}\right)\right)$ roots and these roots are\n
$$
x_k = \left(\frac{\psi_2 \pm \sqrt{\psi_2^2 - 4\nu_2}}{2}\right)^{\frac{1}{r}} e^{\frac{2k\pi i}{r}}, (k = 0, 1, \dots, r-1),
$$
\n(24)

where v_2 and ψ_2 are defined by the equations $\boxed{3}$ $\boxed{3}$ $\boxed{3}$ and $\boxed{3}$, respectively.

Proof. Since $R_{rn+p}^{(r(r-1)n-2r-(1-p)(r-1))}(x)$ is a polynomial of $r\left((r-1)n-\left(\frac{t+(1-p)(r-1)}{r}\right)\right)$ $\binom{p(r-1)}{r}$ th degree then by using the equations $\binom{3}{3}$ and $\binom{3}{5}$ we have $\overline{2}$

$$
\prod_{k=1}^{2} x_k^r = x_1^r x_2^r = v_2 \tag{25}
$$

Let

and

$$
\sum_{k=1}^{2} x_k^r = x_1^r + x_2^r = \psi_2.
$$
 (26)

Since we know that $x_1^r = \frac{v_2}{x_2^r}$ it can be easily seen that

 $x_2^{2r} - \psi_2 \ x_2^r + \upsilon_2 = 0.$

Solving this last equation of the second degree, the roots can be easily found. So the roots of $R_{rn+p}^{(t)}(x)$ must be as in the equation (24) .

Since we have Fibonacci and Tribonacci polynomials for $r = 2$ and $r = 3$, respectively, we can give the following corollaries.

Corollary 4. Let $p \in \{0,1\}$ and $t = 2n - 5 + p$. The zeros of the polynomial $F_{2n+p}^{(t)}(x)$ can be formulized as follows:

$$
x_k = \left(\frac{\psi_2 \pm \sqrt{\psi_2^2 - 4\upsilon_2}}{2}\right)^{\frac{1}{2}} e^{k\pi i}, (k = 0, 1)
$$

where v_2 and ψ_2 are defined by the equations $\boxed{3}$ $\boxed{3}$ $\boxed{3}$ and $\boxed{3}$, respectively.

In $[23]$, J. Wang proved the following equation for any fixed n

$$
L_n^{(t)}(x) = n F_n^{(t-1)}(x), n \ge 1,
$$
\n(27)

where $L_n(x)$ are Lucas polynomials. Hence the zeros of $L_n^{(t+1)}(x)$ and $F_n^{(t)}(x)$ are identical.

Corollary 5. Let $p \in \{0, 1, 2\}$ and $t = 6n - 8 + 2p$. The zeros of the polynomial $T_{3n+p}^{(t)}(x)$ are

$$
x_k = \left(\frac{\psi_2 \pm \sqrt{\psi_2^2 - 4\nu_2}}{2}\right)^{\frac{1}{3}} e^{\frac{2k\pi i}{3}} (k = 0, 1, 2),\tag{28}
$$

where v_2 and ψ_2 are defined by the equations $\boxed{3}$ $\boxed{3}$ $\boxed{3}$ and $\boxed{3}$, respectively.

Now we give some examples.

Example 2. Consider the zeros of the polynomial

$$
T_6^{(iv)}(x) = 5040x^6 + 3360x^3 + 144.
$$

In the equation ([28](#page-92-0)), writing $\psi_2 = 2/3$, $v_2 = 1/35$, we find the zeros of this polynomial as

$$
x_k = \sqrt[3]{\frac{2/3 \pm \sqrt{(2/3)^2 - 4/35}}{2}} e^{\frac{2k\pi i}{3}}, (k = 0, 1, 2)
$$

(see Figure $\boxed{3}$).

FIGURE 4. The roots of $Q_8^{(13)}(x)$

Using the equations $\boxed{3}$ $\boxed{3}$ $\boxed{3}$ and $\boxed{3}$ we have

$$
\prod_{k=1}^{2} x_k^4 = \frac{1}{13566} = v_2
$$

and

$$
\sum_{k=1}^{2} x_k^4 = -\frac{4}{57} = \psi_2.
$$

Then the roots of $Q_8^{(13)}(x)$ are generated by x_k $(k = 0, 1, 2, 3)$. By (24) (24) (24) , the roots of the polynomial $Q_8^{(13)}(x)$ are obtained as

$$
x_1 = \sqrt[4]{\frac{-\frac{4}{57} + \sqrt{(-\frac{4}{57})^2 - \frac{4}{13566}}}{2}} = 0.127788 + 0.127788i,
$$

and

$$
x_2 = \sqrt[4]{\frac{-\frac{4}{57} - \sqrt{(-\frac{4}{57})^2 - \frac{4}{13566}}}{2}} = 0.36255 + 0.36255i
$$

for $k = 0$,

$$
x_3 = \sqrt[4]{\frac{-\frac{4}{57} + \sqrt{(-\frac{4}{57})^2 - \frac{4}{13566}}}{2}} e^{\frac{\pi i}{2}} = -0.36255 + 0.36255i
$$

and

$$
x_4 = \sqrt[4]{\frac{-\frac{4}{57} - \sqrt{(-\frac{4}{57})^2 - \frac{4}{13566}}}{2}} e^{\frac{\pi i}{2}} = -0.127788 + 0.127788i
$$

for $k = 1$,

$$
x_5 = \sqrt[4]{\frac{-\frac{4}{57} + \sqrt{(-\frac{4}{57})^2 - \frac{4}{13566}}}{2}} e^{\pi i} = -0.127788 - 0.127788i
$$

and

$$
x_6 = \sqrt[4]{\frac{-\frac{4}{57} - \sqrt{(-\frac{4}{57})^2 - \frac{4}{13566}}}{2}} e^{\pi i} = -0.36255 - 0.36255i,
$$

for $k = 2$,

$$
x_7 = \sqrt[4]{\frac{-\frac{4}{57} + \sqrt{(-\frac{4}{57})^2 - \frac{4}{13566}}}{2}} e^{\frac{3\pi i}{2}} = 0.127788 - 0.127788i
$$

FIGURE 5. The roots of $B_8^{(18)}(x)$

and

$$
x_8 = \sqrt[4]{\frac{-\frac{4}{57} - \sqrt{(-\frac{4}{57})^2 - \frac{4}{13566}}}{2}} e^{\frac{3\pi i}{2}} = 0.36255 - 0.36255i,
$$

for $k = 3$ (see Figure 4).

Example 4. Let us consider the 5-Bonacci polynomials $B_8^{18}(x)$. In this case, we have $p = 3$, $n = 1$, $r = 5$ and we obtain

$$
B_8^{18}(x) = 96035605585920000 + 1292600836944248832000x^5
$$

 $+84019054401376174080000x^{10}$.

The roots of this polynomial are found as follows (see Figure $\boxed{5}$):

$$
x_k = \sqrt[5]{\frac{\psi_2 \pm \sqrt{\psi_2^2 - 4\nu_2}}{2}} e^{\frac{2k\pi i}{5}}, k = 0, 1, 2, 3, 4.
$$

4. Conclusion and Future Work

In this paper, in order to obtain new formulas for the zeros of R-Bonacci polynomials and their derivatives, the most general form of the j^{th} symmetric polynomials consisting of over the r^{th} zeros of $R_n(x)$ and $R_{rn+p}^{(t)}(x)$ are given. Using some consequences of these symmetric polynomials, some explicit formulas for the zeros of these polynomials, which have been given in (12) and (24) , are found. Although these formulas are simple, they are valuable because they formulate the zero values of many R-Bonacci polynomials, which is the most general form of the Fibonacci polynomials, and their derivatives.

Given the future studies on this topic, the zeros of the remaining R-Bonacci polynomials can be formulated using different methods. For this reason, it is thought that formulating the zeros of a R-Bonacci polynomial will increase the applicability of this problem in different engineering applications. In addition, this study is also thought to be a guide for formulating the zero locations of polynomials with unknown zero locations. Because this method is applicable for all polynomial classes.

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SET-GENERATED SOFT SUBRINGS OF RINGS

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Abstract. This paper focuses on the set-oriented operations and set-oriented algebraic structures of soft sets. Relatedly, in this paper, firstly some essential properties of α -intersection of soft set are investigated, where α is a non-empty subset of the universal set. Later, by using α -intersection of soft set, the notion of set-generated soft subring of a ring is introduced. The generators of soft intersections and products of soft subrings are given. Some related properties about generators of soft subrings are investigated and illustrated by several examples.

1. Introduction

Since the modeling of uncertain data in medical science, economics, sociology, environmental science, engineering and many other fields is very complex, it is difficult to successfully deal with them by classical methods. In the last century, many approaches that are useful in modeling uncertainties have been proposed. The fuzzy set theory $\left|1\right|2$, the interval mathematics $\left|3\right|$, vague set theory $\left|4\right|$ and rough set theory $\sqrt{5}$ 6 and are favorable approaches to describing uncertain data, but each of these theories has its own difficulties in classifying data parametrically. To fill this gap, Molodtsov [\[7\]](#page-109-7) proposed a completely new approach named soft set theory. This approach allowed the uncertain data frequently encountered in many areas to be classified parametrically, thereby providing a better representation of them. In the years following the budding of soft sets, the theoretical and practical aspects of these sets were discussed. Maji et al. conceptualized the some set operations of soft sets [\[8\]](#page-109-8) and made further efforts to show the implementation of soft sets in

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decision making $\left|\frac{9}{1}\right|$. Ali et al. $\left|\frac{10}{10}\right|$ introduced some new soft set operations such as the restricted difference, the restricted intersection, the extended intersection and the restricted union. Çağman and Enginoğlu $\boxed{11}$ revisited some basic operations of soft sets to make them more efficient in some cases. In $[12]$ - $[14]$ $[14]$, the authors studied the operations of difference and symmetric difference of soft sets. Aygün and Kamacı [\[15\]](#page-109-14) developed some functional operations of soft sets and then demonstrated their efficiency in handling decision making problems. Also, they defined XOR and XNOR products of soft sets and derived new soft algebraic structures by using these soft set products $[16]$. Cagman and Enginoglu $[17]$ introduced the soft matrices representing soft sets and their handy operations to create a soft maxmin decision making procedure which can be successfully applied to the problems containing uncertainties. In $[18-21]$ $[18-21]$, the researchers discussed specific kinds of soft matrices and construct new improved types of soft max-min decision making procedure. Moreover, the inverse types of soft matrices were investigated and their applications to decision making were presented $\sqrt{22}$, Recently, the works on the operations of soft sets and soft matrices are progressing rapidly.

On the other hand, many algebraic structures based on the basic principles and operations of soft sets have been proposed. In 2007, Aktas and Ca \tilde{q} and \tilde{q} introduced the rudiments of soft groups and studied their basic properties. Uluçay et al. [\[25\]](#page-110-5) studied soft representation of soft groups. Feng et al. [\[26\]](#page-110-6) defined the concepts of soft subsemirings, soft semirings, soft semiring homomorphisms, soft ideals and idealistic soft semirings. In , the authors $\sqrt{27}$, $\sqrt{28}$ introduced the fundamentals of soft rings and soft normed rings. In $[29]$, Atagün and Sezgin discussed the algebraic soft substructures of rings and defined soft subring of a ring, soft ideal of a ring, soft submodule of a module and soft subfield of a field. Sezgin et al. [\[30\]](#page-110-10) expanded the study of soft near-rings, especially according to the idealistic soft near-rings. Ostadhadi-Dehkordi and Shum [\[31\]](#page-110-11) investigated regular and strongly regular relations on the soft hyperrings. Tahat et al. [\[32\]](#page-110-12) discussed the characterizations of soft topological soft groups and soft rings. Karaaslan [\[33\]](#page-110-13) investigated some outstanding properties of collection of soft sets over AG-groupoid, AG-band and AG[∗] -groupoid. In [\[34\]](#page-110-14), Yousafzai et al. introduced the notion of soft sets in an ordered AG-groupoid and they studied different type ideals and strongly regular elements. Zhan et al. [\[35\]](#page-110-15) defined some new soft algebraic structures such as (M,N) soft union hemiring and (M,N)-soft union h-ideal, which are generalisations of soft union hemiring and soft union h-ideal to tackle many uncertainty problems. Atagün and Sezgin [\[36\]](#page-110-16) described the notions of soft N-subgroups, soft subnear-rings and soft ideals of near-rings and also derived the product operation and bi-intersection of soft N-groups, soft subnear-rings and soft ideals of near-rings. On the other hand, some authors developed soft topology in various aspects and discussed real life examples [\[37–](#page-110-17)[39\]](#page-110-18).

In [\[40\]](#page-111-0), Sezer et al. argued that the set-oriented approaches based on inclusion of soft set can be extended the range of operations, algebraic structures, topological structures, application aspects of soft sets. Thus, they defined the lower α -inclusion and upper α -inclusion of a soft set over the universal set U, where $\alpha \subseteq U$. Moreover, by using the upper α -inclusion of a soft set, they proposed the idea of upper α -semigroups for the soft sets. In [\[41\]](#page-111-1), the authors made some analyzes with respect to group theory and showed that some subgroups of a group can be achieved easily by means of the notions of upper and lower α -inclusions of soft sets. They also demonstrated that a soft uni-group and a soft int-group can be derived by its lower α -subgroup and upper α -subgroup, respectively. In [\[40,](#page-111-0) [41\]](#page-111-1), the authors focused on the α -oriented subgroup structures of soft sets. However, the α -oriented subring structure of soft sets is a gap in the literature. By filling this gap, both the theoretical aspects and practical aspects of the soft sets will be contributed. Relatedly, this paper aims to introduce soft subrings of a ring generated by the set α and to investigate their fundamental properties.

This paper is organized as follows. Section 2 recalls the rudiments of soft sets. Section 3 presents a detailed theoretical study for the α -intersection of a soft set. Section 4 introduces a new concept namely a soft subring of a ring generated by a set and gives many remarkable properties of this concept. Also, this section includes our main theorems, in which we examine generator sets under operations soft intersection and product. Some theoretical results are illustrated by several examples. Section 5 consists of the conclusions of the paper and the direction for future studies.

2. Preliminaries

In this section, we recall the rudiments of rings, soft sets and soft subrings.

By a ring, we mean an algebraic system $(\Re, +, \cdot)$, where (the multiplication . will be omitted in formulas)

- i) $(\Re, +)$ is a abelian group,
- ii) $(\Re, .)$ is a semi-group,
- iii) $a.(b + c) = ab + ac$ and $(a + b)c = ac + bc$ for all $a, b, c \in \Re$ (i.e., left and right distributive rules hold)

Throughout this paper, \Re denotes a ring and the zero of \Re is symbolized by 0_{\Re} .

A subgroup S of $(\Re, +)$ with $SS \subseteq S$ is named a *subring* of \Re and symbolized by $S < \Re$. Therefore, $S < \Re$ if and only if

- i) $S \subseteq \Re$,
- ii) $0_{\Re} \in S$,
- iii) $a b \in S$ for all $a, b \in S$,
- iv) $ab \in S$ for all $a, b \in S$.

Molodtsov $\boxed{7}$ described the soft set in the following manner:

Let U be a universal set and its power set be $P(U)$, T be a set of parameters and $X \subseteq \mathcal{T}$.

Definition 1. ($\boxed{7}$) A pair (Ψ, \mathcal{X}) (or simply $\Psi_{\mathcal{X}}$) is termed to be a soft set over U, where Ψ is a mapping described by

$$
\Psi: \mathcal{X} \to P(U).
$$

Stated in other words, a soft set over the universal set U can be considered as a parameterized family of the subsets of universal set U. For $t \in \mathcal{X}$, $\Psi(t)$ is the set of t-elements of the soft set (Ψ, \mathcal{X}) , or simplistically the set of t-approximate elements of this soft set. To support this idea, Molodtsov presented various examples (see [\[7\]](#page-109-7)). Indeed, there is a mutual correspondence among soft sets and binary relations as given in $\left[42\right]43$. Namely, let $\mathcal T$ and U be non-empty sets and suppose that σ refers to an arbitral binary relation between an element of $\mathcal T$ and an element of U . A set-valued function $\Psi : \mathcal{T} \to P(U)$ can be described as $\Psi(t) = \{u \in U \mid (t, u) \in \sigma\}$ for all $t \in \mathcal{T}$. Hence, the pair (Ψ, \mathcal{T}) is a soft set over U, which is derived from the relation σ .

Definition 2. ($\boxed{8}$) A soft set (Ψ, \mathcal{X}) over U is termed to be a null soft set symbolized by $\Phi_{\mathcal{X}}$, if for all $t \in \mathcal{X}$, $\Psi(t) = \emptyset$ (null set).

Definition 3. ($\boxed{8}$) A soft set (Ψ, \mathcal{X}) over U is termed to be an absolute soft set, if for all $t \in \mathcal{X}$, $\Psi(t) = U$.

Note that we denote the absolute soft set (Ψ, \mathcal{X}) over U by $\mathcal{U}_{\mathcal{X}}$ throughout this paper.

Definition 4. ($\overline{10}$) The relative complement of a soft set (Ψ, \mathcal{X}) is symbolized by $(\Psi, \mathcal{X})^c$ and is defined as $(\Psi, \mathcal{X})^c = (\Psi^c, \mathcal{X})$, where $\Psi^c : \mathcal{X} \to P(U)$ is a mapping given by $\Psi^c(t) = U \setminus \Psi(t)$ for all $t \in \mathcal{X}$.

Definition 5. ($\overline{[8,10]}$ $\overline{[8,10]}$ $\overline{[8,10]}$) Let (Ψ, \mathcal{X}) and (Υ, \mathcal{Y}) be two soft sets over the universal set U.

- a): The restricted intersection of (Ψ, \mathcal{X}) and (Υ, \mathcal{Y}) is denoted and defined as $(\Psi, \mathcal{X}) \cap (\Upsilon, \mathcal{Y}) = (\Theta, \mathcal{Z})$, where $\mathcal{Z} = \mathcal{X} \cap \mathcal{Y} \neq \emptyset$ and $\Theta(t) = \Psi(t) \cap \Upsilon(t)$ for all $t \in \mathcal{Z}$.
- b): The extended intersection of (Ψ, \mathcal{X}) and (Υ, \mathcal{Y}) is denoted and defined as $(\Psi, \mathcal{X}) \sqcap (\Upsilon, \mathcal{Y}) = (\Theta, \mathcal{Z})$, where $\mathcal{Z} = \mathcal{X} \cup \mathcal{Y}$ and for all $t \in \mathcal{Z}$

$$
\Theta(t) = \left\{ \begin{array}{l} \Psi(t), \text{ if } t \in \mathcal{X} \setminus \mathcal{Y} \\ \Upsilon(t), \text{ if } t \in \mathcal{Y} \setminus \mathcal{X} \\ \Psi(t) \cap \Upsilon(t), \text{ if } t \in \mathcal{X} \cap \mathcal{Y} \end{array} \right.
$$

c): The union intersection of (Ψ, \mathcal{X}) and (Υ, \mathcal{Y}) is denoted and defined as $(\Psi, \mathcal{X}) \cup (\Upsilon, \mathcal{Y}) = (\Theta, \mathcal{Z})$, where $\mathcal{Z} = \mathcal{X} \cap \mathcal{Y} \neq \emptyset$ and $\Theta(t) = \Psi(t) \cup \Psi(t)$ for all $t \in \mathcal{Z}$.

d): The extended union of (Ψ, \mathcal{X}) and (Υ, \mathcal{Y}) is denoted and defined as $(\Psi, \mathcal{X}) \sqcup$ $(\Upsilon, \mathcal{Y}) = (\Theta, \mathcal{Z})$, where $\mathcal{Z} = \mathcal{X} \cup \mathcal{Y}$ and for all $t \in \mathcal{Z}$

$$
\Theta(t) = \begin{cases} \Psi(t), \text{ if } t \in \mathcal{X} \setminus \mathcal{Y} \\ \Upsilon(t), \text{ if } t \in \mathcal{Y} \setminus \mathcal{X} \\ \Psi(t) \cup \Upsilon(t), \text{ if } t \in \mathcal{X} \cap \mathcal{Y} \end{cases}
$$

In 2010, Cağman and Enginoğlu $[11]$ redescribed the approximate function Ψ of soft set (Ψ, \mathcal{X}) from T to $P(U)$ such that $\Psi(t) = \emptyset$ if $t \notin \mathcal{X}$. Thus, they revisited the operations of intersection and union of soft sets as follows:

Definition 6. ($\boxed{11}$) Let $\Psi_{\mathcal{X}}$ and $\Upsilon_{\mathcal{Y}}$ be soft sets over U. Then,

- a): the soft union of $\Psi_{\mathcal{X}}$ and $\Upsilon_{\mathcal{Y}}$, denoted by $\Theta_{\mathcal{Z}} = \Psi_{\mathcal{X}} \widetilde{\cup} \Upsilon_{\mathcal{Y}}$, is defined as $\Theta(t) = \Psi(t) \cup \Upsilon(t)$ for all $t \in \mathcal{T}$.
- b): the soft intersection of $\Psi_{\mathcal{X}}$ and $\Upsilon_{\mathcal{Y}}$, denoted by $\Theta_{\mathcal{Z}} = \Psi_{\mathcal{X}} \cap \Upsilon_{\mathcal{Y}}$, is defined as $\Theta(t) = \Psi(t) \cap \Psi(t)$ for all $t \in \mathcal{T}$.

For more details, it can be reviewed the concepts in $[11]$.

The following definition first introduced the soft substructures of an algebraic structure to the literature.

Definition 7. ($\boxed{29}$) Let S be a subring of \Re and (Ψ, S) be a soft set over \Re . If for all $t, v \in S$,

- s1) $\Psi(t v) \supset \Psi(t) \cap \Psi(v)$,
- s2) $\Psi(tv) \supseteq \Psi(t) \cap \Psi(v)$,

then it is said to be a soft subring of \Re and symbolized by $(\Psi, S) \widetilde{\leq} \Re$ or simplistically $\Psi_S \widetilde{\leq} \Re$.

Proposition 1. ($\boxed{29}$) If $\Psi_S \times \Re$, then $\Psi(0) \supseteq \Psi(t)$ for all $t \in S$.

Theorem 1. ($\boxed{29}$) If $\Psi_{S_1} \widetilde{\leq} \Re$ and $\Upsilon_{S_2} \widetilde{\leq} \Re$, then $\Psi_{S_1} \cap \Upsilon_{S_2} \widetilde{\leq} \Re$.

Definition 8. ($[26]$) Let (Ψ, \mathcal{X}) be soft set over U. Then, the set

 $supp(\Psi, \mathcal{X}) = \{t \in \mathcal{X} \mid \Psi(t) \neq \emptyset\}$

is said to be the support of the soft set (Ψ, \mathcal{X}) . A soft set (Ψ, \mathcal{X}) is called non-null if $supp(\Psi, \mathcal{X}) \neq \emptyset$.

3. SOME ASPECTS ON α -INTERSECTION OF SOFT SETS

In this section, we present some theoretical findings for the α -intersection of soft sets.

Definition 9. ($\overline{[44]}$ $\overline{[44]}$ $\overline{[44]}$) Let (Ψ, \mathcal{X}) be a soft set over U and $\emptyset \neq \alpha \subseteq U$. Then, the subset of $\mathcal X$ given by

 $(\Psi, \mathcal{X})^{\cap \alpha} = \{t \in \mathcal{X} \mid \Psi(t) \cap \alpha \neq \emptyset\}$

is called the α -intersection of (Ψ, \mathcal{X}) .

It seen that if $\alpha = U$ and $\Psi(t) \neq \emptyset$ for all $t \in \mathcal{X}$, then $(\Psi, \mathcal{X})^{\cap U} = \mathcal{X}$.

Proposition 2. Let (Ψ, \mathcal{X}) be a soft set over U and let $\emptyset \neq \alpha \subseteq U$. Then

- i) $(\Psi, \mathcal{X})^{\cap \alpha} \subseteq supp(\Psi, \mathcal{X})$.
- ii) If $\alpha \subseteq \Psi(t)$ for all $t \in \mathcal{X}$, then $(\Psi, \mathcal{X})^{\cap \alpha} = supp(\Psi, \mathcal{X}) = \mathcal{X}$.
- iii) If $\Psi(t) \neq \emptyset$ and $\Psi(t) \subseteq \alpha$ for all $t \in \mathcal{X}$, $(\Psi, \mathcal{X})^{\cap \alpha} = supp(\Psi, \mathcal{X}) = \mathcal{X}$.
- iv) If $(\Psi, \mathcal{X}) = \mathcal{U}_{\mathcal{X}}$, then $(\Psi, \mathcal{X})^{\cap \alpha} = supp(\Psi, \mathcal{X}) = \mathcal{X}$.
- v) If $(\Psi, \mathcal{X}) = \Phi_{\mathcal{X}}$ or $supp(\Psi, \mathcal{X}) = \emptyset$, then $(\Psi, \mathcal{X})^{\cap \alpha} = \emptyset$.

Proof. The proof of (i) is seen from the Definitions \otimes and \otimes . (ii) Since $\alpha \subseteq \Psi(t)$ for all $t \in \mathcal{X}$ and $\emptyset \neq \alpha \subseteq U$, then $\Psi(t) \neq \emptyset$ for all $t \in \mathcal{X}$ and $supp(\Psi, \mathcal{X}) = \mathcal{X}$. Under the assumption $\Psi(t) \cap \alpha \neq \emptyset$ for all $t \in \mathcal{X}$, then $supp(\Psi, \mathcal{X}) \subseteq (\Psi, \mathcal{X})^{\cap \alpha}$. Hence the equality obtained from (*i*). The proof of (iii) is similar to proof of (ii). The rest of the proof is easily seen. \Box

Proposition 3. Let (Ψ, \mathcal{X}) be a soft set over U and $\emptyset \neq \alpha \subseteq U$. Then

- i) If $\alpha \subseteq \beta$, then $(\Psi, \mathcal{X})^{\cap \alpha} \subseteq (\Psi, \mathcal{X})^{\cap \beta}$.
- ii) $(\Psi^c, \mathcal{X})^{\cap \alpha} = \{t \in \mathcal{X} \mid \alpha \setminus \Psi(t) \neq \emptyset\}.$
- iii) $(\Psi, \mathcal{X})^{\cap (U \setminus \alpha)} = \{t \in \mathcal{X} | \Psi(t) \setminus \alpha \neq \emptyset\}.$
- iv) $(\Psi^c, \mathcal{X})^{\cap (U \setminus \alpha)} = \{t \in \mathcal{X} | U \setminus (\Psi(t) \cup \alpha) \neq \emptyset\}.$

Proof. If $\alpha \subset \beta$, then $\Psi(t) \cap \alpha \neq \emptyset$ implies $\Psi(t) \cap \beta \neq \emptyset$. Hence the proof of (i) is done. The rest of proof is obtained using algebraic operations, easily. \Box

Proposition 4. Let (Ψ, \mathcal{X}) be a soft set over U and $\emptyset \neq \alpha \subseteq U$. If $(\Psi^c, \mathcal{X})^{\cap (U \setminus \alpha)} =$ \emptyset then $(\Psi, \mathcal{X})^{\cap \alpha} \cup (\Psi^c, \mathcal{X})^{\cap \alpha} \cup (\Psi, \mathcal{X})^{\cap (U \setminus \alpha)} = \mathcal{X}$.

Proof. Let (Ψ, \mathcal{X}) be a soft set over U and $\emptyset \neq \alpha \subseteq U$. We assume that $(\Psi^c, \mathcal{X})^{\cap (U \setminus \alpha)} =$ $θ$. Then, by Proposition [3](#page-103-0) (iv), we have $Ψ(t) ∪ α = U$ for all $t ∈ X$. Hence, the proof is obvious from Definition $\boxed{9}$ and Proposition $\boxed{3}$ (ii) and (iii).

Proposition 5. Let (Ψ, \mathcal{X}) be a soft set over U and let $\emptyset \neq \alpha \subsetneq U$. If $\Psi(t) \cup \alpha \neq U$ for all $t \in \mathcal{X}$, then

- i) $supp(\Psi, \mathcal{X}) \subseteq (\Psi^c, \mathcal{X})^{\cap (U \setminus \alpha)}$.
- ii) $(\Psi, \mathcal{X})^{\cap \alpha} \subseteq (\Psi^c, \mathcal{X})^{\cap (U \setminus \alpha)}$.

Proof. (i) Let $t \in supp(\Psi, \mathcal{X})$. Since $\Psi(t) \cup \alpha \neq U$ for all $t \in \mathcal{X}$, then $U \setminus (\Psi(t) \cup \alpha) \neq \emptyset$ \emptyset , which implies $t \in (\Psi^c, \mathcal{X})^{\cap (U \setminus \alpha)}$ by Proposition \mathfrak{Z} (iv). (ii) It is seen from the assertion (i) and Proposition $\boxed{2}$ (i).

Proposition 6. Let (Ψ, \mathcal{X}) be a soft set over U and $\emptyset \neq \alpha, \beta \subseteq U$. Then

- i) $(\Psi, \mathcal{X})^{\cap \alpha} \cap (\Psi, \mathcal{X})^{\cap \beta} \subseteq (\Psi, \mathcal{X})^{\cap (\alpha \cup \beta)}$. Here the equality does not hold in general, even if $\alpha \cap \beta = \emptyset$.
- ii) $(\Psi, \mathcal{X})^{\cap \alpha} \cup (\Psi, \mathcal{X})^{\cap \beta} = (\Psi, \mathcal{X})^{\cap (\alpha \cup \beta)}.$
- iii) $(\Psi, \mathcal{X})^{\cap (\alpha \cap \beta)} \subseteq (\Psi, \mathcal{X})^{\cap \alpha} \cap (\Psi, \mathcal{X})^{\cap \beta}.$

Proof. (i) Let $t \in (\Psi, \mathcal{X})^{\cap \alpha} \cap (\Psi, \mathcal{X})^{\cap \beta}$. Then $\Psi(t) \cap \alpha \neq \emptyset$ and $\Psi(t) \cap \beta \neq \emptyset$, which implies $\Psi(t) \cap (\alpha \cup \beta) \neq \emptyset$. For the rest of the proof, we have the Example [1.](#page-104-0) (ii)

$$
(\Psi, \mathcal{X})^{\cap \alpha} \cup (\Psi, \mathcal{X})^{\cap \beta} = \{ t \in supp(\Psi, \mathcal{X}) | (\Psi(t) \cap \alpha \neq \emptyset) \vee (\Psi(t) \cap \beta \neq \emptyset) \}
$$

=
$$
\{ t \in supp(\Psi, \mathcal{X}) | \Psi(t) \cap (\alpha \cup \beta) \neq \emptyset) \}
$$

=
$$
(\Psi, \mathcal{X})^{\cap (\alpha \cup \beta)}
$$

(iii) Let $t \in (\Psi, \mathcal{X})^{\cap(\alpha \cap \beta)}$. Then $\Psi(t) \cap (\alpha \cap \beta) \neq \emptyset$, which implies $\Psi(t) \cap \alpha \neq \emptyset$ and $\Psi(t) \cap \beta \neq \emptyset$. Therefore $t \in (\Psi, \mathcal{X})^{\cap \alpha} \cap (\Psi, \mathcal{X})^{\cap \beta}$.

Example 1. Let the universe $U = \{u_1, u_2, u_3, u_4, u_5, u_6\}$, the parameter set $\mathcal{T} =$ $\{t_1, t_2, t_3, t_4, t_5, t_6, t_7\}$, and $\mathcal{X} = \{t_1, t_3, t_4, t_5\}$ and $\mathcal{Y} = \{t_1, t_2, t_3, t_4\}$ be two subsets of $\mathcal T$. Suppose that corresponding soft sets of $\mathcal X$ and $\mathcal Y$ are

 $(\Psi, \mathcal{X}) = \{(t_1, \{u_1, u_2\}), (t_3, \{u_1, u_4, u_5\}), (t_4, \{u_6\}), (t_5, \emptyset)\}\$

and

$$
(\Upsilon, \mathcal{Y}) = \{ (t_1, \{u_3, u_4\}), (t_2, \{u_1, u_2, u_5\}), (t_3, \{u_3, u_5\}), (t_4, \{u_1, u_5, u_6\}) \}.
$$

If $\alpha = \{u_4, u_6\}$ and $\beta = \{u_1\}$, then it is seen that $(\Psi, \mathcal{X})^{\cap \alpha} = \{t_3, t_4\}$, $(\Psi, \mathcal{X})^{\cap \beta} =$ ${t_1, t_3}$ and $(\Psi, \mathcal{X})^{\cap(\alpha \cup \beta)} = {t_1, t_3, t_4}$. (Because, it is obtained that $\Psi(t_3) \cap \alpha =$ ${u_4} \neq \emptyset, \Psi(t_4) \cap \alpha = {u_6} \neq \emptyset, \Psi(t_1) \cap \beta = {u_1} \neq \emptyset, \Psi(t_3) \cap \beta = {u_1} \neq \emptyset,$ $\Psi(t_1) \cap (\alpha \cup \beta) = \{u_1\} \neq \emptyset$, $\Psi(t_3) \cap (\alpha \cup \beta) = \{u_1, u_4\} \neq \emptyset$ and $\Psi(t_3) \cap (\alpha \cup \beta) =$ ${u_6}\neq \emptyset$. Thus, the proof of Proposition $\overline{6}$ (i) is completed. Since $\alpha \cap \beta = \emptyset$, we have $(\Psi, \mathcal{X})^{\cap (\alpha \cap \beta)} = \emptyset$.

If $\alpha = \{u_3, u_6\}$ and $\beta = \{u_5, u_6\}$ (i.e., $\alpha \cap \beta = \{u_6\}$), then it is seen that $(\Upsilon, \mathcal{Y})^{\cap \alpha} = \{t_1, t_3, t_4\}, \ (\Upsilon, \mathcal{Y})^{\cap \beta} = \{t_2, t_3, t_4\} \text{ and } (\Upsilon, \mathcal{Y})^{\cap (\alpha \cap \beta)} = \{t_4\}.$ So, we have $(\Upsilon, \mathcal{Y})^{\cap (\alpha \cap \beta)} \subseteq (\Upsilon, \mathcal{Y})^{\cap \alpha} \cap (\Upsilon, \mathcal{Y})^{\cap \beta}$.

4. Set-Generated Soft Subrings of Rings

In this section, we propose the set-generated soft subrings of a ring by employing the α -intersection of soft sets. We also discuss some of the main properties and theoretical implications of this newly emerging soft algebraic structure.

Throughout this section, \Re is a ring and (Ψ, \Re) is a soft set over \Re . A subring S of \Re denoted by $S < \Re$.

Definition 10. Let \Re be a ring, $\emptyset \neq \alpha \subseteq \Re$ and (Ψ, \Re) be a soft set over \Re . If the soft set $(\Psi, (\Psi, \mathfrak{R})^n)$ is a soft subring of \mathfrak{R} , then this soft set is said to be a soft subring of \Re generated by the set α and denoted by $\langle \Psi^{\cap \alpha} \rangle_{\Re}$. If the set $\alpha = \{t\},$ $\langle \Psi \cap \alpha \rangle_{\mathfrak{R}}$ is a soft subring of \mathfrak{R} generated by the element $t \in \mathfrak{R}$.

As can be seen Definition $\overline{10}$, $\langle \Psi \cap \alpha \rangle_{\Re} \tilde{\prec} \Re$ if and only if there exists at least an $\emptyset \neq \alpha \subseteq \Re$ such that $(\Psi, \Re)^{\overline{\alpha}}$ is a subring of \Re and the conditions s_1, s_2 of the Definition $\overline{7}$ are satisfied for $S = (\Psi, \Re)^{\cap \alpha}$.

Example 2. Given the ring $\Re = (\mathbb{Z}_6, +, .),$ a soft set (Ψ, \Re) over \Re , where Ψ : $\Re \to P(\Re)$ is a set-valued function defined by $\Psi(0) = \{0, 1, 4, 5\}, \Psi(1) = \{3\}, \Psi(2) =$ ${2}, \Psi(3) = \{0, 4, 5\}, \Psi(4) = \{1, 2\}$ and $\Psi(5) = \{3\}.$ Let $\alpha = \{4, 5\} \subseteq \Re$. Then, $(\Psi, \mathfrak{R})^{\cap \alpha} = \{0, 3\}$ is a subring of $\mathfrak R$ and the soft set $(\Psi, (\Psi, \mathfrak{R})^{\cap \alpha}) = \{(0, \{0, 1, 4, 5\}),\}$ $(3,\{0,4,5\})\}$ satisfies the conditions s1, s2 of the Definition $\overline{7}$. (That is, $\Psi^{\cap\alpha}(t-\overline{7})$ $(v) \supseteq \Psi^{\cap\alpha}(t) \cap \Psi^{\cap\alpha}(v)$ and $\Psi^{\cap\alpha}(tv) \supseteq \Psi^{\cap\alpha}(t) \cap \Psi^{\cap\alpha}(v)$ for all $t, v \in \Re$). Hence $(\Psi, (\Psi, \mathbb{R})^n)^\alpha = \langle \Psi^{n\alpha} \rangle_{\mathbb{R}} \widetilde{\leq} \mathbb{R}$. If $\beta = \{4\}$ a single point set, then it is seen that
 $(\Psi, \mathbb{R})^n$ (H, \mathbb{R}^n) \Box^{α} and then $\langle \Psi^{n\alpha} \rangle_{\alpha}$ (H) \Box^{α} $(\Psi, \mathcal{R})^{\cap \alpha} = (\Psi, \mathcal{R})^{\cap \beta}$ and then $\langle \Psi^{\cap \alpha} \rangle_{\mathcal{R}} = \langle \Psi^{\cap \beta} \rangle_{\mathcal{R}}$. Therefore, the soft set $\{(0, \{0, 1, 4, 5\},\)$ $(3,\{0,4,5\})\}$ is a soft subring of \Re , generated by the element $4 \in \Re$.

Let (Ψ, \Re) be a soft set over \Re . Since $\{0_{\Re}\}\$ is a subring of \Re , it is easily seen that $(\Psi, \{0_{\Re}\})\tilde{\leq} \Re$.

Definition 11. The soft subring $(\Psi, \{0\})$ of \mathbb{R} is called a trivial soft subring of \Re and denoted by $\langle 0_{\Re} \rangle_{\Psi}$.

It is important to note that the soft sets $\langle 0_{\Re} \rangle_{\Psi}$ and $(\Psi, (\Psi, \mathfrak{R})^{\cap \{0_{\Re}\}})$ are different, in general. In Example $\overline{2}$, $(0_{\Re})_{\Psi} = (\Psi, \{0_{\Re}\}) = \{(0, \{0, 1, 4, 5\})\}$ and $(\Psi, (\Psi, \mathfrak{R}) \cap \{0\}\mathfrak{R}) = \{ (0, \{0, 1, 4, 5\}), (3, \{0, 4, 5\}) \}.$ Furthermore, $(\Psi, (\Psi, \mathfrak{R}) \cap \{0\}\mathfrak{R})$ does not have to be a soft subring of ℜ.

Proposition 7. If $\langle \Psi \cap \{\alpha \} \rangle_{\Re} \leq \Re$, then the generator α doesn't have to be unique.
Furthermore, if $\langle \Psi \cap {\{\iota\}} \rangle_{\Re} = \langle \Psi \cap {\{\iota\}} \rangle_{\Re}$ for $t, v \in \Re$, then $\langle \Psi \cap {\{\iota\}} \rangle_{\Re} = \langle \Psi \cap {\{\iota\}} \rangle_{\Re}$.

Proof. In the example $\boxed{2}$ if we take $\eta = \{5\} \subseteq \Re$, then it is seen that $(\Psi, \Re)^{\cap \eta} =$ $(\Psi, \mathfrak{R})^{\cap \beta}$ and then $\langle \Psi^{\overline{\cap} \eta} \rangle_{\mathfrak{R}} = \langle \Psi^{\cap \beta} \rangle_{\mathfrak{R}}$. Hence the generator doesn't have to be unique, even if it is a single point set. Now, let $\langle \Psi \cap {\{t\}} \rangle_{\Re} = \langle \Psi \cap {\{v\}} \rangle_{\Re}$ for $t, v \in \Re$. Consider the sets $\mathcal{X} = \{t \in \Re : \Psi(t) \cap \{t\} \neq \emptyset\} = \{t \in \Re : \Psi(t) \cap \{v\} \neq \emptyset\}$ and $\mathcal{Y} = \{t \in \mathbb{R} : \Psi(t) \cap \{t, v\} \neq \emptyset\}.$ Obviously $\mathcal{X} \subseteq \mathcal{Y}$. Let $t \in \mathcal{Y}$. Then,

$$
\Psi(t) \cap \{t, v\} \neq \emptyset \Rightarrow \Psi(t) \cap \{t\} \neq \emptyset \text{ or } \Psi(t) \cap \{v\} \neq \emptyset
$$

\n
$$
\Rightarrow t \in \mathcal{X} \text{ or } t \in \mathcal{X}
$$

\n
$$
\Rightarrow t \in \mathcal{X}
$$

Hence $\mathcal{Y} \subseteq \mathcal{X}$. Therefore, $(\Psi, \Re)^{\cap \{t\}} = (\Psi, \Re)^{\cap \{v\}} = (\Psi, \Re)^{\cap \{t,v\}}$, which implies that $\langle \Psi \cap \{t,v\} \rangle_{\Re} = \langle \Psi \cap \{t\}$ \rangle_{\Re} . \Box

Proposition 8. If $\langle \Psi \cap \alpha \rangle_{\Re} \leq \Re$, then $\Psi(0_{\Re}) \cap \alpha \neq \emptyset$. But the reverse implication is not true, in general.

Proof. Let $\langle \Psi \cap \alpha \rangle_{\Re} \tilde{\prec} \Re$. Then the set $(\Psi, \Re) \cap \alpha = \{t \in \Re : \Psi(t) \cap \alpha \neq \emptyset\}$ is a subring of R. Then $\Psi(0_{\Re}) \supseteq \Psi(t)$ for all $t \in (\Psi, \Re)^{\cap \alpha}$ by Proposition $\boxed{1}$. Since $\Psi(t) \cap \alpha \neq \emptyset$ and $\Psi(0_{\Re}) \supseteq \Psi(t)$ for all $t \in (\Psi, \Re)^{\cap \alpha}$, then $\Psi(0_{\Re}) \cap \alpha \neq \emptyset$. For the rest of the proof, let $\lambda = \{0, 1, 5\} \subseteq \Re$ in Example [2.](#page-105-0) Then it is seen that $\Psi(0_{\Re}) \cap \lambda \neq \emptyset$, but

 $(\Psi, \mathfrak{R})^{\cap \lambda} = \{0, 3, 4\}$ is not a subring of \mathfrak{R} . Therefore, $(\Psi, (\Psi, \mathfrak{R})^{\cap \lambda})$ is not a soft subring of \Re .

Proposition 9. Let $\langle \Psi \cap \alpha \rangle_{\Re} \tilde{\prec} \Re$ and $\langle \Psi \cap \beta \rangle_{\Re} \tilde{\prec} \Re$. If $\alpha \subseteq \beta$, then $\langle \Psi \cap \alpha \rangle_{\Re} \subseteq \langle \Psi \cap \beta \rangle_{\Re}$.

Proof. Let $t \in (\Psi, \mathfrak{R})^{\cap \alpha}$. Then $\Psi(t) \cap \alpha \neq \emptyset$. Since $\alpha \subseteq \beta$, $\Psi(t) \cap \beta \neq \emptyset$. Hence $(\Psi, \mathfrak{R})^{\cap \alpha} \subseteq (\Psi, \mathfrak{R})^{\cap \beta}$. Therefore $(\Psi, (\Psi, \mathfrak{R})^{\cap \alpha}) \subseteq (\Psi, (\Psi, \mathfrak{R})^{\cap \beta})$, which completes the proof. \Box

The following Theorem shows that Theorem $\mathbf{1}$ is also true for the operation soft intersection instead of restricted intersection when taking the soft set (Ψ, \Re) instead of (Ψ, S) .

Theorem 2. If $(\Psi, \Re) \tilde{\leq} \Re$ and $(\Upsilon, \Re) \tilde{\leq} \Re$, then $(\Psi, \Re) \tilde{\cap} (\Upsilon, \Re) \tilde{\leq} \Re$.

Proof. By Definition $\overline{6}(\Psi, \Re) \widetilde{\cap}(\Upsilon, \Re) = (\Theta, \Re)$, where $\Theta(t) = \Psi(t) \cap \Upsilon(t)$ for all $t \in \Re$. Then for all $t, v \in \Re$,

$$
\Theta(t - v) = \Psi(t - v) \cap \Upsilon(t - v)
$$

\n
$$
\supseteq (\Psi(t) \cap \Psi(v)) \cap (\Upsilon(t) \cap \Upsilon(v))
$$

\n
$$
= (\Psi(t) \cap \Upsilon(t)) \cap (\Psi(v) \cap \Upsilon(v))
$$

\n
$$
= \Theta(t) \cap \Theta(v),
$$

\n
$$
\Theta(tv) = \Psi(tv) \cap \Upsilon(tv)
$$

\n
$$
\supseteq (\Psi(t) \cap \Psi(v)) \cap (\Upsilon(t) \cap \Upsilon(v))
$$

\n
$$
= (\Psi(t) \cap \Upsilon(t)) \cap (\Psi(v) \cap \Upsilon(v))
$$

\n
$$
= \Theta(t) \cap \Theta(v).
$$

Therefore $(\Psi, \Re) \widetilde{\cap} (\Upsilon, \Re) \widetilde{\leq} \Re$.

Now, some problems arise such that: Is the soft intersection of two set-generated soft subrings of \mathcal{R} , again a set-generated soft subring of \mathcal{R} ? And, if $\langle \Psi \cap \alpha \rangle_{\mathcal{R}} \tilde{\leq} \mathcal{R}$, $\langle \Psi^{\cap \beta} \rangle_{\Re} \tilde{\ll} \Re$ such that $\langle \Psi^{\cap \alpha} \rangle_{\Re} \tilde{\cap} \langle \Psi^{\cap \beta} \rangle_{\Re} = \langle \Psi^{\cap \xi} \rangle_{\Re}$, then can the subset ξ be ex-
present using a and β ? The ex-wear of the first problem is "Ne" we have the pressed using α and β ? The answer of the first problem is "No", we have the following example:

Example 3. Given the ring $\mathbb{R} = (\mathbb{Z}_{12}, +, .),$ a soft set (Ψ, \mathbb{R}) over \mathbb{R} , where Ψ : $\Re \to P(\Re)$ is a set-valued function defined by $\Psi(0) = \{1,3,5,6,7,9,11\}, \Psi(1) =$ ${2, 4}, \Psi(2) = {3, 6, 7, 11}, \Psi(3) = {1, 5, 9}, \Psi(4) = {3, 6, 7, 11}, \Psi(5) = {8, 10},$ $\Psi(6) = \{1, 3, 5, 6, 7, 9, 11\}, \Psi(7) = \{2, 10\}, \Psi(8) = \{3, 6, 7, 11\}, \Psi(9) = \{1, 5, 9\},\$ $\Psi(10) = \{3, 6, 7, 11\}$ and $\Psi(11) = \{2, 8\}$. Let $\alpha = \{11\}$ and $\beta = \{5\}$. Then, $(\Psi, \Re)^{\cap \alpha} = \{0, 2, 4, 6, 8, 10\}$ and $(\Psi, \Re)^{\cap \beta} = \{0, 3, 6, 9\}$ are subrings of \Re and the soft sets

 $(\Psi, (\Psi, \Re)^{\cap \alpha}) = \begin{cases} (0, \{1, 3, 5, 6, 7, 9, 11\}), (2, \{3, 6, 7, 11\}), (4, \{3, 6, 7, 11\}), (4, \{3, 6, 7, 11\}), (5, 6, 7, 11), (10, \{3, 6, 7, 11\}), (2, \{4, 7, 11\}), (3, 6, 7, 11), (4, \{3, 6, 7, 11\}), (5, 6, 7, 11), (10, \{3, 6, 7, 11\}), (10, \{3,$ $(6, \{1, 3, 5, 6, 7, 9, 11\}), (8, \{3, 6, 7, 11\}), (10, \{3, 6, 7, 11\})$ <u>)</u>

and

$$
(\Psi, (\Psi, \mathfrak{R})^{\cap \beta}) = \left\{ \begin{array}{c} (0, \{1, 3, 5, 6, 7, 9, 11\}), (3, \{1, 5, 9\}), \\ (6, \{1, 3, 5, 6, 7, 9, 11\}), (9, \{1, 5, 9\}) \end{array} \right\}
$$

satisfy the conditions s1, s2 of Definition λ . Hence $\langle \Psi^{\cap \alpha} \rangle_{\Re} \tilde{\prec} \Re$ and $\langle \Psi^{\cap \beta} \rangle_{\Re} \tilde{\prec} \Re$. Then,

$$
\langle \Psi^{\cap \alpha} \rangle_{\Re} \widetilde{\cap} \langle \Psi^{\cap \beta} \rangle_{\Re} = \{ (0, \{1, 3, 5, 6, 7, 9, 11\}), (6, \{1, 3, 5, 6, 7, 9, 11\}) \} = (\Psi, S) \widetilde{\leq} \Re.
$$

But, there is no subset ξ of \Re such that $(\Psi, S) = \langle \Psi \cap \xi \rangle_{\Re}$.

Corollary 1. The soft intersection of two set-generated soft subrings of \Re is not a set-generated soft subring of ℜ, in general.

But, we have the following:

Theorem 3. Let $\langle \Psi \cap \alpha \rangle_{\Re} \widetilde{\prec} \Re$ and $\langle \Psi \cap \beta \rangle_{\Re} \widetilde{\prec} \Re$. Then, either $\langle \Psi \cap \alpha \rangle_{\Re} \widetilde{\cap} \langle \Psi \cap \beta \rangle_{\Re} =$ $\langle 0_{\Re} \rangle_{\Psi}$ trivial soft subring or if $\langle \Psi^{\cap \alpha} \rangle_{\Re} \cap \langle \Psi^{\cap \beta} \rangle_{\Re} = \langle \Psi^{\cap \xi} \rangle_{\Re}$, then there exists $\xi \subseteq \Re$ such that $\emptyset \neq \xi \subset \alpha \cup \beta$.

Proof. If $\langle \Psi^{\cap \alpha} \rangle_{\Re} \tilde{\cap} \langle \Psi^{\cap \beta} \rangle_{\Re} = \langle 0_{\Re} \rangle_{\Psi}$, it is obvious. Assume that

$$
\langle 0_{\Re} \rangle_{\Psi} \neq \langle \Psi^{\cap \alpha} \rangle_{\Re} \widetilde{\cap} \langle \Psi^{\cap \beta} \rangle_{\Re} = \langle \Psi^{\cap \xi} \rangle_{\Re}.
$$

Then $\langle \Psi^{\cap \xi} \rangle_{\Re} \tilde{\prec} \Re$ by Theorem <mark>2.</mark> Since $\langle \Psi^{\cap \alpha} \rangle_{\Re} \tilde{\cap} \langle \Psi^{\cap \beta} \rangle_{\Re} = \langle \Psi^{\cap \xi} \rangle_{\Re}$, then we have

$$
U \cap \alpha \neq \emptyset \land U \cap \beta \neq \emptyset \Leftrightarrow U \cap \xi \neq \emptyset
$$

for $(t, U) \in \langle \Psi \cap \xi \rangle_{\Re}$. The requirement (1) holds for:

i) $\alpha \subseteq \beta \Rightarrow \xi = \beta$, ii) $\beta \subseteq \alpha \Rightarrow \xi = \alpha$, iii) $\alpha \cap \beta = \emptyset \Rightarrow \xi = \alpha \cup \beta$, iv) $\alpha \neq \beta$ and $\alpha \cap \beta \neq \emptyset \Rightarrow \xi = \alpha \cup \beta$.

Although the requirement (1) also holds for $\xi \supseteq \alpha \cup \beta$, it is enough to show existing $\xi \subseteq \Re$ such that $\emptyset \neq \xi \subseteq \alpha \cup \beta$ to complete the proof. \Box

Definition 12. ($[29]$) Let \Re_1 and \Re_2 be two rings, and (Ψ, S_1) and (Υ, S_2) be two soft subrings of \Re_1 and \Re_2 , respectively. The product of soft subrings (Ψ, S_1) and (Υ, S_2) is defined as $(\Psi, S_1) \times (\Upsilon, S_2) = (\Omega, S_1 \times S_2)$, where $\Omega(t, v) = \Psi(t) \times \Upsilon(v)$ for all $(t, v) \in S_1 \times S_2$.

Theorem 4. ($\boxed{29}$) If $\Psi_{S_1} \widetilde{\times} \Re_1$ and $\Upsilon_{S_2} \widetilde{\times} \Re_2$, then $\Psi_{S_1} \times \Upsilon_{S_2} \widetilde{\times} \Re_1 \times \Re_2$.

Theorem $\sqrt{4}$ leads to the problem: Is the product of two set-generated soft subrings of two rings, again a set-generated soft subring of the ring of product of rings? The answer is "Yes", we have the following:
Theorem 5. Let \Re_1 and \Re_2 be two rings and let $(\Psi, \Re_1), (\Upsilon, \Re_2)$ be two soft sets over \Re_1 and \Re_2 , respectively. If there exist $\alpha \subseteq \Re_1$ and $\beta \subseteq \Re_2$ such that $\langle \Psi^{\cap \alpha} \rangle_{\Re_1} \tilde{\leq} \Re_1$ and $\langle \Upsilon^{\cap \beta} \rangle_{\Re_2} \tilde{\leq} \Re_2$, then

$$
\langle \Psi^{\cap \alpha} \rangle_{\Re_1} \times \langle \Upsilon^{\cap \beta} \rangle_{\Re_2} = \langle \Theta^{\cap (\alpha \times \beta)} \rangle_{\Re_1 \times \Re_2}.
$$

Proof. Let $\langle \Psi \cap \alpha \rangle_{\Re_1} \widetilde{\leq} \Re_1$ and $\langle \Upsilon \cap \beta \rangle_{\Re_2} \widetilde{\leq} \Re_2$. Then $(\Psi, \Re_1) \cap \alpha$ is a subring of \Re_1 and $\langle \Upsilon, \Re \rangle_{\Re_2} \widetilde{\leq} \Re_2$. Then $(\Psi, \Re_1) \cap \alpha$ is a subring of \Re_1 and $(\Upsilon, \Re_2)^{\cap \beta}$ is a subring of \Re_2 . So $(\Psi, \Re_1)^{\cap \alpha} \times (\Upsilon, \Re_2)^{\cap \beta}$ is a subring of $\Re_1 \times \Re_2$. Therefore, $\langle \Psi^{\cap \alpha} \rangle_{\Re_1} \times \langle \Upsilon^{\cap \beta} \rangle_{\Re_2} \tilde{\ll} \Re_1 \times \Re_2$ by Theorem [4.](#page-107-0) Now, let $(t, \Psi(t)) \in \langle \Psi^{\cap \alpha} \rangle_{\Re_1}$
and $(v, \Upsilon(v)) \in \langle \Upsilon^{\cap \beta} \rangle_{\Re_2}$. Then $\Psi(t) \cap \alpha \neq \emptyset$ and $\Upsilon(v) \cap \beta \neq \emptyset$. Since

$$
\Psi(t) \cap \alpha \neq \emptyset \land \Upsilon(v) \cap \beta \neq \emptyset \Leftrightarrow (\Psi(t) \times \Upsilon(v)) \cap (\alpha \times \beta) \neq \emptyset,
$$

then we have

$$
(t,\Psi(t))\in \langle \Psi^{\cap\alpha}\rangle_{\Re_1}\wedge (v,\Upsilon(v))\in \langle \Upsilon^{\cap\beta}\rangle_{\Re_2}\Leftrightarrow ((t,v),\Psi(t)\times \Upsilon(v))\in \langle \Theta^{\cap(\alpha\times\beta)}\rangle_{\Re_1\times \Re_2}.
$$

 $(\Theta, \Re_1 \times \Re_2)$ is a soft set over $\Re_1 \times \Re_2$, where $\Theta : \Re_1 \times \Re_2 \to P(\Re_1 \times \Re_2)$ is a set-valued function defined by $\Theta(t, v) = \Psi(t) \times \Upsilon(v)$. Hence, the proof is completed. \square

Example 4. Over the ring $\Re_1 = (\mathbb{Z}_4, +, .),$ a soft set (Ψ, \Re_1) given by $\Psi(0) =$ $\{1, 2, 3\}, \Psi(1) = \{0\}, \Psi(2) = \{1, 3\}, \Psi(3) = \{2\}.$ For $\alpha = \{3\}, (\Psi, \Re_1)^{\cap \alpha} = \{0, 2\}$ and $\langle \Psi \cap \alpha \rangle_{\Re_1} = \{ (0, \{1, 2, 3\}), (2, \{1, 3\}) \} \tilde{\leq} \Re_1$. Given the ring $\Re_2 = (\mathbb{Z}_6, +, .),$ a soft set (Υ, \Re_2) over \Re_2 , defined by $\Upsilon(0) = \{0, 1, 2, 5\}$, $\Upsilon(1) = \{3, 4\}$, $\Upsilon(2) = \{4\}$, $\Upsilon(3) = \{0, 2\}, \ \Upsilon(4) = \{3\} \ and \ \Upsilon(5) = \{4\}. \ \ \text{For } \beta = \{2\}, \ (\Upsilon, \Re_2)^{\cap \beta} = \{0, 3\} \ \text{and}$ $\langle \Upsilon \cap \beta \rangle_{\Re_2} = \{ (0, \{0, 1, 2, 5\}), (3, \{0, 2\}) \} \tilde{\leq} \tilde{\Re_2}. \ \langle \Psi \cap \alpha \rangle_{\Re_1} \times \langle \Upsilon \cap \beta \rangle_{\Re_2} \ \text{is a soft set given}$ by

$$
\left\{\n\begin{array}{l}\n((0,0),\{(1,0),(1,1),(1,2),(1,5),(2,0),(2,1),(2,2),(2,5),(3,0),(3,1),(3,2),(3,5)\},\\
((2,0),\{(1,0),(1,1),(1,2),(1,5),(3,0),(3,1),(3,2),(3,5)\}),\\
((0,3),\{(1,0),(1,2),(2,0),(2,2),(3,0),(3,2)\}),\\
((2,3),\{(1,0),(1,2),(3,0),(3,2)\})\n\end{array}\n\right\}
$$

Now, let the soft set $(\Theta, \Re_1 \times \Re_2)$ over $\Re_1 \times \Re_2$, where $\Theta : \Re_1 \times \Re_2 \to P(\Re_1 \times \Re_2)$ is a set-valued function defined by $\Theta(t, v) = \Psi(t) \times \Upsilon(v)$. Then, for $\alpha \times \beta = \{(3, 2)\}\,$, it is easily seen that $\langle \Theta^{\cap(\alpha \times \beta)} \rangle_{\Re_1 \times \Re_2} = \langle \Psi^{\cap \alpha} \rangle_{\Re_1} \times \langle \Upsilon^{\cap \beta} \rangle_{\Re_2}.$

5. Conclusions

In this paper, we are interested in the algebraic soft substructures of rings given in the article [\[29\]](#page-110-0). We introduced set-generated soft subrings of rings using nonempty subsets of rings. By theoretical directions, we applied some of the operations derived on soft sets to set-generated soft subrings. Moreover, we gave some relationships between the generators of soft subrings and studied their related various properties with assorted examples. To further this work, one could study the setgenerated soft substructures of other algebraic structures such as fields, modules, vector spaces and algebras. Our future work will be based on the derivation of these algebraic structures and the investigation their application aspects.

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CHEN INVARIANTS FOR RIEMANNIAN SUBMERSIONS AND THEIR APPLICATIONS

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Abstract. In this paper, an optimal inequality involving the delta curvature is exposed. With the help of this inequality some characterizations about the vertical motion and the horizontal divergence are obtained.

1. INTRODUCTION

The celebrated divergence theorem states that divergence of a vector field indicates how much the vector spreads out from the certain point. In fluid kinematics, if a vector field X is considered as velocity of a fluid or a gas, then sign of $div(X)$ describes the expansion or compression of flow. Therefore, the total expansion or compression of flow can be calculated by the help of divergence theorem so divergence is a useful tool to measuring the net flow of fluid diverging from a point or approaching a point. The first phenomenon is called as horizontal divergence and the other is called as horizontal convergence.

The continuity equation simple states that any matter can either be created or destroyed and implies for the atmosphere that its mass may be redistributed but can never be disappeared. Therefore, this equation gives us that

$$
\operatorname{div}(U) = 0 \tag{1}
$$

for any vector field $U = (u^1, u^2, u^3)$ on E^3 . It can be written from Π that

$$
\operatorname{div}_H(U) + \frac{\partial u^3}{\partial z} = 0,\t\t(2)
$$

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¹⁰⁰⁷

where $\text{div}_H(U)$ is the horizontal divergence of U defined by

$$
\operatorname{div}_H(U) = \frac{\partial u^1}{\partial x} + \frac{\partial u^2}{\partial y}.
$$
\n(3)

The equation given (2) is also known as the *continuity equation* in literature. Integrating (2) , we have

$$
\omega(p_1, p_0) \equiv u^3(p_1) - u^3(p_0) = -\int_{p_0}^{p_1} \left(\frac{\partial u^1}{\partial x} + \frac{\partial u^2}{\partial y}\right) dz,\tag{4}
$$

where p_1 and p_0 is some pressure levels on the atmosphere. If we assume that p_0 is the surface pressure then $u^3(p_0) = 0$ and thus we get

$$
\omega(p_1) = -\int_{p_0}^{p_1} \left(\frac{\partial u^1}{\partial x} + \frac{\partial u^2}{\partial y}\right) dz.
$$
\n(5)

This formula tells us that w at a given pressure level is proportional to the integral of the horizontal divergence. Here, $\omega(p_1)$ is called the vertical motion at p_1 . If $\omega(p) < 0$ at every point p then this statement is called rising motion, $\omega(p) > 0$ at every point p then this statement is called *descending motion*, (in this case, divergence is called convergence) in meteorology. There is no divergence and it is clear that there is a local maximum or minimum of w.

Beside these facts, B.-Y. Chen $\boxed{7}$ initially introduced a new invariant the socalled delta curvature δ for an *n*-dimensional Riemannian manifold M by

$$
\delta^{k}(p) = \tau(p) - \left(\inf \tau(\Pi_{k})\right)(p),\tag{6}
$$

where $2 \leq k \leq n-1$, $\tau(p)$ is the scalar curvature at $p \in M$ and

 $(\inf \tau (\Pi_k))(p) = \inf \{ \tau (\Pi_k) | \Pi_k \text{ is a } k\text{-plane section } \subset T_pM \}.$

Furthermore, he gave a relation involving the delta curvature, the main intrinsic and extrinsic invariants of submanifolds in a real space form (cf. Lemma 3.2 in $\boxed{7}$). Then, this curvature drew attention of many authors and the notion of discovering simple basic relationships between intrinsic and extrinsic invariants of a submanifold becomes one of the most fundamental problems in submanifold theory (cf. $[1, 3, 8]$ $[1, 3, 8]$ $[1, 3, 8]$) $[10, 11, 19, 23, 24]$ $[10, 11, 19, 23, 24]$ $[10, 11, 19, 23, 24]$ $[10, 11, 19, 23, 24]$ $[10, 11, 19, 23, 24]$ $[10, 11, 19, 23, 24]$, etc.). Furthermore, various inequalities and their applications on Riemannian submersions were studied recently in $[4, 12, 15, 22]$ $[4, 12, 15, 22]$ $[4, 12, 15, 22]$ $[4, 12, 15, 22]$.

Apart from isometric immersions and submanifolds theory, Riemannian submersions have played a substantial role in differential geometry since this frame of maps also makes possible to compare geometrical properties between smooth manifolds. Besides the mathematical significance, Riemannian submersions have important physical and engineering aspects. There exist very nice applications of these mappings in the Kaluza-Klein theory $\left|13, 16, 25\right|$ $\left|13, 16, 25\right|$ $\left|13, 16, 25\right|$, in the statical machine learning process $[26]$, in the medical imaging $[18]$, in the statical analysis $[6]$, in the robotic theory $\sqrt{2, 20, 21}$ $\sqrt{2, 20, 21}$ $\sqrt{2, 20, 21}$.

Motivated by these facts, we firstly establish an optimal inequality involving the delta curvature for Riemannian manifolds admitting a Riemannian submersion.

Then, we investigate this inequality for some special cases. Finally, we obtain some results dealing the vertical motion and horizontal divergence.

2. Preliminaries

Let (M, g) be and n dimensional Riemannian manifold with Riemannian metric g. The sectional curvature, denoted $K_M(e_i \wedge e_j)$, of the plane section spanned by orthogonal unit vectors e_i and e_j at $p \in M$ is

$$
K(e_i \wedge e_j) \equiv R(e_i, e_j, e_j, e_i) = R(e_j, e_i, e_i, e_j), \tag{7}
$$

where R is the Riemann curvature tensor. Usually the sectional curvature $K(e_i \wedge e_j)$ is denoted by K_{ii} .

Let $\{e_1, \ldots, e_n\}$ be any orthonormal basis for T_pM . In particular, the Ricci curvature Ric is defined by

$$
Ric(X) = \sum_{j=1}^{n} K(X \wedge e_j).
$$
 (8)

for each fixed $e_i, i \in \{1, ..., n\}$ we have

$$
Ric(e_i) = \sum_{j \neq i}^{n} K(e_i \wedge e_j).
$$

The scalar curvature $\tau(p)$ at p is defined by

$$
\tau(p) = \sum_{1 \le i < j \le n} K(e_i \wedge e_j). \tag{9}
$$

In particular, for a 2-dimensional Riemannian manifold, the scalar curvature is its Gaussian curvature.

Let Π_k be a k-plane section of T_pM and X a unit vector in Π_k . If $k = n$ then $\Pi_n = T_pM$; and if $k = 2$ then Π_2 is a plane section of T_pM . We choose an orthonormal basis $\{e_1, \ldots, e_k\}$ of Π_k such that $e_1 = X$. The k-Ricci curvature of Π_k at X, denoted $\text{Ric}_{\Pi_k}(X)$, is defined by $[9]$

$$
Ric_{\Pi_k}(X) = K(e_1 \wedge e_2) + K(e_1 \wedge e_3) + \dots + K(e_1 \wedge e_k). \tag{10}
$$

Thus for each fixed $e_i, i \in \{1, \ldots, k\}$ we get

$$
Ric_{\Pi_k}(e_i) = \sum_{j \neq i}^{k} K(e_i \wedge e_j) = \sum_{j \neq i}^{k} K_{ij}.
$$
 (11)

We note that an *n*-Ricci curvature $\text{Ric}_{T_pM}(e_i)$ is the usual Ricci curvature of e_i , denoted Ric (e_i) . Thus for any orthonormal basis $\{e_1, \ldots, e_n\}$ for T_pM and for a fixed $i \in \{1, \ldots, n\}$, we have

$$
Ric_{T_pM}(e_i) \equiv \text{Ric}(e_i) = \sum_{j \neq i}^{n} K_{ij}.
$$

The scalar curvature $\tau(\Pi_k)$ of the k-plane section Π_k is given by

$$
\tau(\Pi_k) = \sum_{1 \le i < j \le k} K(e_i \wedge e_j) = \sum_{1 \le i < j \le k} K_{ij},\tag{12}
$$

where $\{e_1, ..., e_k\}$ is any orthonormal basis of the k-plane section Π_k . We note that

$$
\tau(\Pi_k) = \frac{1}{2} \sum_{i=1}^k \sum_{j \neq i}^k K(e_i \wedge e_j) = \frac{1}{2} \sum_{i=1}^k \text{Ric}_{\Pi_k}(e_i). \tag{13}
$$

Given an orthonormal basis $\{e_1, ..., e_n\}$ for T_pM , $\tau_{1\cdots k}$ will denote the scalar curvature of the k-plane section spanned by $e_1, ..., e_k$.

The scalar curvature $\tau(p)$ of M at p is identical with the scalar curvature of the tangent space T_pM of M at p, that is,

$$
\tau(p) = \tau(T_p M).
$$

Let (M, g) and (B, \widetilde{g}) be m and n dimensional Riemannian manifolds with Riemannian metrics g and \tilde{g} , respectively. A smooth map $\pi : (M, g) \to (B, \tilde{g})$ is called a Riemannian submersion if

i) π has maximal rank.

ii) The differential π_* preserves the lengths of horizontal vectors.

Now, let $\pi : (M, g) \to (B, \tilde{g})$ be a Riemannian submersion. For any $b \in B$, $\pi^{-1}(b)$ is closed r-dimensional submanifold of M. The submanifolds $\pi^{-1}(b)$ are called fibers. A vector field tangent to fibers is called vertical and a vector field orthogonal to fibers is called horizontal. If we put

$$
\mathcal{V}_p = \text{kernel}(\pi_*)\tag{14}
$$

at a point $p \in M$, then it can be obtained an integrable distribution V corresponding to the foliation of M determined by the fibres of π . The distribution \mathcal{V}_p is called vertical space at $p \in M$.

Let $\mathcal H$ be a complementary distribution of $\mathcal V$ determined by the Riemannian metric g. For any $p \in M$, the distribution $\mathcal{H}_p = (\mathcal{V}_p)^{\perp}$ is called *horizontal space* on M [\[17\]](#page-126-19). Thus, we have the following orthogonal decomposition:

$$
TM = \mathcal{V} \oplus \mathcal{H}.\tag{15}
$$

A vector field E on M is called *basic* if it is horizontal and π -related to a vector field E_* on B i.e., $\pi_* E_p = E_{*\pi(p)}$ for all $p \in M$. Furthermore, it is known that if E and F are the basic vector fields respectively π -related to E_* and F_* , one has

$$
g(E, F) = \widetilde{g}(E_*, F_*) \circ \pi. \tag{16}
$$

Let h and v are the projections of $\Gamma(TM)$ onto $\Gamma(\mathcal{H})$ and $\Gamma(\mathcal{V})$, respectively. The fundamental tensor fields of π , denoted by A and T, are defined respectively by

$$
A_E F = h \nabla_{hE} v F + v \nabla_{hE} h F, \qquad (17)
$$

$$
T_E F = h \nabla_{vE} v F + v \nabla_{vE} h F \tag{18}
$$

for any $E, F \in \Gamma(TM)$, where ∇ is the Levi-Civita connection on M.

Now, let us define the following mappings:

$$
T^{\mathcal{H}} : \Gamma(\mathcal{V}) \times \Gamma(\mathcal{V}) \rightarrow \Gamma(\mathcal{H}),
$$

\n
$$
(U, V) \rightarrow T^{\mathcal{H}}(U, V) = h \nabla_U V,
$$

\n
$$
T^{\mathcal{V}} : \Gamma(\mathcal{V}) \times \Gamma(\mathcal{H}) \rightarrow \Gamma(\mathcal{V}),
$$

\n
$$
(U, X) \rightarrow T^{\mathcal{V}}(U, X) = v \nabla_U X,
$$

and

$$
A^{\mathcal{H}} : \Gamma(\mathcal{H}) \times \Gamma(\mathcal{V}) \rightarrow \Gamma(\mathcal{H}),
$$

\n
$$
(X, U) \rightarrow A^{\mathcal{H}}(X, U) = h \nabla_X U,
$$

\n
$$
A^{\mathcal{V}} : \Gamma(\mathcal{H}) \times \Gamma(\mathcal{H}) \rightarrow \Gamma(\mathcal{V}),
$$

\n
$$
(X, Y) \rightarrow A^{\mathcal{V}}(X, Y) = v \nabla_X Y,
$$

Then, it is clear from (17) and (18) that $T^{\mathcal{H}}$ is a symmetric operator on $\Gamma(\mathcal{V}) \times \Gamma(\mathcal{V})$ and $A^{\mathcal{V}}$ is an anti-symmetric operator on $\Gamma(\mathcal{H}) \times \Gamma(\mathcal{H})$. If Γ and Γ and Γ into account in (15) , we can write

$$
\nabla_U V = T^{\mathcal{H}}(U, V) + v \nabla_U V,\tag{19}
$$

$$
\nabla_V X = h \nabla_V X + T^{\mathcal{V}}(V, X),\tag{20}
$$

$$
\nabla_X U = A^{\mathcal{H}}(X, U) + v \nabla_X U,\tag{21}
$$

$$
\nabla_X Y = h \nabla_X Y + A^{\mathcal{V}}(X, Y) \tag{22}
$$

for any $U, V \in \Gamma(\mathcal{V})$ and $X, Y \in \Gamma(\mathcal{H})$.

Let $\{U_1, \ldots, U_r, X_1, \ldots, X_n\}$ be an orthonormal basis on T_pM , where $V = \text{Span}\{U_1, \ldots, U_r\}$ and $\mathcal{H} = \text{Span}\{X_1, \ldots, X_n\}$. The mean curvature vector field $\hbar(p)$ of any fibre is defined by

$$
\mathcal{N}(p) = \frac{1}{r} \sum_{j=1}^{r} T^{\mathcal{H}}(U_j, U_j).
$$
 (23)

Note that each fiber is a minimal submanifold of M if and only if $\hbar(p) = 0$ for all $p \in M$. Furthermore, each fiber is called *totally geodesic* if both $T^{\mathcal{H}}$ and $T^{\mathcal{V}}$ vanish identically and it is called totally umbilical if

$$
T^{\mathcal{H}}(U,V) = g(U,V)\,\hbar
$$

for all $U, V \in \Gamma(V)$.

Now we recall the following Theorem [\[14\]](#page-126-20):

Theorem 1. Let π : $(M, g) \rightarrow (B, \tilde{g})$ be a Riemann submersion. Then the horizontal space H is an integrable distribution if and only if A vanishes identically.

Remark 1. As a consequence of Theorem \overline{A} , we see that both $A^{\mathcal{H}}$ and $A^{\mathcal{V}}$ are related to integrability of H , that is, they are identically zero if and only if H is integrable.

Let R, \tilde{R} and \hat{R} are the curvature tensors on M, B and be the collection of all curvature tensors on fibers $\pi^{-1}(b)$ respectively, and $\check{R}(X,Y)Z$ be the horizontal lift of $R_{\pi(b)}(\pi_{*p}X_b, \pi_{*p}Y_b)Z_b$ at any point $b \in M$ satisfying

$$
\pi_*(\check{R}(X,Y)Z) = \check{R}(\pi_*X, \pi_*Y)\pi_*Z.
$$

Then, there exist the following relations between these tensors:

$$
R(U, V, W, G) = \hat{R}(U, V, W, G) + g((T^{\mathcal{H}}(U, G), T^{\mathcal{H}}(V, W))
$$

\n
$$
-g(T^{\mathcal{H}}(V, G), T^{\mathcal{H}}(U, W)),
$$

\n
$$
R(X, Y, Z, H) = \tilde{R}(X, Y, Z, H) - 2g(A^{\mathcal{V}}(X, Y), A^{\mathcal{V}}(Z, H))
$$

\n
$$
+g(A^{\mathcal{V}}(Y, Z), A^{\mathcal{V}}(X, H)) - g(A^{\mathcal{V}}(X, Z), A^{\mathcal{V}}(Y, H)),
$$

\n
$$
R(X, V, Y, W) = g((\nabla_X T) (V, W), Y) + g((\nabla_V A) (X, Y), W)
$$

\n
$$
-g(T^{\mathcal{V}}(V, X), T^{\mathcal{V}}(W, Y))
$$

\n
$$
+g(A^{\mathcal{H}}(X, V), A^{\mathcal{H}}(Y, W)),
$$

\n(26)

for any $U, V, W, G \in \Gamma(\mathcal{V})$ and $X, Y, Z, H \in \Gamma(\mathcal{H})$. Note that the above equalities are known as Gauss–Codazzi equations for a Riemannian submersion. With the help of Gauss–Codazzi equations, we get the following relations between the sectional curvatures as follows:

$$
K(U \wedge V) = \hat{K}(U \wedge V) - ||T^{\mathcal{H}}(U, V)||^{2}
$$

+
$$
g(T^{\mathcal{H}}(U, U), T^{\mathcal{H}}(V, V)),
$$
 (27)

$$
K(X \wedge Y) = \check{K}(\check{X} \wedge \check{Y}) + 3\|A^{\mathcal{V}}(X,Y)\|^{2},
$$
\n(28)

$$
K(X \wedge V) = -g((\nabla_X T)(V, V), X) + ||T^{\mathcal{V}}(V, X)||^2 - ||A^{\mathcal{H}}(X, V)||^2,
$$
 (29)

where K, \hat{K} and \check{K} denote the sectional curvatures in M, any fiber $\pi^{-1}(b)$ and the horizontal distribution H , respectively. The scalar curvatures of the vertical and horizontal spaces at a point $p \in M$ are given respectively by

$$
\hat{\tau}(p) = \sum_{1 \le i < j \le r} \hat{K}(U_i, U_j) \tag{30}
$$

and

$$
\check{\tau}(p) = \sum_{1 \le i < j \le n} \check{K}(X_i, X_j). \tag{31}
$$

Now, we recall the following definition of [\[5\]](#page-126-21).

Definition 1. Let π : $(M, g) \rightarrow (B, \widetilde{g})$ be a Riemann submersion and X be a horizontal vector field on π . Then, horizontal divergence of X is defined by

$$
div_{\mathcal{H}}(X) = \sum_{i=1}^{n} g(\nabla_{X_i} X, X_i).
$$
 (32)

Lemma 1. [\[14\]](#page-126-20) Let π : $(M, g) \rightarrow (B, \tilde{g})$ be a Riemann submersion and $\{U_1,\ldots,U_r\}$ be any orthonormal basis of $\Gamma(V)$. For any $E \in \Gamma(TM)$ and $X \in$ $\Gamma(\mathcal{H})$, we have

$$
g\left(\nabla_E \mathcal{N}, X\right) = \frac{1}{r} \sum_{j=1}^r g\left(\left(\nabla_E T\right) \left(U_j, U_j\right), X\right). \tag{33}
$$

As a consequence of Lemma $\overline{\mathbb{1}}$, we obtain that

$$
\operatorname{div}_{\mathcal{H}}(\mathcal{N}) = \frac{1}{r} \sum_{i=1}^{n} \sum_{j=1}^{r} g\left(\left(\nabla_{X_i} T\right) \left(U_j, U_j\right), X_i\right). \tag{34}
$$

3. An Optimal Inequality for Riemannian Submersions

We begin this section with the following algebraic lemma:

Lemma 2. If $n > k \geq 2$ and a_1, \ldots, a_n, a are real numbers such that

$$
\left(\sum_{i=1}^{n} a_i\right)^2 = (n - k + 1) \left(\sum_{i=1}^{n} a_i^2 + a\right),\tag{35}
$$

then

$$
2\sum_{1\leq i
$$

with equality holding if and only if

$$
a_1 + a_2 + \dots + a_k = a_{k+1} = \dots = a_n.
$$

Proof. By the Cauchy-Schwartz inequality, we have

$$
\left(\sum_{i=1}^{n} a_i\right)^2 \le (n-k+1)\left((a_1+a_2+\cdots+a_k)^2 + a_{k+1}^2 + \cdots + a_n^2\right). \tag{36}
$$

From (35) and (36) , we get

$$
\sum_{i=1}^{n} a_i^2 + a \le (a_1 + a_2 + \dots + a_k)^2 + a_{k+1}^2 + \dots + a_n^2.
$$

The above equation is equivalent to

$$
2\sum_{1\leq i
$$

The equality holds if and only if $a_1 + a_2 + \cdots + a_k = a_{k+1} = \cdots = a_n$.

Let π : $(M, g) \rightarrow (B, \tilde{g})$ be a Riemannian submersion between Riemannian manifolds (M, g) and (B, \tilde{g}) . Suppose $\{U_1, \ldots, U_r, X_1, \ldots, X_n\}$ be an orthonormal basis on T_pM , where $\mathcal{V} = \text{Span}\{U_1, \ldots, U_r\}$ and $\mathcal{H} = \text{Span}\{X_1, \ldots, X_n\}$. Then, we have

$$
||T^{\mathcal{H}}||^2 = \sum_{i,j=1}^r g\left(T^{\mathcal{H}}(U_i, U_j), T^{\mathcal{H}}(U_i, U_j)\right),
$$
\n(37)

$$
||T^{\mathcal{V}}||^2 = \sum_{i=1}^r \sum_{j=1}^n g\left(T^{\mathcal{V}}(U_i, X_j), T^{\mathcal{V}}(U_i, X_j)\right),\tag{38}
$$

$$
||A^{\mathcal{H}}||^2 = \sum_{i=1}^{r} \sum_{j=1}^{n} g\left(A^{\mathcal{H}}(X_j, U_i), A^{\mathcal{H}}(X_j, U_i)\right),
$$
 (39)

$$
||A^{\mathcal{V}}||^2 = \sum_{i,j=1}^n g\left(A^{\mathcal{V}}(X_i, X_j), A^{\mathcal{V}}(X_i, X_j)\right).
$$
 (40)

Putting $(27) - (29)$ $(27) - (29)$, (34) and $(37) - (40)$ $(37) - (40)$ in

$$
\tau(p) = \sum_{1 \le i < j \le n} [K(U_i, U_j) + K(X_i, U_j) + K(X_i, X_j)],
$$

we obtain the following lemma:

Lemma 3. Let (M, g) and (B, \tilde{g}) be a Riemannian manifolds admitting a Riemannian submersion $\pi : (M, g) \to (B, \tilde{g})$. For any point $p \in M$, we have

$$
2\tau(p) = 2\hat{\tau}(p) + 2\check{\tau}(p) + r^2 \|\hbar(p)\|^2 - \|T^{\mathcal{H}}\|^2 + 3\left\|A^{\mathcal{V}}\right\|^2
$$

$$
-r \operatorname{div}_{\mathcal{H}}(\hbar(p)) + \|T^{\mathcal{V}}\|^2 - \|A^{\mathcal{H}}\|^2. \tag{41}
$$

Now, we are going to give an optimal inequality involving the δ -curvature for Riemannian manifolds admitting a Riemannian submersion.

Theorem 2. Let π : $(M, g) \rightarrow (B, \tilde{g})$ be a Riemannian submersion. Then, for each point $p \in M$ and each k-plane section $L_k \subset V_p$ $(r > k \geq 2)$, we have

$$
\delta(k) \leq \hat{\tau}(p) - \hat{\tau}(L_k) + \check{\tau}(p) + \frac{r^2(r-k)}{2(r-k+1)} \|\hbar\|^2 - \frac{r}{2} \operatorname{div}_{\mathcal{H}}(\hbar(p)) + \frac{3}{2} \|A^{\mathcal{V}}\|^2 + \frac{1}{2} \|T^{\mathcal{V}}\|^2.
$$
\n(42)

The equality of $\sqrt{42}$ holds at $p \in M$ if and only if $A^{\mathcal{H}}$ vanishes identically and the shape operators S_{X_1}, \ldots, S_{X_n} of \mathcal{V}_p take forms as follows:

$$
S_{X_1} = \begin{pmatrix} T_{11}^1 & 0 & \cdots & 0 \\ 0 & T_{22}^1 & \cdots & 0 & & 0 \\ \vdots & \vdots & \ddots & \vdots & & 0 \\ 0 & 0 & \cdots & T_{kk}^1 & & 0 \\ 0 & 0 & \cdots & T_{kk}^1 & & 0 \end{pmatrix}, \qquad (43)
$$

$$
S_{X_s} = \begin{pmatrix} T_{11}^s & T_{12}^s & \cdots & T_{1k}^s \\ T_{12}^s & T_{22}^s & \cdots & T_{2k}^s \\ \vdots & \vdots & \ddots & \vdots & 0 \\ T_{1k}^s & T_{2k}^s & \cdots & -\sum_{i=1}^{k-1} T_{ii}^s & & 0 \\ 0 & 0 & 0_{n-k} & & 0 \end{pmatrix}, \qquad s \in \{2, \ldots, n\}. \qquad (44)
$$

Proof. Let L_k be a k-plane section of V_p . We choose an orthonormal basis $\{U_1, \ldots, U_r, X_1, \ldots, X_n\}$ on T_pM such that $\mathcal{V} = \text{Span}\{U_1, \ldots, U_r\}$ and $\mathcal{H} = \text{Span}\{X_1, \ldots, X_n\}.$ We write

$$
T_{ij}^s = g(T^{\mathcal{H}}(U_i, U_j), X_s)
$$
\n⁽⁴⁵⁾

for any $i, j \in \{1, ..., r\}$ and $s \in \{1, ..., n\}$. Suppose that the mean curvature vector $\hbar(p)$ is in the direction of X_1 and $X_1, ..., X_n$ diagonalize the shape operator S_{X_1} . If we put

$$
\eta = 2\tau(p) - 2\tilde{\tau}(p) - 2\tilde{\tau}(p) - \frac{r^2(r - k)}{(r - k + 1)} ||\hbar||^2 + r \ div_{\mathcal{H}}(\hbar(p))
$$

$$
-3 ||A^{\mathcal{V}}||^2 - ||T^{\mathcal{V}}||^2 + ||A^{\mathcal{H}}||^2
$$
 (46)

in (41) , it follows that

$$
r^{2}||\hbar||^{2} = (n - k + 1)(\eta + ||T^{\mathcal{H}}||^{2}).
$$
\n(47)

The equation (47) is equivalent to

$$
\left(\sum_{i=1}^{r} T_{ii}^{1}\right)^{2} = (n - k + 1) \left(\eta + \sum_{i=1}^{r} \left(T_{ii}^{1}\right)^{2} + \sum_{s=2}^{n} \sum_{i,j=1}^{r} \left(T_{ij}^{s}\right)^{2}\right). \tag{48}
$$

Applying Lemma $\boxed{2}$ to equation $\boxed{48}$, we get

$$
2\sum_{1 \le i < j \le k} T_{ii}^{n+1} T_{jj}^{n+1} \ge \eta + \sum_{s=2}^{n} \sum_{i,j=1}^{r} (T_{ij}^s)^2. \tag{49}
$$

On the other hand, we have from (41) that

$$
\tau(L_k) = \hat{\tau}(L_k) + \sum_{1 \le i < j \le k} T_{ii}^1 T_{jj}^1 + \sum_{s=2}^n \sum_{1 \le i < j \le k} \left(T_{ii}^s T_{jj}^s - \left(T_{ij}^s \right)^2 \right). \tag{50}
$$

From (49) and (50) , we get

$$
\tau(L_k) \geq \hat{\tau}(L_k) + \frac{1}{2}\eta + \sum_{s=2}^n \sum_{j>k} \{ (T_{1j}^s)^2 + (T_{2j}^s)^2 + \dots + (T_{kj}^s)^2 \} + \frac{1}{2} \sum_{s=2}^n (T_{11}^s + T_{22}^s + \dots + T_{kk}^s)^2 + \frac{1}{2} \sum_{s=2}^n \sum_{i,j>k} (T_{ij}^s)^2.
$$
 (51)

In view of (51) , we see that

$$
\tau(\Pi_k) \ge \widetilde{\tau}(\Pi_k) + \frac{1}{2}\eta. \tag{52}
$$

From (47) and (52) , we obtain (42) .

If the equality case of (42) holds, then we have $A^{\mathcal{H}}$ vanishes identically and

$$
\begin{cases}\nT_{1j}^1 = T_{2j}^1 = T_{kj}^1 = 0, & j = k+1, \dots, r, \\
T_{ij}^s = 0, & i, j = k+1, \dots, r, \\
T_{11}^r + T_{22}^r + \dots + T_{kk}^r = 0\n\end{cases}
$$
\n(53)

for $s = 2, \ldots, n$ $s = 2, \ldots, n$ $s = 2, \ldots, n$. Applying Lemma 2, we also have

$$
T_{11}^1 + T_{22}^1 + \dots + T_{kk}^1 = T_{ll}^1, \qquad l = k + 1, \dots, n. \tag{54}
$$

Thus, with respect to a suitable orthonormal basis $\{X_1, \ldots, X_m\}$ on \mathcal{H}_p , the shape operator of V_p becomes of the form given by (43) and (44) . The proof of the converse part is straightforward. □

In particular case of $k = 2$, we have the following:

Corollary 1. Let π : $(M, g) \rightarrow (B, \tilde{g})$ be a Riemannian submersion. Then, for each point $p \in M$ and each plane section $L \subset V_p$, we have

$$
\delta(2) \leq \hat{\tau}(p) - \hat{K}(L) + \check{\tau}(p) + \frac{r^2(r-2)}{2(r-1)} \|\hbar\|^2 - \frac{r}{2} \operatorname{div}_{\mathcal{H}}(\hbar) + \frac{3}{2} \|A^{\mathcal{V}}\|^2 + \frac{1}{2} \|T^{\mathcal{V}}\|^2.
$$
 (55)

The equality of (55) holds at $p \in M$ if and only if $A^{\mathcal{H}}$ vanishes identically and the shape operators S_{X_1}, \ldots, S_{X_n} of \mathcal{V}_p take forms

$$
S_{X_1} = \begin{pmatrix} a & 0 & 0 \\ 0 & b & 0 \\ 0 & 0 & (a+b)I_{r-2} \end{pmatrix}, \tag{56}
$$

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$$
S_{X_s} = \begin{pmatrix} c_s & d_s & 0 \\ d_s & -c_s & 0 \\ 0 & 0 & 0_{r-2} \end{pmatrix}, \qquad s \in \{2, ..., n\}.
$$
 (57)

In particular case of $k = r - 1$, we have the following

Corollary 2. Let π : $(M, g) \rightarrow (B, \tilde{g})$ be a Riemannian submersion. For each vertical unit vector U, we have

$$
\operatorname{Ric}_{\mathcal{V}}(U) \le \operatorname{Ric}(U) + \check{\tau}(p) + \frac{r^2}{4} \|\hbar\|^2 - \frac{r}{2} \operatorname{div}_{\mathcal{H}}(\hbar) + \frac{3}{2} \|A^{\mathcal{V}}\|^2 + \frac{1}{2} \|T^{\mathcal{V}}\|^2. \tag{58}
$$

The equality case of (58) holds for all unit vectors $U \in V_p$ if and only if $A^{\mathcal{H}}$ vanishes identically and we have either

- (i) if $r = 2$, π has totally umbilical fibers at $p \in M$,
- (i) if $r \neq 2$, π has totally geodesic fibers at $p \in M$.

Proof. Let L_{r-1} be a $(r-1)$ -plane section of \mathcal{V}_p . We get from Theorem [2](#page-119-3) that

$$
\delta(r-1) \leq \hat{\tau}(p) - \hat{\tau}(L_{r-1}) + \check{\tau}(p) + \frac{r^2}{4} ||\hbar||^2 - \frac{r}{2} \operatorname{div}_{\mathcal{H}}(\hbar) + \frac{3}{2} ||A^{\mathcal{V}}||^2 + \frac{1}{2} ||T^{\mathcal{V}}||^2.
$$
 (59)

Now, let U be a unit vertical vector field such that $U = U_r$. By a straightforward computation, we obtain [\(58\)](#page-122-0).

The equality of (59) holds if and only if the forms of shape operators S_{X_s} , $s = 1, \ldots, n$, become

$$
S_{X_1} = \begin{pmatrix} T_{11}^1 & 0 & \cdots & 0 & 0 \\ 0 & T_{22}^1 & \cdots & 0 & 0 \\ \vdots & \vdots & \ddots & \vdots & \vdots \\ 0 & 0 & \cdots & T_{(r-1)(r-1)}^1 & 0 \\ 0 & 0 & \cdots & 0 & \left(\sum_{i=1}^{r-1} T_{ii}^1\right) \end{pmatrix}, \qquad (60)
$$

$$
S_{X_s} = \begin{pmatrix} T_{11}^s & T_{12}^s & \cdots & T_{1(r-1)}^r & 0 \\ T_{12}^s & T_{22}^s & \cdots & T_{2(r-1)}^s & 0 \\ \vdots & \vdots & \ddots & \vdots & \vdots \\ T_{1(r-1)}^s & T_{2(r-1)}^s & \cdots & -\sum_{i=1}^{r-2} T_{ii}^s & 0 \\ 0 & 0 & \cdots & 0 & 0 \end{pmatrix}, \quad r \in \{2, \ldots, n\}. \tag{61}
$$

From $\boxed{60}$ and $\boxed{61}$, we see that the equality in $\boxed{58}$ is valid for a unit vertical vector field $U = U_r$ if and only if

$$
\begin{cases}\nT_{rr}^s = T_{11}^s + T_{22}^s + \dots + T_{(r-1)(r-1)}^s \\
T_{1r}^s = T_{2r}^s = \dots = T_{(r-1)r}^s = 0.\n\end{cases} \tag{62}
$$

for $s \in \{1, ..., n\}$.

Assuming the equality case of [\(58\)](#page-122-0) holds for all unit vertical vector fields, in view of (62) , for each $s \in \{1, \ldots, n\}$, we have

$$
\begin{cases}\n2T_{ii}^s = T_{11}^s + T_{22}^s + \dots + T_{rr}^s, \\
T_{ij}^s = 0, \quad i \neq j\n\end{cases} \tag{63}
$$

for all $i \in \{1, ..., r\}$ and $s \in \{1, ..., n\}$. Thus, we have two cases, namely either $r = 2$ or $r \neq 2$. In the first case we see that π has totally umbilical fibers, while in the second case π has totally geodesic fibers. The proof of converse part is straightforward. \Box

Remark 2. We note that (58) was also proved in $\overline{15}$ (see Theorem 4.1 in $\overline{15}$). In Theorem $\sqrt{2}$, we gave a new proof for this inequality.

4. Main Conclusions

In this section, we shall present a solution way with the help of differential geometry tools for the following natural problem:

"Which conditions should provide to the horizontal divergence or the convergence receives to the maximum value or minimum value?"

To obtain minimum or maximum values of the vertical motion (or horizontal divergence) it can be considered a Riemannian submersion on E^3 to E^2 . Moreover, we can regard to different Riemannian submersions such as a Riemannian submersion on a three dimensional Riemannian manifold to two dimensional Riemannian manifold as

$$
\pi : M^3 \to N^2. \tag{64}
$$

It can also be considered globally in high dimensional Riemannian manifolds with taking a Riemannian submersion on m-dimensional Riemannian manifold to ndimensional Riemannian manifold.

Taking into account of the continuity equation and (42) , (55) and (58) inequalities, we get some result dealing minimum or maximum values of vertical motion for a manifold admitting a Riemannain submersion.

As a consequence of $\left(\frac{42}{1}\right)$, we obtain the following:

Corollary 3. Let $\pi : E^{n+r} \to E^n$ be a Riemannian submersion. Then we have

$$
\frac{r}{2}\omega(p) \ge \delta(k) - \frac{r^2(r-k)}{2(r-k+1)}\|\hbar\|^2 - \frac{3}{2}\|A^{\mathcal{V}}\|^2 - \frac{1}{2}\|T^{\mathcal{V}}\|^2. \tag{65}
$$

The vertical motion at a point p takes the minimum value if and only if $A^{\mathcal{H}}$ vanishes identically and the matrixes of shape operators of the vertical space of M take the form as (43) and (44) .

As a consequence of (55) , we obtain the followings:

Corollary 4. Let π : $E^{n+r} \to E^n$ be a Riemannian submersion with integrable horizontal distribution. Then we have

$$
\frac{r}{2}\omega(p) \ge \delta(2) - \frac{r^2(r-2)}{2(r-1)}\|\hbar\|^2 - \frac{1}{2}\|T^{\mathcal{V}}\|^2. \tag{66}
$$

The vertical motion takes the minimum value if and only if the matrixes of shape operators S_{x_1}, \ldots, S_{x_n} of the vertical space of M take the form as [\(56\)](#page-121-4) and [\(57\)](#page-122-5).

Corollary 5. Let π : $E^{n+r} \to E^n$ be a Riemannian submersion with totally geodesic leaves and integrable horizontal distribution. Then we have

$$
\frac{r}{2}\omega(p) = \delta(2). \tag{67}
$$

From [\(58\)](#page-122-0), we get the followings:

Corollary 6. Let $\pi : E^{n+r} \to E^n$ be a Riemannian submersion. For each vertical unit vector U, we have

$$
\frac{r}{2}\omega(p) \ge \text{Ric}_{\mathcal{V}}\left(U\right) - \frac{r^2}{4} \|\hbar\|^2 - \frac{3}{2} \|A^{\mathcal{V}}\|^2 - \frac{1}{2} \|T^{\mathcal{V}}\|^2. \tag{68}
$$

The equality case of $\boxed{68}$ holds for all unit vectors $U \in \mathcal{V}_p$ if and only if $A^{\mathcal{H}}$ vanishes identically and we have either

- (i) if $r = 2$, π has totally umbilical fibers at $p \in M$,
- (ii) if $r \neq 2$, π has totally geodesic fibers at $p \in M$.

Corollary 7. Let $\pi : \mathbb{E}^{n+r} \to \mathbb{E}^n$ be a Riemannian submersion with totally geodesic fibers. For each vertical unit vector U, we have

$$
\frac{r}{2}\omega(p) = \text{Ric}_{\mathcal{V}}\left(U\right) - \frac{3}{2}||A^{\mathcal{V}}||^2.
$$
\n
$$
(69)
$$

Now we shall mention some examples:

Example 1. Consider the mapping $\pi : E^5 \to E^2$ which is defined by

$$
\pi(x_1, x_2, x_3, x_4, x_5) = \left(\frac{1}{\sqrt{2}}(x_1 + x_2), \frac{1}{\sqrt{2}}(x_3 + x_4)\right).
$$

Then, it is clear that π is a Riemannian submersion and the Jacobian of π is equal to

$$
\left(\begin{array}{cccc} \frac{1}{\sqrt{2}} & \frac{1}{\sqrt{2}} & 0 & 0 & 0 \\ 0 & 0 & \frac{1}{\sqrt{2}} & \frac{1}{\sqrt{2}} & 0 \end{array}\right).
$$

The horizontal space and the vertical space are given by

$$
\mathcal{H} = Span\{X_1 = \frac{1}{\sqrt{2}}\frac{\partial}{\partial x_1} + \frac{1}{\sqrt{2}}\frac{\partial}{\partial x_2}, X_2 = \frac{1}{\sqrt{2}}\frac{\partial}{\partial x_3} + \frac{1}{\sqrt{2}}\frac{\partial}{\partial x_4}\}
$$

and

$$
\mathcal{V} = Span\{U_1 = -\frac{1}{\sqrt{2}}\frac{\partial}{\partial x_1} + \frac{1}{\sqrt{2}}\frac{\partial}{\partial x_1}, U_2 = -\frac{1}{\sqrt{2}}\frac{\partial}{\partial x_3} + \frac{1}{\sqrt{2}}\frac{\partial}{\partial x_4}, U_3 = \frac{\partial}{\partial x_5}\},\
$$

respectively. By a straightforward computation, we get the tensor fields A, T, Ric $_{\mathcal{V}}$ vanish and $\omega(p) = div_{\mathcal{H}}(\hbar) = 0$ from $\ddot{\mathcal{B}}$. Therefore, π is a trivial example satisfying Corollary [3](#page-123-1)- Corollary [7](#page-124-1).

Example 2. *(Example 5.1 in* $\overline{15}$ *)*

Let us consider the Remannian submesion $\pi : M \to \mathbb{E}^3$ defined by

 $\pi(x_1, x_2, x_3, x_4, x_5) = (x_1 \cos x_3 + x_2 \sin x_3, x_4, x_5),$

where M is a non-flat submanifold of E^5 such that $\cot x_3 = \frac{x_1}{x_2}$, $x_2 \neq 0$ and $x_3 \in$ $(0, \frac{\pi}{2})$. Here, the horizontal space and the vertical space of M are given by

$$
\mathcal{H} = Span\{X_1 = \sin x_3 \frac{\partial}{\partial x_1} + \cos x_3 \frac{\partial}{\partial x_2}, X_2 = \frac{\partial}{\partial x_4}, X_3 = \frac{\partial}{\partial x_5}\}
$$

and

$$
\mathcal{V} = Span\{U_1 = -\cos x_3 \frac{\partial}{\partial x_1} + \sin x_3 \frac{\partial}{\partial x_2}, U_2 = \frac{\partial}{\partial x_3}\},\
$$

respectively. By straightforward computations, we have $T^{\mathcal{V}}(U_2, X_1) = -U_1$, $T^{\mathcal{H}}(U_1,U_2)=X_1$ and the other components of operators $T^{\mathcal{H}}, T^{\mathcal{V}}, A^{\mathcal{H}}, A^{\mathcal{V}}$ vanish *identically. Moreover, we have* $\text{Ric}(U_1) = 1$, $\text{Ric}_{\mathcal{V}}(U_1) = \text{Ric}_{\mathcal{V}}(U_2) = 0$ and $\omega(p) =$ 0 from $\sqrt{3}$. Considering these facts, we obtain the left hand side of $\sqrt{68}$ $\sqrt{68}$ $\sqrt{68}$ is equal to 0 and the right hand side of (68) is equal to -1 for $U = U_1$. This inequality also satisfies for $U = U_2$. This shows that the correctness of [\(68\)](#page-124-0) and π is an example of Corollary [6](#page-124-2).

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PARITY OF AN ODD DOMINATING SET

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ABSTRACT. For a simple graph G with vertex set $V(G) = \{v_1, ..., v_n\}$, we define the closed neighborhood set of a vertex u as $N[u] = \{v \in V(G) \mid v$ is adjacent to u or $v = u$ } and the closed neighborhood matrix $N(G)$ as the matrix obtained by setting to 1 all the diagonal entries of the adjacency matrix of G. We say a set S is odd dominating if $N[u] \cap S$ is odd for all $u \in V(G)$. We prove that the parity of an odd dominating set of G is equal to the parity of the rank of G , where the rank of G is defined as the dimension of the column space of $N(G)$. Using this result we prove several corollaries in one of which we obtain a general formula for the nullity of the join of graphs.

1. Introduction

Let $N[u]$ denote the *closed neighborhood set* of a vertex u in a simple graph G , i.e.;

$$
N[u] = \{ v \in V(G) \mid v \text{ is adjacent to } u \text{ or } v = u \}.
$$

Then, we say a subset S of vertices is *odd (even) dominating* if $N[u] \cap S$ is odd (even) for all $u \in V(G)$. In general, for an arbitrary subset C of vertices, we say a set S is a C-parity set if $N[u] \cap S$ is odd for all $u \in C$ and even otherwise $[2]$. If there is a C -parity set for a given set C , we say that C is *solvable*. If there exists a C-parity set for every set C of vertices in a graph G , then we say G is always solvable.

Let *n* be the order of G, $V(G) = \{v_1, ..., v_n\}$ and W be a subset of $V(G)$. The column vector $\mathbf{x}_W = (x_1, ..., x_n)^t$, which is defined as $x_i = 1$ if $v_i \in W$ and $x_i = 0$ otherwise, is called the characteristic vector of W. The closed neighbourhood matrix $N = N(G)$ of a graph G is obtained by setting to 1 all the diagonal entries of the adjacency matrix of G. Equivalently, $N(G)$ is the matrix whose *i*th column

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is equal to $\mathbf{x}_{N[v_i]}$. It is easy to observe that S is a C-parity set if and only if

$$
N(G)\mathbf{x}_S = \mathbf{x}_C \tag{1}
$$

over the field \mathbb{Z}_2 [\[9\]](#page-133-0), [\[10\]](#page-133-1).

Let us denote the vectors whose components are all 0 and all 1 by 0 and 1 , respectively. Then the following are equivalent. $(a1)$ S is an odd dominating set, (a2) S is a $V(G)$ -parity set, (a3) $N(G)\mathbf{x}_S = 1$. Similarly, (b1) S is an even dominating set, (b2) S is a Ø-parity set, (b3) $N(G)\mathbf{x}_{S} = \mathbf{0}$, are equivalent statements. Note that every graph has an even dominating set, which is \emptyset . On the other hand, it is proved by Sutner that every graph has an odd dominating set as well $\boxed{9}$ (see also $[6], [7], [8]$ $[6], [7], [8]$ $[6], [7], [8]$ $[6], [7], [8]$ $[6], [7], [8]$.

Let $Ker(N)$ and $Col(N)$ denote the kernel and column space of N, respectively. Let $\nu(G) := dim(Ker(N(G))$ and $\rho(G) := dim(Col(N(G))$. We call $\nu(G)$, the nullity of G (Amin et al. $\boxed{3}$ call it the parity dimension of G) and $\rho(G)$, the rank of G. We have $\nu(G) + \rho(G) = n$ by the rank nullity theorem.

From the matrix equation $\left| \mathbf{I} \right|$, we see that G is always solvable if and only if $\nu(G) = 0$. Moreover, $\nu(G) > 0$ if and only if G has a nonempty even dominating set.

We write $pr(a)$ to denote the parity function of a number a, i.e.; $pr(a) = 0$ if a is even and $pr(a) = 1$ if a is odd. In the case where A is a matrix, $pr(A)$ is the parity function of the sum of its entries. For a set S , we write $pr(S)$ to denote the parity function of the cardinality of S and say the parity of S instead of the parity of the cardinality of S. Note that $pr(S) = pr(\mathbf{x}_S)$. It was first noticed by Amin et al. $\left[\right]$, Lemma 3, and follows immediately from Sutner's theorem, that for a given graph, the parity of all odd dominating sets are the same. Hence, the value of $pr(S)$, where S is an odd dominating set of a graph is independent of the particular odd dominating set S taken into account.

Our main result Theorem \prod states that the parity of an odd dominating set is equal to the parity of the rank of the graph.

2. Main Result

Lemma 1. Let A be a $n \times n$, symmetric, invertible matrix over the field \mathbb{Z}_2 with diagonal entries equal to 1. Then $pr(A^{-1}) = pr(A) = pr(n)$.

Proof. In the proof, all algebraic operations are considered over the field \mathbb{Z}_2 . First of all, note that since A is a symmetric matrix with nonzero diagonal entries, we have

$$
pr(A) = \sum_{i,j} A_{ij} = \sum_i A_{ii} = \sum_i 1 = pr(n).
$$

Similarly,

$$
pr(A^{-1}) = \sum_{i} (A^{-1})_{ii}.
$$

On the other hand,

$$
pr(n) = Tr(I) = Tr(AA^{-1})
$$

=
$$
\sum_{i,j} A_{ij} (A^{-1})_{ij}
$$

=
$$
\sum_{i} A_{ii} (A^{-1})_{ii}
$$

=
$$
\sum_{i} (A^{-1})_{ii}.
$$

We call a vertex a *null vertex* of a graph G if it belongs to an even dominating set of G . Since the set of all characteristic vectors for even dominating sets of G is a subspace of the vector space of all binary *n*-tuples, if v is a null vertex of G , then precisely half of the even dominating sets of G contain v .

Lemma 2. Let G be a graph and v be a null vertex of G. Then there exists an odd dominating set of G which does not contain v.

Proof. Let R be an even dominating set containing v and S_1 be an odd dominating set of G. Assume S_1 contains v, otherwise we are done. Let S_2 be the symmetric difference of S_1 and R. Clearly S_2 is an odd dominating set which does not contain $v.$

Let $G - v$ denote the graph obtained by removing a vertex v and all its incident edges from a graph G. The number $nd(v) := \nu(G - v) - \nu(G)$ is called the null difference number. It turns out that $nd(v)$ can be either -1 , 0, or 1. Moreover, Ballard et al. proved the following lemma in $\lceil \sqrt{5} \rceil$, Proposition 2.4.

Lemma 3 ($\overline{5}$). Let v be a vertex of a graph G. Then v is a null vertex if and only if $nd(v) = -1$.

Now we are ready to state our main result.

Theorem 1. Let G be a graph and S be an odd dominating set of G. Then $pr(S)$ = $pr(\rho(G))$. Equivalently, $pr(V(G)\backslash S) = pr(\nu(G))$.

Proof. We prove the claim by applying induction on the nullity of the graph. Let n be the order of G. In the case where $\nu(G) = 0$, there exists a unique odd dominating set S such that $N \mathbf{x}_S = 1$. Note that N satisfies the conditions of Lemma [1.](#page-129-1) Hence, together with the rank nullity theorem, we have

$$
pr(S) = pr(\mathbf{x}_S) = pr(N^{-1}\mathbf{1}) = pr(N^{-1}) = pr(N) = pr(n) = pr(\rho(G)).
$$

Now assume that $\nu(G) > 0$ and the claim holds true for all graphs with nullity less than $\nu(G)$. Since $\nu(G)$ is nonzero, there exists a non-empty even dominating

□

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set. Hence, there exists a null vertex v of G. By Lemma $\overline{2}$, there is an odd dominating set S of G which does not contain v. Since S does not contain v, it is also an odd dominating set of the graph $G-v$. Moreover, by Lemma [3,](#page-130-2) $nd(v) = -1$. Hence, $\nu(G - v) = \nu(G) + nd(v) = \nu(G) - 1 < \nu(G)$. By the induction hypothesis $pr(S) = pr(\rho(G-v))$. On the other hand, using the rank nullity theorem we obtain $\rho(G - v) = n - 1 - \nu(G - v) = n - 1 - \nu(G) + 1 = n - \nu(G) = \rho(G)$. We complete the proof by noting that all odd dominating sets in G have the same parity. \Box

3. Some Corollaries

Corollary 1. Let G be an always solvable graph of order n. Then the odd dominating set of G has odd (even) cardinality if n is odd (even).

Note that if every vertex of a graph G has even degree, then $V(G)$ itself is an odd dominating set. This, together with Theorem \prod gives the following.

Corollary 2. If every vertex of a graph G has even degree, then $\nu(G)$ is even.

Corollary 3. If the number of even degree vertices of a tree T is at most one, then every odd dominating set of T has odd cardinality.

Proof. Let n be the order of T. By $\lceil \sqrt{3} \rceil$, Theorem 3 if every vertex of T has odd degree, then $\nu(T) = 1$. By the handshaking lemma, n must be even, hence $\rho(T)$ is odd. By $\lceil 3 \rceil$, Theorem 4, if exactly one vertex of T has even degree, then $\nu(T) = 0$. Since n must be odd, $\rho(T)$ is also odd. Hence in either case, every odd dominating set has odd cardinality by Theorem $\boxed{1}$.

Corollary 4. Every odd dominating set of a graph G has an odd (even) number of vertices of odd degree if and only if $\nu(G)$ is odd (even). In particular, the odd dominating set of an always solvable graph has an even number of odd degree vertices.

Proof. Observe that for any subsets A, B of $V(G)$, $pr(A \cap B) = \mathbf{x}_A^t \mathbf{x}_B$. In particular, $pr(A) = \mathbf{x}_A^t \mathbf{1}$. Let A^c be the complement of A in $V(G)$. Then we have $\mathbf{x}_{A^c} = \mathbf{x}_A + \mathbf{1}$. Now let S be an odd dominating set of G and D be the set of vertices with odd degree. Observe that $N\mathbf{1} = \mathbf{x}_{D^c}$. Therefore $N\mathbf{x}_{S^c} = N(\mathbf{x}_S + \mathbf{1}) = \mathbf{1} + \mathbf{x}_{D^c} = \mathbf{x}_D$. Then, $pr(D \cap S) = \mathbf{x}_D^t \mathbf{x}_S = (N \mathbf{x}_{S^c})^t \mathbf{x}_{S} = \mathbf{x}_{S^c}^t N \mathbf{x}_S = \mathbf{x}_{S^c}^t \mathbf{1} = pr(S^c)$. On the other hand, $pr(S^c) = pr(\nu(G))$ by Theorem **1.** Hence, the result follows. \Box

We define the join $G_1 \oplus ... \oplus G_m$ of m pairwise disjoint graphs $G_1, ..., G_m$ as follows. We take the vertex set as $V(G_1 \oplus ... \oplus G_m) = \bigcup_{i=1}^m V(G_i)$ and the edge set as $E(G_1 \oplus ... \oplus G_m) = \bigcup_{i=1}^m E(G_i) \cup \{(u,v) \mid u \in V(G_k), v \in V(G_l) \mid k, l \in$ $\{1, ..., m\}$ such that $k \neq l\}$. Then Amin et al. prove the following proposition in $[4,$ Corollary 6.

Proposition 1 ($\overline{4}$). $\nu(G_1 \oplus G_2) = \nu(G_1) + \nu(G_2)$ if either G_1 or G_2 has an odd dominating set of even cardinality, and $\nu(G_1 \oplus G_2) = \nu(G_1) + \nu(G_2) + 1$, otherwise.

Together with Theorem $\overline{\mathbb{1}}$, the above proposition implies the following.

$$
\nu(G_1 \oplus G_2) = \nu(G_1) + \nu(G_2) + pr(\rho(G_1)\rho(G_2)).
$$
\n(2)

Equivalently,

$$
\rho(G_1 \oplus G_2) = \rho(G_1) + \rho(G_2) - pr(\rho(G_1)\rho(G_2)).
$$
\n(3)

Equivalence of $\left(2\right)$ and $\left(3\right)$ follows from the rank nullity theorem.

Expressing the nullity/rank of $G_1 \oplus G_2$ as a single formula involving nullities/ranks of G_1 and G_2 as above enables us to extend this result and to write a formula for the nullity/rank of the join of arbitrary number of graphs as follows.

Proposition 2. Let $\{G_1, ..., G_m\}$ be a collection of pairwise disjoint graphs. Let j be the number of graphs in $\{G_1, ..., G_m\}$ with odd rank. Then

$$
\nu(G_1 \oplus \ldots \oplus G_m) = \begin{cases} \sum_{i=1}^m \nu(G_i) & \text{if } j = 0 \\ \sum_{i=1}^m \nu(G_i) + j - 1 & \text{otherwise} \end{cases} \tag{4}
$$

Equivalently,

$$
\rho(G_1 \oplus \ldots \oplus G_m) = \begin{cases} \sum_{i=1}^m \rho(G_i) & \text{if } j = 0 \\ \sum_{i=1}^m \rho(G_i) - j + 1 & \text{otherwise} \end{cases} \tag{5}
$$

Proof. We prove $\overline{5}$, then $\overline{4}$ follows from the rank nullity theorem. If $j = 0$, then all graphs have even rank and the result follows applying $\boxed{3}$ successively. Now let $j \neq 0$. Without loss of generality, we can assume that the first j graphs have odd rank. Then, by $\langle 3\rangle$, $\rho(G_1 \oplus G_2) = \rho(G_1) + \rho(G_2) - 1$, which is odd. Hence, $\rho(G_1 \oplus G_2 \oplus G_3) = \rho(G_1) + \rho(G_2) - 1 + \rho(G_3) - 1 = \rho(G_1) + \rho(G_2) + \rho(G_3) - 2$, which is odd, and so on, yielding $\rho(G_1 \oplus G_2 \oplus \cdots \oplus G_i) = \rho(G_1) + \rho(G_2) + \cdots + \rho(G_i) - (j-1),$ which is odd. Since the rank of the joins of the $m - j$ even ones is the sum of the ranks (which is even), the join of all m of them is the sum of the ranks minus $(j-1)$. □

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FAREY GRAPH AND RATIONAL FIXED POINTS OF THE EXTENDED MODULAR GROUP

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Abstract. Fixed points of matrices have many applications in various areas of science and mathematics. The extended modular group $\overline{\Gamma}$ is the group of 2×2 matrices with integer entries and determinant ± 1 . There are strong connections between the extended modular group, continued fractions and Farey graph. The Farey graph is a graph with vertex set $\mathbb{Q}_{\infty} = \mathbb{Q} \cup \{\infty\}$. In this study we consider the elements in $\overline{\Gamma}$ that fix rationals. For a given rational number, we use its Farey neighbours to obtain the matrix representation of the element in $\overline{\Gamma}$ that fixes the given rational. Then we express such elements as words in terms of generators using the relations between the Farey graph and continued fractions. Finally we give the new block reduced form of these words which all blocks have Fibonacci numbers entries.

1. INTRODUCTION

The modular group $\Gamma = PSL(2, \mathbb{Z})$ is the projective special linear group of 2×2 matrices over the ring of integers with determinant one. This group is the quotient group $SL(2,\mathbb{Z})/\pm I$, hence each matrix $\begin{pmatrix} a & b \\ c & d \end{pmatrix}$ represents the same element with its negative $\begin{pmatrix} -a & -b \\ 1 & -a \end{pmatrix}$ $-c$ $-d$). The modular group acts on the upper half plane $\mathbb H$ via linear fractional transformations $z \to \frac{az+b}{cz+d}$. These transformations are orientation preserving isometries of H. Modular group is generated by two elements;

$$
T = \begin{pmatrix} 0 & -1 \\ 1 & 0 \end{pmatrix} \quad U = \begin{pmatrix} 1 & 1 \\ 0 & 1 \end{pmatrix}
$$

Keywords. Extended modular group, fixed points, Farey sequence, Farey graph.

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The presentation of Γ is;

$$
\Gamma = \overline{\sim} \mathbb{Z}_2 * \mathbb{Z}_3,
$$

the free product of \mathbb{Z}_2 and \mathbb{Z}_3 where $S = TU = \begin{pmatrix} 0 & -1 \\ 1 & 1 \end{pmatrix}$. Let us denote the set $G = \left\{ \begin{pmatrix} a & b \\ c & d \end{pmatrix} : a, b, c, d \in \mathbb{Z}, ad - bc = -1 \right\}.$ The corresponding transformations of elements in G are anti-automorphisms. Thus the extended modular group can be defined as $\overline{\Gamma} = PSL(2, \mathbb{Z}) \cup G$. Hence, the extended modular group is the projective linear group $PGL(2, \mathbb{Z})$ and isomorphic to the free product of two dihedral groups of order four and six amalgamated with the cyclic group of order 2 i.e.

$$
\overline{\Gamma} = \langle T, S, R : T^2 = S^3 = R^2 = (TR)^2 = (SR)^2 = I \rangle \approx D_2 *_{\mathbb{Z}_2} D_3
$$

where $R = \begin{pmatrix} 0 & 1 \\ 1 & 0 \end{pmatrix}$ as a reflection map. So the modular group is normal in the extended modular group with index 2.

For each $V \in \overline{\Gamma}$; the number $z \in \mathbb{C} \cup \{\infty\}$ is called a fixed point of V if $V(z) = z$ where $V(z)$ is the corresponding transformation. There is a relation between the number fixed points and trace of V. Elements of $\overline{\Gamma}$ are classified according to the number of fixed points. There are five types of elements in $\overline{\Gamma}$. Now we list the certain types of elements.

If $V \in \Gamma$ then V has at most two fixed points. Also if;

- $|trV| > 2$, then there are two fixed points in $\mathbb{R} \cup {\infty}$ and V is called a hyperbolic element.
- $|trV| = 2$, then there is one fixed point in $\mathbb{R} \cup {\infty}$ and V is called a parabolic element.
- $|trV|$ < 2, then there are two conjugate fixed points in $\mathbb{C} \cup {\infty}$ and V is called an elliptic element.

If $V \in G$ then it has either two fixed points in the real line or the fixed point set is a circle perpendicular to real line. Also if;

- $|trV| \neq 0$, then there is one fixed point in $\mathbb{R} \cup {\infty}$ and V is called a glide reflection.
- $|trV| = 0$, then the set of fixed points is a circle perpendecular to the real line and V is called a reflection.

For more information see $\left| \overline{1,2,11} \right|$.

There are impressive relations between the modular group and continued frac-tions. In [\[25\]](#page-148-0), Rosen defined λ continued fractions for $\lambda \in \mathbb{R}$;

$$
[r_0 \lambda; r_1 \lambda, ..., r_n \lambda] = r_0 \lambda - \frac{1}{r_1 \lambda - \frac{1}{r_2 \lambda - \frac{1}{r_{n-1} \lambda - \frac{1}{r_n \lambda}}}}
$$

In this expansion, for $i \leq n$, $C_i = \frac{p_i}{q_i} = [r_0 \lambda; r_1 \lambda, ..., r_i \lambda]$ is called *ith* convergent of the expansion, for $i \leq n$, $\bigcirc_i - q_i = [0, 0, 1, 1, 0, ..., 1, 0]$ is calculated the convergent of the expansion. And it can be seen by calculation $p_i \cdot q_{i-1} - q_i \cdot p_{i-1} = \pm 1$. Owing to this viewpoint, Rosen revealed a criteria for membership problem for Hecke groups $H(\lambda)$, a general class of modular group. He proved that an element $\begin{pmatrix} a & b \\ c & d \end{pmatrix}$ ∈ $H(\lambda)$ if and only if $\frac{a}{c}$ has a finite λ continued fraction expansion. For $\lambda = 1$ this expression is called integer continued fraction and related to the modular group, on the contrary the membership problem for the modular group is obvious because $\Gamma = PSL(2, \mathbb{Z})$. On the other hand, for $\lambda = 1$ it is possible to make connections between Rosen's fractions and the Farey sequence.

The Farey sequence of order n is a complete and ordered set of reduced rational numbers in the interval $[0, 1]$ which have the denominators not exceeding n.

$$
F_1 = \left\{ \frac{0}{1}, \frac{1}{1} \right\}
$$

$$
F_2 = \left\{ \frac{0}{1}, \frac{1}{2}, \frac{1}{1} \right\}
$$

$$
F_3 = \left\{ \frac{0}{1}, \frac{1}{3}, \frac{1}{2}, \frac{2}{3}, \frac{1}{1} \right\}
$$

$$
F_4 = \left\{ \frac{0}{1}, \frac{1}{4}, \frac{1}{3}, \frac{1}{2}, \frac{2}{3}, \frac{3}{4}, \frac{1}{1} \right\}
$$

It can be seen that if $\frac{a}{c}$ and $\frac{b}{d}$ appears one after another in some F_n then $ad - bc =$ ±1. We called such rationals Farey neighbours. All Farey neighbours of a rational x is denoted by $\mathcal{N}(x)$. The Farey sum of $\frac{a}{c}$ and $\frac{b}{d}$ defined as;

$$
\frac{a}{c} \oplus \frac{b}{d} = \frac{a+b}{c+d}
$$

All Farey neigbours of a rational number can be obtained by Farey sum. More precisely if a rational $\frac{p}{q}$ first appears in F_n by the Farey sum of $\frac{a}{c}$ and $\frac{b}{d}$ in F_{n-1} i.e. $\frac{a}{c} \oplus \frac{b}{d} = \frac{a+b}{c+d} = \frac{p}{q}$ then $\frac{a}{c}$ and $\frac{b}{d}$ are Farey neighbours of $\frac{p}{q}$. Here $\frac{a}{c}$ and $\frac{b}{d}$ are called the Farey parents of $\frac{p}{q}$, and conversely $\frac{p}{q}$ is called the Farey child of $\frac{a}{c}$ and $\frac{b}{d}$. If $\frac{a_i}{c_i}$ is a Farey neighbour of $\frac{p}{q}$ then $\frac{a_i}{c_i} \oplus \frac{p}{q}$ is also a Farey neighbour of $\frac{p}{q}$.

Observe that every F_n includes F_{n-1} and new members are obtained by Farey sum of its neighbours. For instance $\frac{1}{2} \in F_2$ is the Farey sum of $\frac{0}{1}$ and $\frac{1}{1}$ in F_1 . This rule is known as the mediant rule. It should be noted that if the denominator of a Farey sum of two neighbours in F_{n-1} exceeds n then this will not be appear in F_n since the definition of Farey sequence. Definition of Farey sequence can be extended to \mathbb{Q}_{∞} by assuming $\infty = \frac{1}{0}$. Hence for a given rational $\frac{a}{c}$; it is known that $\frac{a}{c}$ has finite integer continued fraction expansion. In addition $\frac{b}{d}$ is the penultimate convergent of the integer continued fraction expansion of $\frac{a}{c}$. This

Figure 1. Farey graph

yields $ad - bc = \pm 1$; in other words $\frac{a}{c}$ and $\frac{b}{d}$ are Farey neighbours. As a result $\begin{pmatrix} a & b \\ c & d \end{pmatrix} \in \overline{\Gamma}.$

The Farey graph is a graph with vertex set \mathbb{Q}_{∞} . And two reduced fractions $\frac{p}{q}$ and $\frac{r}{s}$ are adjacent if and only if $ps - rq = \pm 1$, i.e they are Farey neighbours. An edge between two vertices is drawn by a hyperbolic line in H. The edges between $\frac{1}{0} = \infty$ and every integer a are vertical lines. To construct the graph, first join the vertices $\frac{1}{0}$, $\frac{0}{1}$ and $\frac{1}{1}$ and obtain a big triangle. Then by induction if the endpoints of a long edge are $\frac{a}{c}$ and $\frac{b}{d}$, the label of the third vertex of the triangle is $\frac{a}{c} \oplus \frac{b}{d} = \frac{a+b}{c+d}$, see Figure [1.](#page-137-0)

2. MOTIVATION

There are numerous studies about modular and extended modular group in the literature, related to many branches of mathematics such as group theory, number theory automorphic functions, etc. Algebric structures of subgroups of modular and extended modular group and related topics are studied in $\mathbb{E}[\mathbf{4}|\mathbf{8}, 17]$ [31,](#page-148-4) [33,](#page-148-5) [34\]](#page-148-6). In recent years, many studies have contributed the theory of continued fractions related to the action of some subgroups of Möbius transformations. Series studied the relations between geodesics on the quotient of the hyperbolic plane by the modular group and continued fractions $\boxed{28}$. In $\boxed{2}$, integer continued fraction expansions and geodesic expansions were studied from the perspective of graph theory. Short and Walker represented Rosen continued fractions by path in a class of graphs in hyperbolic geometry [\[30\]](#page-148-8). Same authors defined even integer continued fractions which all digits are even integers. And they studied connections between even integer continued fractions and the Farey graph [\[29\]](#page-148-9).

The fixed points of automorphisms and anti-automorphisms of the extended complex plane have especially been of great interest in many fields of mathematics such as number theory, functional analysis, theory of complex functions, geometry and group theory $[22, 24, 27]$ $[22, 24, 27]$ $[22, 24, 27]$. Also fixed points of elements in $GL(2, \mathbb{R})$ in tropical algebra are discussed in [\[7\]](#page-147-7). In this study we focus on the fixed points of the elements in extended modular group $\overline{\Gamma}$.

Fixed points of an element $V = \begin{pmatrix} a & b \\ c & d \end{pmatrix} \in \overline{\Gamma}$ can be calculated by solving the equation $\frac{az+b}{cz+d} = z$ i.e.

$$
z = \frac{a - d \pm \sqrt{(a + d)^2 - 4}}{2c} \tag{1}
$$

where $ad - bc = 1$ in other words the corresponding transformation $V(z)$ is automorphism. And similarly fixed points of an anti-automorphism are

$$
z = \frac{a - d \pm \sqrt{(a + d)^2 + 4}}{2c}
$$
 (2)

where $ad - bc = -1$. The action of extended modular group on extended rational numbers \mathbb{Q}_{∞} is intriguing. This action is defined as;

$$
\begin{pmatrix} a & b \ c & d \end{pmatrix} \cdot \begin{pmatrix} p \\ q \end{pmatrix} = \begin{pmatrix} ap + bq \\ cp + dq \end{pmatrix}
$$

where $\begin{pmatrix} a & b \\ c & d \end{pmatrix} \in \overline{\Gamma}$ and the column vector $\begin{pmatrix} p \\ q \end{pmatrix}$ q represents the rational number $\frac{p}{q}$. Fixed points of an element in $\overline{\Gamma}$ are rationals if and only if $a + d = 2$ or -2 for the equation \Box and $a + d = 0$ for the equation \Box . This means that rational numbers are fixed only by parabolic or reflection elements.

In this study we establish relations between the Farey graph and elements of $\overline{\Gamma}$ that fixes a given rational $\frac{p}{q}$. Firstly we obtain matrix representation of the element fixing the rational $\frac{p}{q}$ via the Farey neighbours of $\frac{p}{q}$. Then, we consider the relations between paths in the Farey graph and integer continued fractions and obtain the element as a word of the generators U and T . Afterwards, we express this word in block reduced forms and new block reduced forms, related to Fibonacci numbers. Finally, we give some relevant examples of our results.

3. Matrix Representations of the Parabolic and Reflection **ELEMENTS**

In this section we obtain the parabolic and reflection elements in $\overline{\Gamma}$ as matrices that fix a given rational. To do this, we use Farey neighbours.

Theorem 1. Let $z = \frac{p}{q} \in \mathbb{Q}_{\infty}$ and $\frac{r}{s}, \frac{m}{k} \in \mathcal{N}(z)$ then the element

$$
V = \begin{pmatrix} ps - mq & pm - pr \\ qs - qk & pk - qr \end{pmatrix}
$$
 (3)

fixes the rational z.

Proof. Since $\frac{r}{s}$, $\frac{m}{k}$ are Farey neighbours of z, the elements $V_1 = \begin{pmatrix} p & r \\ q & s \end{pmatrix}$ and $V_2 =$ $\begin{pmatrix} p & m \\ q & k \end{pmatrix}$ belong to $\overline{\Gamma}$. Furhermore V_1 and V_2 both send ∞ to $\frac{p}{q}$. As a result $V = V_2 \cdot V_1^{-1}$ is the desired element. \Box

Let $\frac{p}{q}$ and $\frac{r}{s}$ are adjacent such that $\frac{r}{s} < \frac{p}{q}$ then $ps - qr = 1$ otherwise -1. The trace of the element mentioned in \mathcal{B} is $ps - mq + pk - qr$. By the fact that $\frac{r}{s}$, $\frac{m}{k}$ are Farey neighbours of $\frac{p}{q}$ we have $ps - qr = \pm 1$ and $pk - mq = \pm 1$. Hence $tr (V) = 0, \pm 2$ and we have proved the following corollary.

Corollary 1. Let $z = \frac{p}{q} \in \mathbb{Q}_{\infty}$ and $\frac{r}{s}$, $\frac{m}{k} \in \mathcal{N}(z)$. If $\frac{r}{s}$ and $\frac{m}{k}$ are at the same side of $\frac{p}{q}$ then the element in $\boxed{3}$ is parabolic otherwise a reflection.

We know that the fixed point set of a reflection map is a circle perpendicular to real axis. If the element V mentioned in $\boxed{3}$ is a reflection then we know from $\boxed{5}$ that V fixes the circle centered at $\left(\frac{ps-mq}{qs-qk}, 0\right)$ with radius $\frac{1}{\lfloor qs-qk \rfloor}$.

Example 1. For the rational $\frac{8}{3}$ one can choose Farey neighbours as $\frac{5}{2}$ and $\frac{13}{5}$. Then, we have the parabolic element

$$
V = \begin{pmatrix} -23 & 64 \\ -9 & 25 \end{pmatrix}
$$

fixes $\frac{8}{3}$. And if one chooses the neighbours as $\frac{5}{2}$ and $\frac{11}{4}$ then the reflection element

$$
V' = \begin{pmatrix} -17 & 48 \\ -6 & 17 \end{pmatrix}
$$

fixes not only $\frac{8}{3}$ but also the circle centered at $(\frac{17}{6},0)$ with radius $r=\frac{1}{6}$.

Suppose a Farey neighbour of $\frac{p}{q}$ is $\frac{r}{s}$. Then some other neighbours can be obtained by the mediant rule. The following two theorems based on this idea.

Theorem 2. Let $\frac{p}{q} \in \mathbb{Q}_{\infty}$ then the parabolic element that fixes $\frac{p}{q}$ is

$$
V = \begin{pmatrix} \pm 1 - pq & p^2 \\ -q^2 & \pm 1 + pq \end{pmatrix}
$$

Proof. Let $\frac{p}{q} \in \mathbb{Q}_{\infty}$ and $\frac{r}{s}$ is a Farey neighbour of $\frac{p}{q}$. By the mediant rule we have another Farey neighbour $\frac{p+r}{q+s}$ on the same side with $\frac{r}{s}$. Using the same technique in the proof of Theorem $\boxed{1}$ we have the element V as stated. Additionally the trace of the element V is ± 2 with determinant 1 which proves V is parabolic in Γ . \Box

Unlike the Theorem $\boxed{1}$, Theorem $\boxed{2}$ gives an algorithm to obtain a parabolic element that fixes a given rational, without using anything but the rational. Here we do similar things to obtain a reflection whose fixed circle includes a given rational.

Theorem 3. Let $\frac{p}{q} \in \mathbb{Q}_{\infty}$ and $\frac{r}{s}$ is a Farey parent of $\frac{p}{q}$. Then the reflection element in $\overline{\Gamma}$ that fixes $\frac{p}{q}$ is

$$
V = \begin{pmatrix} ps - pq + qr & p^2 - 2pr \\ 2qs - q^2 & -qr + qp - ps \end{pmatrix}
$$

Proof. Suppose $\frac{p}{q} \in \mathbb{Q}_{\infty}$ and $\frac{r}{s}$ is a Farey parent of $\frac{p}{q}$. Another Farey parent of $\frac{p}{q}$ which is at the opposite side of $\frac{r}{s}$ can be obtained by the madiant rule. So we have this parent as $\frac{p-r}{q-s}$. The elements $V_1 = \begin{pmatrix} p & r \\ q & s \end{pmatrix}$ and $V_2 = \begin{pmatrix} p & p-r \\ q & q-s \end{pmatrix}$ $q \mid q - s$) belong to $\overline{\Gamma}$. Although one of them is automorphism, the other is anti-automorphism since the Farey parents are at the opposite side of $\frac{p}{q}$. Hence the element $V = V_2 \cdot V_1^{-1} \in \overline{\Gamma}$ fixes $\frac{p}{q}$. Since $trV = 0$, V is a reflection that the fixed point set is a circle that centered at $\left(\frac{ps-pq+qr}{2qs-q^2},0\right)$ with radius $r=\frac{1}{|2qs-q^2|}$ which proves the result. \Box

So far to this point, we have focused on Farey neighbours. Now observe all the Farey neighbours of a given reduced rational $\frac{p}{q}$. Suppose $\frac{r}{s}$ and $\frac{m}{k}$ are Farey parents of $\frac{p}{q}$ such that $\frac{r}{s} < \frac{p}{q} < \frac{m}{k}$. Then $\frac{p}{q}$ appears in F_q via $\frac{r}{s} \oplus \frac{m}{k} = \frac{p}{q}$. In other words, the hyperbolic line segment joining $\frac{r}{s}$ and $\frac{m}{k}$ covers all the neighbours. Consequently all neighbours of $\frac{p}{q}$ can be obtained by the mediant rule;

$$
\frac{r}{s}<\frac{r}{s}\oplus \frac{p}{q}=\frac{p+r}{q+s}<\frac{p+r}{q+s}\oplus \frac{p}{q}=\frac{2p+r}{2q+s}<...<\frac{p}{q}<...{\frac{p}{q}}\oplus \frac{m}{k}=\frac{p+m}{q+k}<\frac{m}{k}
$$

4. Farey Paths, Integer Continued Fractions and Blocks in Extended Modular Group Γ

In this section, the relation between integer continued fractions and Farey paths is used to obtain the word form of the element in $\overline{\Gamma}$, which fixes a given rational number, in terms of generators. A path in a graph consists of consequtive adjacent vertices. So a Farey path $\langle v_1, v_2, ..., v_n \rangle$ is a path such that $v_i = \frac{p_i}{q_i}$ for $i =$ 1, 2, ..., *n* are reduced rationals and since the consequtive v_i 's are adjacent $p_i \cdot q_{i-1}$ $q_i.p_{i-1} = \pm 1$. The Farey graph is connected hence there is a natural distance between two rationals v and w that is $d(v, w)$, the minimum number of edges in any path from v to w in F_n . The distance of an integer to ∞ is $d(\infty, x) = 1$.

Lemma 1. $\boxed{25}$ Let $\frac{p}{q} = [r_0; r_1, r_2, ..., r_n]$ be a reduced rational number then;

$$
U^{r_0}TU^{r_1}TU^{r_2}T...U^{r_n}T\begin{pmatrix}1\\0\end{pmatrix}=\begin{pmatrix}p\\q\end{pmatrix}
$$

Theorem 4. Let $\frac{p}{q}$ be a reduced rational and have an integer continued fraction expansion as $[r_0; r_1, r_2, ..., r_n]$, then the parabolic element fixing $\frac{p}{q}$ is

$$
U^{r_0}TU^{r_1}TU^{r_2}T...U^{r_n}T.U.TU^{-r_n}TU^{-r_{n-1}}T...U^{-r_1}TU^{-r_0}
$$
\n
$$
\tag{4}
$$

Proof. Let $\frac{p}{q} = [r_0; r_1, r_2, ..., r_n]$. By Lemma $\boxed{1}$, we have

$$
U^{r_0}TU^{r_1}TU^{r_2}T...U^{r_n}T.U.TU^{-r_n}TU^{-r_{n-1}}T...U^{-r_1}TU^{-r_0}\begin{pmatrix}p\\q\end{pmatrix}
$$

=
$$
U^{r_0}TU^{r_1}TU^{r_2}T...U^{r_n}T.U\begin{pmatrix}1\\0\end{pmatrix}
$$

=
$$
U^{r_0}TU^{r_1}TU^{r_2}T...U^{r_n}T\begin{pmatrix}1\\0\end{pmatrix}
$$

=
$$
\begin{pmatrix}p\\q\end{pmatrix}
$$

Since conjugacy preserves the trace we have

$$
tr(U^{r_0}TU^{r_1}TU^{r_2}T...U^{r_n}T.U.TU^{-r_n}TU^{-r_{n-1}}T...U^{-r_1}TU^{-r_0})
$$

=
$$
tr(U) = 2
$$

which proves the element given in $\left(\frac{1}{4} \right)$ is parabolic. \Box

We know from $\boxed{9}$ that stabilizer of a point in Γ is an infinite cyclic group. So we can give the following corollary.

Corollary 2. Let
$$
\frac{p}{q} = [r_0; r_1, r_2, ..., r_n] \in \mathbb{Q}
$$
; then for all $0 \neq k \in \mathbb{Z}$;

$$
U^{r_0}TU^{r_1}TU^{r_2}T...U^{r_n}T.U^k.TU^{-r_n}TU^{-r_{n-1}}T...U^{-r_1}TU^{-r_0}
$$

is a parabolic element in Γ whose fixed point is $\frac{p}{q}$.

Now we obtain a reflection element as a word in generators of $\overline{\Gamma}$ that fixes a given rational $\frac{p}{q}$.

Theorem 5. Let $\frac{p}{q}$ be a reduced rational and have an integer continued fraction expansion as $[r_0; r_1, r_2, ..., r_n]$, then the reflection element in $\overline{\Gamma}$ fixing $\frac{p}{q}$ is

$$
U^{r_0}TU^{r_1}TU^{r_2}T...U^{r_n}T.RTU.TU^{-r_n}TU^{-r_{n-1}}T...U^{-r_1}TU^{-r_0}
$$

Proof. We have $RTU = \begin{pmatrix} 1 & 1 \\ 0 & 1 \end{pmatrix}$ $0 -1$) as a reflection map. Furthermore $RTU\begin{pmatrix} 1\\ 0 \end{pmatrix}$ 0 $=$ (1) $\boldsymbol{0}$. The rest of the proof follows similar to the proof of Theorem $\overline{4}$.

Example 2. Choose the rational $\frac{8}{5}$. The integer continued fraction expansion of $\frac{8}{5}$ is

$$
\frac{8}{5} = 2 - \frac{1}{3 - \frac{1}{2}} = [2; 3, 2].
$$

Then the parabolic element fixing $\frac{8}{5}$ is

$$
U^2TU^3TU^2TUTU^{-2}TU^{-3}TU^{-2} = \begin{pmatrix} -39 & 64 \\ -25 & 41 \end{pmatrix}.
$$

And the reflection element is

$$
U^{2}TU^{3}TU^{2}TRTUTU^{-2}TU^{-3}TU^{-2} = \begin{pmatrix} -89 & 104 \\ -55 & 89 \end{pmatrix}
$$

Here we mention about relations between paths in the Farey graph and integer continued fractions. The convergents of a certain continued fraction expansion of a reduced rational $\frac{p}{q} = [r_0; r_1, ..., r_n]$, are defined as $C_i = \frac{p_i}{q_i} = [r_0; r_1, ..., r_i]$ for $0 \leq i \leq n$, where $C_0 = \frac{p_0}{q_0} = \frac{r_0}{1}$ and $C_n = \frac{p_n}{q_n} = \frac{p}{q}$. Furthermore we know that $p_i \tcdot q_{i-1} - q_i \tcdot p_{i-1} = \pm 1$. Hence every consequtive pair C_i and C_{i-1} are Farey neighbours. Also, since $C_0 = r_0 \in \mathbb{Z}$ and every integer is adjacent to infinity with a vertical line, $\langle \infty, C_0, C_1, ..., C_{n-1}, C_n \rangle$ is a path from ∞ to $\frac{p}{q}$. To sum up every integer fraction expansion of a rational $\frac{p}{q}$ is related to a path from ∞ to $\frac{p}{q}$. Moreover the shortest integer continued fraction of $\frac{p}{q}$ is related to a geodesic path from ∞ to $\frac{p}{q}$. In Theorem **4** and Theorem **5**, the integer continued fraction expansion of a given rational is related to an element in $\overline{\Gamma}$ that fixes the rational. It is possible to make connections with Farey paths.

5. BLOCK REDUCED FORMS IN THE EXTENDED MODULAR GROUP $\overline{\Gamma}$

Every element in $\overline{\Gamma}$ can be expressed as a word of T, S and R denoted by $W(T, S, R)$. Consider the blocks

$$
TS = \begin{pmatrix} 1 & 1 \\ 0 & 1 \end{pmatrix} \quad \text{and} \quad TS^2 = \begin{pmatrix} 1 & 0 \\ 1 & 1 \end{pmatrix}
$$

Using these blocks every reduced word $W(T, S, R)$ in $\overline{\Gamma}$ where the sum of exponents of R is an even number can be expressed as;

$$
S^{i} (TS)^{m_0} (TS^2)^{n_0} \dots (TS)^{m_k} (TS^2)^{n_k} T^j,
$$

and every reduced word $W(T, S, R)$ in $\overline{\Gamma}$ where the sum of exponents of R is an odd number can be expressed as;

$$
S^{i} (TS)^{m_0} (TS^2)^{n_0} \dots (TS)^{m_k} (TS^2)^{n_k} T^j R
$$

Here $i = 0, 1, 2, j = 0, 1, m_0$ and n_k may be zero and other exponents are positive integers. This representetion is known as the block reduced form [\[13\]](#page-147-10). For example, the block reduced form of the word $W(T, S, R) = TSTSTSSTSTST$ is $(TS)^2$. $(TS^2)^2 T$. And the block reduced form of the word $W(T, S, R) = RTS^2RTS^2R$ is (TS) . $(TS^2)^2 R$. Trace classes of the modular group and extended modular group are studied in $[6, 13]$ $[6, 13]$ by using the block reduced form. In this section we give the block reduced form of the element in $\overline{\Gamma}$ fixing a given rational $\frac{p}{q}$.

Theorem 6. Let $\frac{p}{q}$ be a reduced rational number and have an integer continued fraction expansion $[r_0; r_1, ..., r_n]$ then the block form of the parabolic element fixing $rac{p}{q}$ is

$$
W(T, S, R) = (TS)^{r_0 - 1} (TS^2) (TS)^{r_1 - 2} (TS^2) ... (TS)^{r_{n-1} - 2} (TS^2).
$$

\n
$$
(TS)^{r_n - 1} (TS^2)^{-1} (TS)^{-r_n - 1} (TS^2) (TS)^{-r_{n-1} - 2} (TS^2).
$$

\n
$$
... (TS)^{-r_1 - 2} (TS^2) (TS)^{-r_0 - 1}
$$

Proof. By Theorem $\overline{4}$, we know that

$$
U^{r_0}TU^{r_1}TU^{r_2}T...U^{r_n}T.U.TU^{-r_n}TU^{-r_{n-1}}T...U^{-r_1}TU^{-r_0}
$$

fixes $\frac{p}{q}$. Considering $U = TS$ we have

$$
W(T, S, R) = (TS)^{r_0} .T. (TS)^{r_1} .T... (TS)^{r_{n-1}} .T. (TS)^{r_n} .T
$$

\n
$$
(TS) .T. (TS)^{-r_n} .T. (TS)^{-r_{n-1}} .T.... (TS)^{-r_1} .T. (TS)^{-r_0}
$$

\n
$$
= (TS)^{r_0-1} TS.T.T.S. (TS)^{r_1-2} TS.T...TS (TS)^{r_{n-1}-2} TS.
$$

\n
$$
T.T.S (TS)^{r_n-1} .T (TS) .T. (TS)^{-r_n-1} TS.T.T.S (TS)^{-r_n-2} .
$$

\n
$$
T S.T....TS (TS)^{-r_1-2} TS.T.T.S (TS)^{-r_0-1}
$$

Using the relations $T^2 = I$ and $(TS^2)^{-1} = ST$,

$$
W(T, S, R) = (TS)^{r_0 - 1} \cdot (TS^2) \cdot (TS)^{r_1 - 2} \cdot (TS^2) \cdot (TS^2) \cdot (TS)^{r_{n-1} - 2} \cdot (TS^2) \cdot (TS)^{r_n - 1} \cdot (TS^2) \cdot (TS)^{-r_n - 1} \cdot (TS^2) \cdot (TS) \cdot (TS) \cdot (TS^2) \cdot (TS)^{-r_1 - 2} \cdot (TS^2) \cdot (TS)^{-r_0 - 1}
$$

Theorem 7. Let $\frac{p}{q}$ be a reduced rational number and have an integer continued fraction expansion $[r_0; r_1, ..., r_n]$ then the block form of the reflection element fixing $\frac{p}{q}$ is

$$
W(T, S, R) = (TS)^{r_0 - 1} \cdot (TS^2) \cdot (TS)^{r_1 - 2} \cdot (TS^2) \cdot \dots \cdot (TS)^{r_{n-1} - 2} \cdot (TS^2) \cdot \dots
$$

$$
(TS)^{r_n} \cdot (TS^2)^{-r_n - 2} \cdot (TS) \cdot (TS^2)^{-r_{n-1} - 2} \cdot (TS) \cdot \dots
$$

$$
(TS) \cdot (TS^2)^{-r_1 - 2} \cdot (TS) (TS^2)^{-r_0 - 1} \cdot R
$$

Proof. From Theorem $\overline{5}$, the reflection element fixing $\frac{p}{q}$ is

$$
U^{r_0}TU^{r_1}TU^{r_2}T...U^{r_n}T.RTU.TU^{-r_n}TU^{-r_{n-1}}T...U^{-r_1}TU^{-r_0}.
$$

After substituting $U = TS$ in the word above, we have

$$
W(T, S, R) = (TS)^{r_0} T (TS)^{r_1} T \dots (TS)^{r_{n-1}} T (TS)^{r_n} T
$$

\n
$$
RT (TS) T (TS)^{-r_n} T (TS)^{-r_{n-1}} T \dots (TS)^{-r_1} T (TS)^{-r_0}
$$

\n
$$
= (TS)^{r_0-1} T STTS (TS)^{r_1-2} TSTTS \dots TS (TS)^{r_{n-1}-2} TS
$$

\n
$$
TTS (TS)^{r_n-1} TRST (TS) (TS)^{-r_n-2} TSTTS (TS)^{-r_{n-1}-2}
$$
$$
TST...TS (TS)^{-r_1-2} TSTTS (TS)^{-r_0-1}
$$

Since $(TR)^2 = (SR)^2 = I$ we obtain $TR = RT$ and $SR = RS^2$. Hence,

$$
W (T, S, R) = (TS)^{r_0-1} (TS^2) (TS)^{r_1-2} (TS^2) ... (TS)^{r_{n-1}-2} (TS^2)
$$

$$
(TS)^{r_n} (TS^2)^{-r_n-2} (TS) (TS^2)^{-r_{n-1}-2} (TS) ...
$$

$$
(TS) (TS^2)^{-r_1-2} (TS) (TS^2)^{-r_0-1} R
$$

We can obtain elements which fix a given rational $\frac{p}{q}$ in terms of TS and TS² by finding a path from ∞ to $\frac{p}{q}$ in the Farey graph. We explain this with an example:

Example 3. Suppose the given rational is $\frac{-10}{3}$. Then one may choose the path < $\infty, -3, \frac{-13}{4}, \frac{-10}{3}$, see Figure [2.](#page-145-0) We know the consequtive vertices in this path are consequtive convergents of the integer continued fraction expansion of the rational $\frac{-10}{3}$ i.e., $C_0 = -3$, $C_1 = \frac{-13}{4}$ and $C_2 = \frac{-10}{3}$. Hence, we obtain the integer continued fraction expansion as

$$
-3 - \frac{1}{4 - \frac{1}{1}} = [-3, 4, 1]
$$

Using the values $r_0 = -3$, $r_1 = 4$ and $r_2 = 1$ in Theorem $\boxed{6}$ we have the parabolic element fixing $\frac{-10}{3}$ in blocks TS and TS² as follows:

$$
W(T, S, R) = (TS)^{-4} (TS^2) (TS)^2 (TS^2) (TS)^0 (TS^2)^{-1} (TS)^{-2}
$$

$$
(TS^2) (TS)^{-6} . (TS^2) (TS)^2
$$

We can reduce this word by the presentation of Γ as;

$$
W(T, S, R) = S^2 \cdot (TS^2)^2 \cdot (TS)^2 \cdot (TS^2)^4 \cdot (TS)^3
$$

For the reflection element fixing $\frac{-10}{3}$ we use Theorem [7;](#page-143-0)

$$
W(T, S, R) = (TS)^{-4} (TS^2) (TS)^2 (TS^2) (TS)^1 (TS^2)^{-3} (TS)
$$

$$
(TS^2)^{-6} (TS) (TS^2)^2 R
$$

The block reduced form of this word can be obtained by the relators of $\overline{\Gamma}$;

$$
W(T, S, R) = S^{2}. (TS^{2})^{2}. (TS)^{3}. (TS^{2})^{3}. (TS)^{3}. (TS^{2})^{3}. R
$$

6. Fibonacci Sequence and New Block Reduced Forms

Jones and Thornton obtained relations between elements of extended modular group and Fibonacci numbers in $\vert 10 \vert$. Özgür defined two new sequences which are group and Friomacci numbers in **[10]**. Ozgur defined two new sequences which are generalizations of Fibonacci and Lucas sequences for the Hecke group $H(\sqrt{q})$ where $q \geq 5$ prime $\boxed{32}$. Also there are some results for Modular group and Pell Fibonacci and Lucas numbers in $\left| \frac{14}{16} \right| \left| \frac{23}{8} \right|$. Koruoğlu and Şahin used a generalized version of Fibonacci sequence to get relations with extended Hecke groups $\overline{H}(\lambda)$ in [\[12\]](#page-147-3). In

FIGURE 2. The path $<\infty, -3, \frac{-13}{4}, \frac{-10}{3}$

same study they give an application to extended modular group $\overline{\Gamma}$. They considered the following elements:

$$
f = RTS = \begin{pmatrix} 0 & 1 \\ 1 & 1 \end{pmatrix} \text{ and } h = TSR = \begin{pmatrix} 1 & 1 \\ 1 & 0 \end{pmatrix}
$$

The k^{th} power of f and h are;

$$
f^k = \begin{pmatrix} f_{k-1} & f_k \\ f_k & f_{k+1} \end{pmatrix} \text{ and } h^k = \begin{pmatrix} f_{k+1} & f_k \\ f_k & f_{k-1} \end{pmatrix}
$$

where f_k denotes the kth Fibonacci number. Hence every element in extended modular group can be expressed as a word in f and h . This reduced word called New Block Reduced Form. The relations between block reduced forms and new block reduced forms are;

$$
TS = Rf = hR \tag{5}
$$

$$
TS^2 = Rh = fR \tag{6}
$$

It is proved that every block reduced word has a New Block Reduced Form. From this viewpoint we can express the element given in Theorem $\boxed{6}$ and Theorem $\boxed{7}$ in new block reduced form. We explain this with an example.

In example 3 the parabolic element fixing $\frac{-10}{3}$ is;

$$
S^{2} (TS^{2})^{2} (TS)^{2} (TS)^{4} (TS)^{3}
$$

Using the relations $\overline{5}$ and $\overline{6}$ and $S^2 = TfR$; we can write this word;

 $T f R.(Rh.fR).(Rf.hR).(Rh.fR.Rh.fR).(Rf.hR.Rf)$

Since $R^2 = I$ we have the block reduced form;

$$
T.f.h.f^2.h^2.f.h.f^2.h.f = \begin{pmatrix} 0 & -1 \\ 1 & 0 \end{pmatrix} \begin{pmatrix} 0 & f_1 \\ f_1 & f_2 \end{pmatrix} \cdot \begin{pmatrix} f_2 & f_1 \\ f_1 & 0 \end{pmatrix} \begin{pmatrix} f_1 & f_2 \\ f_2 & f_3 \end{pmatrix} \begin{pmatrix} f_3 & f_2 \\ f_2 & f_1 \end{pmatrix}
$$

$$
\begin{pmatrix} 0 & f_1 \\ f_1 & f_2 \end{pmatrix} \begin{pmatrix} f_2 & f_1 \\ f_1 & 0 \end{pmatrix} \begin{pmatrix} f_1 & f_2 \\ f_2 & f_3 \end{pmatrix} \begin{pmatrix} f_2 & f_1 \\ f_1 & 0 \end{pmatrix} \begin{pmatrix} 0 & f_1 \\ f_1 & f_2 \end{pmatrix}
$$

It is stated in the same example that the reflection element fixing $\frac{-10}{3}$ is;

$$
S^{2} (TS^{2})^{2} (TS)^{3} (TS^{2})^{3} (TS)^{3} (TS^{2})^{3} R
$$

Following the same procedure above we have the new block reduced form of this word as;

$$
T.f.h. (f2.h)4.f = \begin{pmatrix} 0 & -1 \\ 1 & 0 \end{pmatrix} \begin{pmatrix} 0 & f_1 \\ f_1 & f_2 \end{pmatrix} \begin{pmatrix} f_2 & f_1 \\ f_1 & 0 \end{pmatrix} \begin{bmatrix} f_1 & f_2 \\ f_2 & f_3 \end{bmatrix} \begin{pmatrix} f_2 & f_1 \\ f_1 & f_0 \end{pmatrix} \begin{bmatrix} 0 & f_1 \\ f_1 & f_2 \end{bmatrix}
$$

7. CONCLUSION

In this article, elements in the extended modular group $\overline{\Gamma}$ which fix rationals, are considered. Matrix representations of parabolic and reflection elements which fix a given rational are mentioned in Section \mathbb{S} via Farey neighbours. In Section $\overline{4}$ $\overline{4}$ $\overline{4}$ relationship between Farey paths and elements of $\overline{\Gamma}$ which have rational fixed points, is established. And these elements obtained as words in generators U and T. Then, block reduced form of these words are given in Section $\overline{5}$. We use new block reduced forms in Section 6 to establish relations with Fibonacci numbers. As a summary of this work we give a final example, see Table \Box

Path	$<\infty, 0, \frac{1}{2}, \frac{3}{7}>$
ICF	$[0;-2,3]$
$W(U,T)$ for parabolic element	$T.U^{-2}.T.U^3.T.U.T.U^{-3}.T.U^2.T$
BRF for parabolic element	$(TS^2)^2$, $(TS)^2$, $(TS^2)^4$, $(TS)^2$, T
NBRF for parabolic element	$f.h^2.f^2.h.f.h^2.f.T$
$W(U,T)$ for reflection element	$\overline{T.U^{-2}.T.U^{3}.T.R.T.U.T.U^{-3}.T.U^{2}.T}$
BRF for reflection element	$(TS^2)^2$, $(TS)^3$, (TS^2) , $(TS)^3$, $(TS^2)^2$, $T.R$
NBRF for reflection element	$f.h^2.f.h^3.f.h^2.f.T$
	$m_{1} = 1$ m \cdots \overline{R} e \cdots 3

TABLE 1. Elements in $\overline{\Gamma}$ fixing $\frac{3}{7}$

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SOME RESULTS ON PSEUDOSYMMETRIC NORMAL PARACONTACT METRIC MANIFOLDS

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ABSTRACT. In this article, the M -projective and Weyl curvature tensors on a normal paracontact metric manifold are discussed. For normal paracontact metric manifolds, pseudosymmetric cases are investigated and some interesting results are obtained. We show that a semisymmetric normal paracontact manifold is of constant sectional curvature. We also obtain that a pseudosymmetric normal paracontact metric manifold is an η –Einstein manifold. Finally, we support our topic with an example.

1. INTRODUCTION

The notion of odd-dimensional manifolds with contact and almost contact structures was initiated by Boothby and Wang $\left[\frac{1}{2}\right]$. In $\left[\frac{2}{2}\right]$, Sasaki and Hatakeyama reinvestigated them using tensor calculus. Tanno in [\[3\]](#page-162-3) classified connected almost contact metric manifolds whose automorphism groups possess maximum dimension. For such manifolds, the sectional curvature of a plane section containing ξ is a constant named c. He showed that it can be divided into the following three classes.

• Class-1⇒ Homogeneous normal contact Riemannian manifolds with $c > 0$.

• Class- $2 \Rightarrow$ Global Riemannian products of a line or a circle with a Kähler manifold of constant holomorphic sectional curvature if $c = 0$.

Keywords. M-projective curvature tensor, Weyl curvature tensor, Einstein manifold, η−Einstein manifold.

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• Class-3 \Rightarrow A warped product space $\mathbb{R} \times_f C$ if $c < 0$.

It is well known that the manifolds of class-1 are characterized by admitting a Sasakian structure. In $[4]$, Kenmotsu investigated the differential geometric properties of the manifolds of class-3. In general, these structures are not Sasakian $\vert 5 \vert$.

In $[6]$, S. Zamkovoy and G. Nakova reviewed the decomposition of almost contact metric manifolds in eleven classes. In addition to almost paracontact metric manifolds, K. Mandal and U.C De in $|7|$, N. Özdemir, S. Aktay and M.solgun in [\[8\]](#page-162-8) examined paracontact metric manifolds and obtained their various geometric properties. Also, in $\left[9\right]$, H. Pandey and A. Kumar examined the anti-invariant submanifolds of almost paracontact manifolds. Similarly, J. Welyczko [\[10\]](#page-162-10) studied Legendre curves on 3-dimensional normal paracontact metric manifolds.

After then, in $[11]$, Pokhariyal and Mishra have introduced an *M*-projective curvature tensor on a Riemannian manifold. The properties of the \mathcal{M} -projective curvature tensor in Sasakian and Kähler manifolds were developed by Ojha in $\boxed{12}$. He showed that it bridges the gap between conformal curvature tensor, conharmonic curvature tensor, and concircular curvature tensor. M-projective curvature tensor on different manifolds studied by many geometers such as Kenmotsu, Sasakian, and generalized Sasakian space form.

In [\[14\]](#page-162-13), by using some tensors, invariant submanifolds of an almost Kenmotsu (κ, μ, ν) -space are characterized. Similarly, many authors have presented important work with various manifolds and some curvature tensors on them $\left(\sqrt{13}, \sqrt{15}-\sqrt{18}\right)$.

Motivated by these ideas, we have attempted to study properties of the Weylconformal curvature tensor in a normal paracontact metric manifold. We think that some interesting results contribute differential geometry.

The present paper is organized as follows.

In section 2, we give the notations and preliminary results of normal paracontact metric manifolds that will be used later. In section 3, we show that a normal paracontact metric manifold satisfying $R(X, Y) \cdot R = 0$ if and only if it has constant sectional curvature and $R(X, Y) \cdot \mathcal{M} = 0$ implies that it η -Einstein manifold.

2. Preliminaries

An almost paracontact structure on a n -dimensional smooth manifold M is given by a (1, 1)-type tensor field φ , a vector field ξ , and a 1-form η satisfying the following condition

$$
\varphi^2 = I - \eta \otimes \xi, \quad \eta(\xi) = 1. \tag{1}
$$

As an immediate consequent $\varphi \xi = 0$, $\eta \circ \varphi = 0$ and the tensor φ has constant rank $n-1$. If an almost paracontact manifold is endowed with a semi-Riemannian metric g such that

$$
g(\varphi X, \varphi Y) = -g(X, Y) + \eta(X)\eta(Y), \quad g(X, \xi) = \eta(X),\tag{2}
$$

for any $X, Y \in \Gamma(TM)$, then $M^n(\varphi, \xi, \eta, g)$ is called an almost paracontact metric manifold, where $\Gamma(TM)$ is the set of the differentiable vector fields on M. It follows that

$$
g(\varphi X, Y) = -g(X, \varphi Y).
$$

The fundamental 2-form of the almost paracontact metric manifold is given by

$$
\Phi(X, Y) = g(\varphi X, Y).
$$

If $d\eta = \Phi$, then η becomes a contact form, that is, $\eta \wedge (d\eta)^n \neq 0$ and $M^n(\varphi, \xi, \eta, g)$ is said to be a paracontact metric manifold. Any such pseudo-Riemannian metric manifold is of signature $(\frac{n+1}{2}, \frac{n-1}{2})$ for $n = 2m + 1$. In this case, we have

$$
(\nabla_X \varphi)Y = -g(X, Y)\xi - \eta(Y)X + 2\eta(X)\eta(Y)\xi,
$$
\n(3)

for any $X, Y \in \Gamma(TM)$, where ∇ denote the Levi-Civita connection on M. [\(1\)](#page-150-0) and [\(3\)](#page-151-0) imply that

$$
\nabla_X \xi = \varphi X.
$$

An almost paracontact structure is said to be normal if the tensor $N_{\varphi} - 2d\eta \oplus \xi = 0$ [\[13\]](#page-162-14), where N_{φ} the Nijenhuis tensor of φ given by

$$
N_{\varphi}(X,Y) = \varphi^{2}[X,Y] + [\varphi X, \varphi Y] - \varphi[\varphi X, Y] - \varphi[X, \varphi Y]
$$

For the sake of brevity, a normal paracontact metric manifold is said to be paracontact metric manifold [\[8\]](#page-162-8).

A normal paracontact metric manifold M is of a constant sectional curvature c , then its Riemannian curvature tensor R is given by

$$
R(X,Y)Z = \frac{c+1}{4} \Biggl\{ g(Y,Z)X - g(X,Z)Y \Biggr\}
$$
\n
$$
+ \frac{c-1}{4} \Biggl\{ \eta(X)\eta(Z)Y - \eta(Y)\eta(Z)X + g(X,Z)\eta(Y)\xi
$$
\n
$$
- g(Y,Z)\eta(X)\xi + g(\varphi Y,Z)\varphi X - g(\varphi X,Z)\varphi Y - 2g(\varphi X,Y)\varphi Y \Biggr\},
$$
\n(4)

for any $X, Y, Z \in \Gamma(TM)$ [\[8\]](#page-162-8).

For a $(0, k)$ -type tensor field T and a $(0, 2)$ -type tensor field A on a semi-Riemannian manifold (M, g) , the Tachibana tensor $Q(A, T)$ is defined as

$$
Q(A,T)(X_1, X_2, ..., X_k; X, Y) = -T((X \wedge_A Y)X_1, X_2, ...X_k) -T(X_1, (X \wedge_A Y)X_2, ...X_k) \n\vdots -T(X_1, ...X_{k-1}, (X \wedge_A Y)X_k), \qquad (5)
$$

for all $X_1, X_2, ... X_k, X, Y \in \Gamma(TM)$, where $X \wedge_A Y$ is an endomorphism defined by

$$
(X \wedge_A Y)Z = A(Y, Z)X - A(X, Z)Y.
$$
\n⁽⁶⁾

A semi-Riemannian manifold (M, g) is pseudosymmetric if its the Riemannian curvature tensor R satisfies

$$
R \cdot R = LQ(g, R),\tag{7}
$$

where L is a function on M. Particularly, if $L = 0$, it is called a semisymmetric manifold.

On a normal paracontact metric manifold M^n , the following relations hold;

$$
R(X,Y)\xi = \eta(X)Y - \eta(Y)X\tag{8}
$$

$$
R(\xi, X)Y = \eta(Y)X - g(X, Y)\xi \tag{9}
$$

$$
\eta(R(X,Y)Z) = \eta(Y)g(X,Z) - \eta(X)g(Y,Z) \tag{10}
$$

$$
S(X,\xi) = (1-n)\eta(X), \ \ Q\xi = (1-n)\xi, \tag{11}
$$

for any $X, Y, Z \in \Gamma(TM)$, where S and Q are, respectively, the Ricci tensor and Ricci operatory of M given by $g(QX, Y) = S(X, Y)$.

On the other hand, the Weyl-conformal curvature and M-projective curvature tensors play an important role in differential geometry as well as in relativity. The Weyl-conformal curvature tensor and M-projective curvature tensor of a Riemannian manifold (M^n, g) , $n > 2$, are respectively, defined by

$$
C(X,Y)Z = R(X,Y)Z - \frac{1}{n-2} \{g(Y,Z)QX - g(X,Z)QY + S(Y,Z)X - S(X,Z)Y \} + \frac{\tau}{(n-1)(n-2)} \{g(Y,Z)X - g(X,Z)Y \}
$$
(12)

and

$$
\mathcal{M}(X,Y)Z = R(X,Y)Z - \frac{1}{2(n-1)} \{ S(Y,Z)X - S(X,Z)Y + g(Y,Z)QX - g(X,Z)QY \},
$$
\n(13)

for any $X, Y, Z \in \Gamma(TM)$, where τ denote the scalar curvature of M.

A normal paracontact metric manifold M is called η -Einstein if its Ricci tensor S is of the form

$$
S(X,Y) = ag(X,Y) + b\eta(X)\eta(Y),\tag{14}
$$

for any $X, Y \in \Gamma(TM)$, where a and b are arbitrary constants. If $b = 0$, then manifold is said to be Einstein.

If a normal paracontact manifold $M^n(\varphi, \eta, \xi, g)$ is an η -Einstein, from [\(11\)](#page-152-0) and (14) , we get $1 - n = a + b$, $\tau = na + b$, that is,

$$
a = 1 + \frac{\tau}{n-1}
$$
 and $b = -n - \frac{\tau}{n-1}$.

Thus (14) takes form

$$
S(X,Y) = g(X,Y)(1 + \frac{\tau}{n-1}) - (n + \frac{\tau}{n-1})\eta(X)\eta(Y).
$$
 (15)

Theorem 1. An n-dimensional M-projectively flat normal paracontact metric manifold M^n is an Einstein manifold.

Proof. Let us assume that normal paracontact metric manifold M^n is M -projectively flat, then from (8) and (13) , we obtain

$$
R(X,Y)Z = \frac{1}{2(n-1)} \{ S(Y,Z)X - S(X,Z)Y + g(Y,Z)QX - g(X,Z)QY \}.
$$

Here replacing $Z = \xi$ and using [\(8\)](#page-152-0), we obtain

$$
\eta(X)Y - \eta(Y)X = \frac{1}{n-1} \{\eta(Y)X - \eta(X)Y\},\
$$

which implies that

$$
QX = (1 - n)X,
$$

or

$$
S(X,Y) = (1-n)g(X,Y),
$$
\n(16)

for all $X, Y \in \Gamma(TM)$.

Proposition 1. If A normal paracontact metric manifold $M^n(\varphi, \eta, \xi, g)$ is Weylconformally flat, then it an η-Einstein manifold.

Next, let us suppose that normal paracontact metric manifold M^n is Weylconformal flat, then from (12) , we have

$$
R(X,Y)Z = \frac{1}{n-2} \left\{ g(Y,Z)QX - g(X,Z)QY + S(Y,Z)X \right\}
$$

$$
- S(X,Z)Y \left\} - \frac{\tau}{(n-1)(n-2)} \left\{ g(Y,Z)X - g(X,Z)Y \right\}, (17)
$$

for any $X, Y, Z \in \Gamma(TM)$. Taking $Z = \xi$ and making use of $\boxed{8}$ and $\boxed{11}$, we have

$$
\eta(X)Y - \eta(Y)X = \frac{1}{n-2} \left\{ \eta(Y)QX - \eta(X)QY + (n-1)\eta(Y)X - (n-1)\eta(X)Y \right\} - \frac{\tau}{(n-1)(n-2)} \{\eta(Y)X - \eta(X)Y\}.
$$
 (18)

This implies that

$$
(1 + \frac{\tau}{n-1})(\eta(X)Y - \eta(Y)X) + \eta(Y)QX - \eta(X)QY = 0.
$$

It follows for $Y = \xi$,

$$
QX = -(n + \frac{\tau}{n-1})\eta(X)\xi + (1 + \frac{\tau}{n-1})X,
$$

that is, the Weyl- projectively flat normal paracontact metric manifold is an η -Einstein. Thus we have

$$
S(X,Y) = (1 + \frac{\tau}{n-1})g(X,Y) - (n + \frac{\tau}{n-1})\eta(X)\eta(Y).
$$
 (19)

From (15) and (19) , we have the following Proposition.

Proposition 2. A normal paracontact metric manifold $M^n(\varphi, \eta, \xi, g)$ is an η -Einstein manifold if it is Weyl-projectively flat.

3. Pseudosymmetric Normal Paracontact Metric Manifolds

In this section, we consider pseudosymmetric normal paracontact metric manifolds.

Theorem 2. If a normal paracontact metric manifold $M^n(\varphi, \xi, \eta, g)$ is pseudosymmetric provided $L \neq -1$, then it is an η -Einstein manifold. Furthermore, it is a semisymmetric if and only if it has a constant sectional curvature 1.

Proof. We suppose that *n*-dimensional normal paracontact metric manifold M^n is pseudosymmetric. Then from $\sqrt{7}$, we have

$$
(R(X, Y) \cdot R)(U, V, Z) = LQ(g, R)(U, V, Z; X, Y),
$$

for all $X, Y, Z, U, V \in \Gamma(TM)$. It follows that

$$
R(X,Y)R(U,V)Z - R(R(X,Y)U,V)Z - R(U,R(X,Y)V)Z
$$

- R(U,V)R(X,Y)Z = -L{R((X \wedge_g Y)U,V)Z
+ R(U,(X \wedge_g Y)V)Z + R(U,V)(X \wedge_g Y)Z}. (20)

Putting $Y = Z = \xi$ in [\(20\)](#page-154-1) and by virtue of [\(9\)](#page-152-0), we have

$$
R(X,\xi)R(U,V)\xi - R(R(X,\xi)U,V)\xi - R(U,R(X,\xi)V)\xi - R(U,V)R(X,\xi)\xi = -L\{R(\eta(U)X - g(X,U)\xi,V)\xi + R(U,\eta(V)X - g(X,V)\xi)\xi + R(U,V)(X - \eta(X)\xi)\}.
$$

after necessary arrangements are made, we conclude

$$
\begin{array}{lcl} R(U,V)X+g(X,V)U&-&g(X,U)V=L\{g(X,U)V-g(X,V)U\\&+&g(X,V)\eta(U)\xi-g(X,U)\eta(V)\xi-R(U,V)X\}. \end{array}
$$

if both sides of this expression are multiplied by W , we have

$$
g(R(U,V)X,W) + g(X,V)g(U,W) - g(X,U)g(V,W)
$$

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$$
= L\{g(X, U)g(V, W) - g(X, V)g(U, W) + g(X, V)\eta(U)\eta(W) - g(X, U)\eta(V)\eta(W) - g(R(U, V)X, W)\},
$$
\n(21)

for all $W \in \Gamma(TM)$. Here replacing $X = V = e_1, e_2, ..., e_{n-1}, e_n = \xi$ in [\(21\)](#page-154-2) for the orthonormal basis of $\Gamma(TM)$ and by means of Ricci tensor, we get

$$
S(U, W) + (n-1)g(U, W) = L\{(1-n)g(U, W) + (n-1)\eta(U)\eta(W) - S(U, W)\},\
$$

After the necessary arrangements are made, we conclude

$$
S(U, Z) + (n - 1)g(U, Z) = L\left\{(1 - n)g(U, W) + (n - 1)\eta(U)\eta(W) - S(U, W)\right\},\
$$

that is,

$$
S(U, W) = (1 - n)g(U, W) + (n - 1)\frac{L}{L + 1}\eta(U)\eta(W).
$$
 (22)

If it is a semisymmetric, then $L = 0$ and (21) takes form

$$
R(U,V)X = g(X,U)V - g(X,V)U.
$$

This tells us that M has a constant sectional curvature 1. Conversely, if it has a constant sectional curvature 1, then we have

$$
(R(X,Y)R)(U,V,Z) = R(X,Y)R(U,V)Z - R(R(X,Y)U,V)Z - R(U,R(X,Y)V)Z
$$

\n
$$
= R(U,V)R(X,Y)Z
$$

\n
$$
= R(X,Y)\{g(U,Z)V - g(V,Z)U\} - R(g(X,U)Y - g(Y,U)X,V)Z
$$

\n
$$
= R(U,g(X,V)Y - g(Y,V)X)Z - R(U,V)\{g(X,Z)Y - g(Y,Z)X\}
$$

\n
$$
= g(Z,U)\{g(X,V)Y - g(Y,V)X\} - g(V,Z)\{g(X,U)Y - g(Y,U)X\}
$$

\n
$$
= g(X,U)\{g(Y,Z)V - g(Y,Z)U\} + g(Y,U)\{g(U,Z)X - g(X,Z)U\}
$$

\n
$$
= g(X,Z)\{g(U,Y)V - g(Y,V)U\} + g(Y,Z)\{g(U,X)V - g(V,X)U\}
$$

\n
$$
= 0.
$$

This completes the proof. $\hfill \square$

Now, we will calculate $M(X, Y)\xi$ for later use. From $\boxed{8}$ - $\boxed{11}$, we obtain

$$
\mathcal{M}(X,Y)\xi = \frac{1}{2}\{\eta(X)Y - \eta(Y)X\} + \frac{1}{2(n-1)}\{\eta(X)QY - \eta(Y)QX\},
$$
(23)

$$
\mathcal{M}(\xi,Y)Z = \frac{1}{2}\{\eta(Z)Y - g(Y,Z)\xi\} - \frac{1}{2(n-1)}\{S(Y,Z)\xi - \eta(Z)QY\}
$$
(24)

and

$$
\eta(\mathcal{M}(X,Y)Z) = \frac{1}{2(n-1)} \{ \eta(Y)S(X,Z) - \eta(X)S(Y,Z) \} + \frac{1}{2} \{ \eta(Y)g(X,Z) - \eta(X)g(Y,Z) \},
$$
\n(25)

$$
\mathcal{M}(\xi, X)Y = \frac{1}{2} \{ \eta(Y)X - g(X, Y)\xi \} + \frac{1}{2(n-1)} \{ \eta(Y)QX - S(X, Y)\xi \}.
$$
\n(26)

Theorem 3. A normal paracontact metric manifold M^n satisfying $\mathcal{M} \cdot R = 0$ is an Einstein manifold.

Proof. We suppose that $(\mathcal{M}(X, Y) \cdot R)(U, V, Z) = 0$, for any $X, Y, Z, U, V \in \Gamma(TM)$. This implies that

$$
\mathcal{M}(X,Y)R(U,V)Z = R(\mathcal{M}(X,Y)U,V)Z - R(U,M(X,Y)V)Z
$$

$$
= R(U,V)\mathcal{M}(X,Y)Z = 0.
$$
 (27)

Putting $Y = Z = \xi$ in $\left(\overline{27}\right)$, we obtain

$$
\mathcal{M}(X,\xi)R(U,V)\xi = R(\mathcal{M}(X,\xi)U,V)\xi - R(U,\mathcal{M}(X,\xi)V)\xi
$$

= R(U,V)\mathcal{M}(X,\xi)\xi = 0.

By using (9) and (24) , we conclude

$$
\frac{1}{2}g(X,V)U + \frac{1}{2(n-1)}S(X,V)U + \frac{1}{2}R(U,V)X + \frac{1}{2(n-1)}R(U,V)QX = 0.
$$
\n(28)

Taking the inner product with ξ , we reach

$$
\eta(U)S(X,V) + (n-1)\eta(U)g(X,V) + (n-1)\{\eta(V)g(X,U) - \eta(U)g(X,V)\} + \eta(V)S(X,U) - \eta(U)S(X,V) = 0,
$$

that is,

$$
S(X, U) = (1 - n)g(X, U).
$$

This proves our assertion. \Box

Definition 1. A semi-Riemannian manifold (M, g) is said to be the mathcalM projective pseudosymmetric if there exists a function L on M such that

$$
R \cdot \mathcal{M} = LQ(g, \mathcal{M}),
$$

where R and M denote the Riemannian and M - projectively curvature tensors of M. If $L = 0$, it also called the M-projectively semisymmetric.

Theorem 4. An M-projective pseudosymmetric normal paracontact manifold $M^n(\varphi, \eta, \xi, g)$ is an η -Einstein manifold.

Proof. Let us take M-projective pseudosymmetric normal paracontact manifold $M^n(\varphi, \eta, \xi, g)$. From $\overline{\left[5\right]}, \overline{\left[6\right]},$ we have

$$
-L\left\{M((X \wedge_g Y)U,V)Z + M(U,(X \wedge_g Y)V)Z + M(U,V)(X \wedge_g Y)Z\right\}
$$

= $R(X,Y)M(U,V)Z - M(R(X,Y)U,V)Z$
- $M(U,R(X,Y)V)Z - M(U,V)R(X,Y)Z,$ (29)

for all $X, Y, U, V, Z \in \Gamma(TM)$. Setting $X = Z = \xi$ in (29) , by using (23) - (26) , we have

$$
- L\{\frac{1}{2}[g(Y, U)V - g(Y, V)U + g(V, Y)\eta(U)\xi - g(Y, U)\eta(V)\xi] + \frac{1}{2(n-1)}[g(Y, U)QV - g(Y, V)QU + g(Y, V)\eta(U)Q\xi] - g(Y, U)\eta(V)Q\xi] - M(U, V)Y\} = \frac{1}{2}[g(Y, U)V - g(Y, V)U] + \frac{1}{2(n-1)}[\eta(V)S(Y, U)\xi - \eta(U)S(V, Y)\xi + g(Y, U)QV - \eta(V)g(Y, U)Q\xi + \eta(U)g(Y, V)Q\xi - g(Y, V)QU] - M(U, V)Y.
$$

If both sides of this equality are multiplied by W and by means of definition of the Ricci tensor, we obtain

$$
- L\left\{\frac{1}{2}[g(Y, U)g(V, W) - g(Y, V)g(U, W) + g(V, Y)\eta(U)\eta(W)\right\} - g(Y, U)\eta(V)\eta(W)] + \frac{1}{2(n-1)}[g(Y, U)S(V, W) - g(Y, V)S(U, W)\right]+ g(Y, V)\eta(U)S(\xi, W) - g(Y, U)\eta(V)S(\xi, W)] - g(\mathcal{M}(U, V)Y, W)\right\}= \frac{1}{2}[g(Y, U)g(V, W) - g(Y, V)g(U, W)]+ \frac{1}{2(n-1)}\left[\eta(V)S(Y, U)\eta(W) - \eta(U)S(V, Y)\eta(W)\right]+ g(Y, U)S(V, W) - \eta(V)g(Y, U)S(\xi, W) + \eta(U)g(Y, V)S(\xi, W)- g(Y, V)S(U, W)\right] - g(\mathcal{M}(U, V)Y, W).
$$

Here taking trace boht of sides for $Y = V = e_i$, for $1 \le i \le n$, in the last equality,

$$
-L\sum_{i=1}^{n} \left\{ \frac{1}{2} [\epsilon_i g(e_i, U) g(e_i, W) - \epsilon_i g(e_i, e_i) g(U, W) + \epsilon_i g(e_i, e_i) \eta(U) \eta(W) - \epsilon_i g(e_i, U) \eta(e_i) \eta(W)] + \frac{1}{2(n-1)} [\epsilon_i g(e_i, U) S(e_i, W) - \epsilon_i g(e_i, e_i) S(U, W) + \epsilon_i g(e_i, e_i) \eta(U) S(\xi, W) - \epsilon_i g(e_i, U) \eta(e_i) S(\xi, W)] - \epsilon_i g(M(U, e_i)e_i, W) \right\}
$$

\n
$$
= \sum_{i=1}^{n} \epsilon_i \left\{ \frac{1}{2} [\epsilon_i g(e_i, U) g(e_i, W) - \epsilon_i g(e_i, e_i) g(U, W)] + \frac{1}{2(n-1)} [\epsilon_i \eta(e_i) S(e_i, U) \eta(W) - \epsilon_i \eta(U) S(e_i, e_i) \eta(W) + \epsilon_i g(e_i, U) S(e_i, W) - \epsilon_i \eta(e_i) g(e_i, U) S(\xi, W) + \epsilon_i \eta(U) g(e_i, e_i) S(\xi, W) - \epsilon_i g(e_i, e_i) S(U, W)] - \epsilon_i g(M(U, e_i)e_i, W) \right\},
$$
(30)

where ϵ_i is the signature $\{e_i\}$. On the other hand, by direct calculations, we have

$$
\epsilon_i g(M(U, e_i)e_i, W) = \frac{1}{2(n-1)} \{n.S(U, W) - \tau.g(U, W)\}.
$$

Making use of (30) and after the necessary arrangements are revised, we get

$$
S(U, W) = \frac{(1-n)(n-1) + \tau}{2n-1} g(U, W) + \frac{n(1-n) - \tau}{(2n-1)(1+L)} \eta(U)\eta(W),
$$

which proves the theorem. $\hfill \square$

Definition 2. A normal paracontact manifold $M^n(\varphi, \eta, \xi, g)$ is said to be the Weylpseudosymmetric if there exists a function L on M such that

$$
R \cdot C = LQ(g, C),
$$

where R and C denote the Riemannian and Weyl-conformal curvature tensors of M. If $L = 0$, then it also called the Weyl-semisymmetric.

Now, we consider the Weyl-conformal curvature tensor of M^n given by (12) for later use.

$$
C(X,Y)\xi = \left(\frac{1-n-\tau}{(n-1)(n-2)}\right)(\eta(X)Y - \eta(Y)X) + \frac{1}{n-2}(\eta(X)QY - \eta(Y)QX)
$$
\n(31)

and

$$
C(\xi, X)Y = \left(\frac{1 - n - \tau}{(n - 1)(n - 2)}\right)(\eta(Y)X - g(X, Y)\xi) + \frac{1}{n - 2}(\eta(Y)QX - S(X, Y)\xi).
$$
 (32)

Theorem 5. The Weyl-pseudosymmetric normal paracontact metric manifold $M^n(\varphi, \eta, \xi, g)$ is an η -Einstein manifold.

Proof. Let $M^n(\varphi, \eta, \xi, g)$ be the Weyl-pseudosymmetric, then there is a function L such that

$$
(R(X,Y)\cdot C)(U,V,Z)=LQ(g,C)(U,V,Z;X,Y),
$$

for all $X, Y, U, V, Z \in \Gamma(TM)$. This implies that

$$
R(X,Y)C(U,V)Z - C(R(X,Y)U,V)Z - C(U,R(X,Y)V)Z
$$

$$
- C(U,V)R(X,Y)Z = -L\Big\{C((X \wedge_g Y)U,V)Z
$$

$$
+ C(U,(X \wedge_g Y)V)Z + C(U,V)(X \wedge_g Y)Z\Big\}.
$$
(33)

Here setting $X = Z = \xi$ in [\(33\)](#page-159-0), we have

$$
R(\xi, Y)C(U, V)\xi - C(R(\xi, Y)U, V)\xi - C(U, R(\xi, Y)V)\xi
$$

-
$$
C(U, V)R(\xi, Y)\xi = -L\Big\{C((\xi \wedge_g Y)U, V)\xi
$$

+
$$
C(U, (\xi \wedge_g Y)V)\xi + C(U, V)(\xi \wedge_g Y)\xi\Big\}.
$$
 (34)

After the necessary calculations, we reach at

$$
\frac{1-n-\tau}{(n-1)(n-2)}\{g(Y,U)V - g(Y,V)U\} \n+ \frac{1}{n-2}\{\eta(V)S(Y,U)\xi - \eta(U)S(V,Y)\xi \n+ g(Y,U)QV - \eta(V)g(Y,U)Q\xi \n+ \eta(U)g(Y,V)Q\xi - g(Y,V)QU\} - C(U,V)Y \n= -L\{\frac{1-n-\tau}{(n-1)(n-2)}(g(Y,U)V - g(Y,V)U - g(Y,U)\eta(V)\xi \n+ g(Y,V)\eta(U)\xi) + \frac{1}{n-2}(g(Y,U)QV - g(Y,V)QU \n- \eta(V)g(Y,U)Q\xi + \eta(U)g(Y,V)Q\xi) - C(U,V)Y\}.
$$
\n(35)

If both sides of the equality are multiplied by W , we obtain

$$
\frac{1-n-\tau}{(n-1)(n-2)}\{g(Y,U)g(V,W)-g(Y,V)g(U,W)\}
$$

+
$$
\frac{1}{n-2} \{\eta(V)S(Y,U)\eta(W) - \eta(U)S(Y,Y)\eta(W)
$$

+
$$
g(Y,U)S(V,W) - \eta(V)g(Y,U)S(\xi,W)
$$

+
$$
\eta(U)g(Y,V)S(\xi,W) - g(Y,V)S(W,U)\}
$$

-
$$
g(C(U,V)Y,W)
$$

=
$$
-L\{\frac{1-n-\tau}{(n-1)(n-2)}(g(Y,U)g(V,W) - g(Y,V)g(U,W)
$$

-
$$
g(Y,U)\eta(V)\eta(W) + g(Y,V)\eta(U)\eta(W))
$$

+
$$
\frac{1}{n-2}(g(Y,U)S(V,W) - g(Y,V)S(U,W)
$$

-
$$
\eta(V)g(Y,U)S(\xi,W) + \eta(U)g(Y,V)S(\xi,W))
$$
 (36)

Putting $Y = V = e_1, e_2, ...e_{n-1}, e_n = \xi$ in [\(36\)](#page-159-1) for the orthonormal basis of $\Gamma(TM)$ and taking into account definition of Ricci tensor, we have

$$
\frac{1-n-\tau}{(n-1)(n-2)} \sum_{i=1}^{n} \left\{ \epsilon_i \{ g(e_i, U)g(e_i, W) - \epsilon_i g(e_i, e_i)g(U, W) \} \right\}
$$
\n
$$
+ \frac{1}{n-2} \{ \epsilon_i \eta(e_i)S(e_i, U)\eta(W) - \epsilon_i \eta(U)S(e_i, e_i)\eta(W)
$$
\n
$$
+ \epsilon_i g(e_i, U)S(e_i, W) - \epsilon_i \eta(e_i)g(e_i, U)S(\xi, W)
$$
\n
$$
+ \epsilon_i \eta(U)g(e_i, e_i)S(\xi, W) - \epsilon_i g(e_i, e_i)S(W, U) \}
$$
\n
$$
- \epsilon_i g(C(U, e_i)e_i, W) \right\}
$$
\n
$$
= -L \{ \frac{1-n-\tau}{(n-1)(n-2)} \sum_{i=1}^{n} \left\{ \epsilon_i (g(e_i, U)g(e_i, W) - \epsilon_i g(e_i, e_i)g(U, W) - \epsilon_i g(e_i, U)\eta(e_i)\eta(W) + \epsilon_i g(e_i, e_i)\eta(U)\eta(W)) \right\}
$$
\n
$$
+ \frac{1}{n-2} (\epsilon_i g(e_i, U)S(e_i, W) - \epsilon_i g(e_i, e_i)S(U, W) - \epsilon_i \eta(e_i)g(e_i, U)S(\xi, W) + \epsilon_i \eta(U)g(e_i, e_i)S(\xi, W))
$$
\n
$$
- \epsilon_i g(C(U, e_i)e_i, W) \} \bigg\}.
$$
\n(37)

By using (11) and after the necessary abbreviations, (37) implies that

$$
S(U, W) = (1 - \frac{\tau}{n-1})g(U, W) - (n + \frac{\tau}{n-1})\eta(U)\eta(W).
$$

This proves our assertion. \Box

Now, we will give an non-trivial example for illustration our results.

Example 1. Let us the 5-dimensional manifold

$$
M^5 = \{(x_1, x_2, x_3, x_4, x_5) : x_i \in R, \}
$$

where (x_i) denote the cartesian coordinate in \mathbb{R}^5 for $1 \leq i \leq 5$. Then the vector fields

$$
e_1=\frac{\partial}{\partial x_1},e_2=\frac{\partial}{\partial x_2},e_3=2x_2\frac{\partial}{\partial x_1}+\frac{\partial}{\partial x_3},e_4=2x_3\frac{\partial}{\partial x_1}+\frac{\partial}{\partial x_4},e_5=-2x_4\frac{\partial}{\partial x_1}+\frac{\partial}{\partial x_4}
$$

are linearly independent at each point of M^5 . By g, we denote the semi-Riemannian metric tensor such that

$$
g(e_i, e_j) = \begin{cases} 1; & i = j = 1, 3, 4 \\ -1; & i = j = 2, 5 \\ 0; & i \neq j \end{cases}
$$

Let η be the 1-form defined by $\eta(X) = g(X, e_1)$ for all $X \in \Gamma(TM)$. Now, we definite the paracontact metric structure φ such that

$$
\varphi e_1 = 0
$$
, $\varphi e_2 = -e_3$, $\varphi e_3 = -e_2$, $\varphi e_4 = -e_5$, $\varphi e_5 = -e_4$.

Then we can easily see that

$$
\eta(e_5) = 1, \quad \varphi^2 X = X - \eta(X)\xi, \quad e_5 = \xi
$$

and

$$
g(\varphi X, \varphi Y) = -g(X, Y) + \eta(X)\eta(Y)
$$

for all $X, Y \in \Gamma(\overline{M})$. Thus $M^5(\varphi, \eta, \xi, g)$ defines an almost paracontact metric manifold. By $\widetilde{\nabla}$, we denote the Levi-Civita connection on \widetilde{M} . Then by direct calculations, we have non-zero the components

$$
[e_2, e_3] = 2e_1
$$
, $[e_3, e_4] = 2e_1$, $[e_4, e_5] = -2e_1$.

Let ∇ be the Levi-Civita connection on M. Using the properties of paracontact metric structure and Kozsul formulae, we can observe the non-zero components

 $\nabla e_2e_1 = -e_3 = \varphi e_2, \quad \nabla e_3e_1 = -e_2 = \varphi e_3, \quad \nabla e_4e_1 = -e_5 = \varphi e_4, \quad \nabla e_5e_1 = -e_4 = \varphi e_5$ Thus one can easily verified

$$
\widetilde{\nabla}_X e_1 = \varphi X,
$$

for all $X \in \Gamma(TM)$ This tells us that $M^5(\varphi, \eta, \xi, g)$ is a normal paracontact metric manifold with paracontact metric structure (φ, η, ξ, g) . By straightforward calculations, we can easily see that non-zero components of the Riemannian curvature tensor R,

$$
R(e_i, e_1)e_1 = -e_i, \ \ 2 \le i \le 5.
$$

This tell us that

$$
R(X,Y)Z = g(X,Z)Y - g(Y,Z)X,
$$

for all $X, Y, Z \in \Gamma(TM)$, that is, $\widetilde{M}(\varphi, \eta, \xi, g)$ is real space form with constant sectional curvature 1.

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APPROXIMATION PROPERTIES OF BERNSTEIN'S SINGULAR INTEGRALS IN VARIABLE EXPONENT LEBESGUE SPACES ON THE REAL AXIS

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ABSTRACT. In generalized Lebesgue spaces $L^{p(\cdot)}$ with variable exponent $p(\cdot)$ defined on the real axis, we obtain several inequalities of approximation by integral functions of finite degree. Approximation properties of Bernstein singular integrals in these spaces are obtained. Estimates of simultaneous approximation by integral functions of finite degree in $L^{p(\cdot)}$ are proved.

1. Introduction

In this work we consider approximation properties of Bernstein's singular integrals for functions given in the variable exponent Lebesgue spaces $L^{p(x)}(\mathbb{R})$. This scale of function spaces were studied in detail in books Uribe-Fiorenza [\[15\]](#page-183-0), Diening, Harjulehto, Hästö, Růžička $[17]$ and Sharapudinov $[40]$. $L^{p(x)}(\mathbb{R})$ has many applications in several branches of mathematics such as elasticity theory [\[50\]](#page-184-1), fluid mechanics [\[38\]](#page-184-2), [\[37\]](#page-184-3), differential operators [\[38\]](#page-184-2), [\[18\]](#page-183-2), nonlinear Dirichlet boundary value problems [\[32\]](#page-183-3), nonstandard growth [\[50\]](#page-184-1) and variational calculus. Variable exponent works started with W. Orlicz [\[35\]](#page-184-4) and developed in many directions. For example, $L^{p(x)}(\mathbb{R})$ is a modular space (**33**) and under the condition $p^+ := esssup_{x \in \mathbb{R}} p(x) < \infty$, $L^{p(x)}(\mathbb{R})$ becomes a particular case of the Musielak-Orlicz spaces [\[33\]](#page-183-4). Starting from nineties, studies on $L^{p(x)}(\mathbb{R})$ has reached a positive momentum: See $\boxed{32}$, $\boxed{39}$, $\boxed{20}$, $\boxed{16}$ and many others.

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Keywords. Modulus of smoothness, simultaneous approximation, Bernstein singular integral, forward Steklov mean, mollifiers, Jackson inequality, entire integral functions of finite degree. $r_{\text{rakgun@balikesir.edu.tr;}}$ 0000-0001-6247-8518.

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In variable exponent Lebesgue spaces on $[0, 2\pi]$ (or $[0, 1]$), some fundamental results corresponding to the approximation of function have been obtained by Shara-pudinov [\[41](#page-184-6)]–45. Some results on approximation in $L^{p(x)}([0, 2\pi])$ or other function classes can be seen e.g. in $[1]3-6.8, 9.19, 21-25.27-30.48$.

In this work, we aim to obtain simultaneous theorems on approximation by entire functions of finite degree in variable exponent Lebesgue spaces on the whole real axis R.

Approximation by entire function of finite degree in the real axis started by the works of Bernstein $[11, 12]$ $[11, 12]$, N. Wiener and R. Paley $[36]$, N. I. Ahiezer $[2]$, S. M. Nikolskii [\[34\]](#page-183-12). Note that an entire function of finite exponential type is merely an entire function of order 1 and finite type that in approximation theory, these often play an important role similar to trigonometric polynomials in the case of approximation of periodic functions.

Note that, some results on approximation by entire integral functions of finite degree were obtained by Ibragimov $\sqrt{26}$ and Taberski $\sqrt{46}$, $\sqrt{47}$ in the classical Lebesgue spaces $L^p(\mathbb{R})$.

We can give some required definitions. We denote by P the class of exponents $p(x): \mathbb{R} \to [1,\infty)$ such that $p(x)$ is a measurable function and $p(x)$ satisfy conditions

$$
1 \le p_- := \operatorname{essinf}_{x \in \mathbb{R}} p(x), \quad p^+ := \operatorname{esssup}_{x \in \mathbb{R}} p(x) < \infty. \tag{1}
$$

We define $L^{p(\cdot)} := L^{p(\cdot)}(\mathbb{R})$ as the set of all functions $f : \mathbb{R} \to \mathbb{C}$ such that

$$
I_{p(\cdot)}\left(\frac{f}{\lambda}\right) := \int_{\mathbb{R}} \left|\frac{f(y)}{\lambda}\right|^{p(y)} dy < \infty
$$
 (2)

for some $\lambda > 0$. The set of functions $L^{p(\cdot)}$, with norm

$$
||f||_{p(\cdot)} := \inf \left\{ \eta > 0 : I_{p(\cdot)} \left(\frac{f}{\eta} \right) < 1 \right\}
$$

is a Banach space.

For $p \in P$ we define its conjugate $p'(x) := \frac{p(x)}{p(x)-1}$ for $p(x) > 1$ and $p'(x) := \infty$ for $p(x) = 1$.

For $i \in \mathbb{N}$, all constants c_i (or c) will be some positive numbers such that c_i will depend on main parameters of the problem. In some cases we will use temporaryly some generic constans $C, c > 0$ for clarity (for example in statements of some theorems). We will give explicit constants in the proofs but these constants are not best constants.

Throughout this paper symbol $\mathfrak{A} \leq \mathfrak{B}$ will mean that there exists a constant C depending only on unimportant parameters in question such that inequality $\mathfrak{A} \leq C\mathcal{B}$ holds.

Definition 1. Let P^{Log} be a subclass ($\overline{17}$) of P such that there exist constants $c_1, c_2 > 0, c_3 \in \mathbb{R}$ with properties

$$
|p(x) - p(y)| \ln (e + 1/|x - y|) \le \mathbf{c}_1 < \infty, \quad \forall x, y \in \mathbb{R},\tag{3}
$$

$$
|p(x) - \mathbf{c}_3| \ln(e + |x|) \le \mathbf{c}_2 < \infty, \quad \forall x \in \mathbb{R}.\tag{4}
$$

2. Transference Result

Let C_0^{∞} be class of infinitely times continuously differentiable functions ϕ with compact support $spt\phi := \overline{\{x \in \mathbb{R} : \phi(x) \neq 0\}}$. Let $C(A)$ be the class of continuous functions defined on A. Define $||f||_{C(A)} := \sup \{|f(x)| : x \in A\}$ for $f \in C(A)$.

For given $f \in L^{p(\cdot)}$ we can define an auxiliary function as follows: Define

$$
F_f(u) := F_{f,G}(u) := \int_{\mathbb{R}} f(u+x) |G(x)| dx, \quad u \in \mathbb{R},
$$
\n(5)

where $G \in L^{p'(\cdot)} \cap C_0^{\infty}$ and $||G||_{p'(\cdot)} \leq 1$. Also we set $\mathbf{c}_0 := ||G||_{C(\mathbb{R})}$.

Theorem 1. Let $p \in P^{Log}$ and $f,g \in L^{p(\cdot)}$. If

$$
||F_{f,G}||_{C(\mathbb{R})} \lesssim ||F_{g,G}||_{C(\mathbb{R})},
$$

with an absolute positive constant, then, we have following norm inequality

$$
||f||_{p(\cdot)} \lesssim ||g||_{p(\cdot)}
$$

with a positive constant depending only on p.

3. MOLLIFIERS AND FORWARD STEKLOV MEANS IN $L^{p(\cdot)}$

Definition 2. Suppose that $0 < \delta < \infty$ and $\tau \in \mathbb{R}$. We define ($\overline{44}$) family of translated Steklov operators $\{\mathcal{S}_{\delta,\tau}f\}$, by

$$
S_{\delta,\tau}f(x) := \frac{1}{\delta} \int_{x+\tau-\delta/2}^{x+\tau+\delta/2} f(t) dt, \quad x \in \mathbb{R}
$$
 (6)

for locally integrable function f defined on R.

Let f and g be two real-valued measurable functions on \mathbb{R} . We define the convolution $f * g$ of f and g by setting $(f * g)(x) = \int_{\mathbb{R}} f(y)g(x - y)dy$ for $x \in \mathbb{R}$ for which the integral exists in R.

The following result on mollifiers in variable exponent Lebesgue spaces is obtained by D. Cruz-Uribe and A. Fiorenza (see $[14]$).

Definition 3. Let $\phi \in L^1(\mathbb{R})$ and $\int_{\mathbb{R}} \phi(t) dt = 1$. For each $t > 0$ we define $\phi_t(x) = \frac{1}{t} \phi\left(\frac{x}{t}\right)$. Such sequence $\{\phi_t\}$ will be called approximate identity. A function

$$
\phi^{^{\sim}}(x) = \sup_{|y| \geq |x|} |\phi(y)|
$$

will be called radial majorant of ϕ . If $\phi \in L^1(\mathbb{R})$, then, sequence $\{\phi_t\}$ will be called potential-type approximate identity.

Theorem 2. ($\overline{[14]}$ $\overline{[14]}$ $\overline{[14]}$) Suppose $p \in P^{Log}$, $f \in L^{p(\cdot)}$, ϕ is a potential-type approximate *identity.* Then, for any $t > 0$,

$$
\|f * \phi_t\|_{p(\cdot)} \lesssim \|f\|_{p(\cdot)}
$$

and

$$
\lim_{t\to 0}\left\|f\ast\phi_t-f\right\|_{p(\cdot)}=0
$$

hold with a positive constant depend on p.

As a corollary of Theorem 1 we have

Theorem 3. Suppose that $p \in P^{Log}$, $0 < \delta < \infty$ and $\tau \in \mathbb{R}$. Then, the family of operators $\{S_{\delta, \tau} f\}$, defined by ϕ , is uniformly bounded (in δ and τ) in $L^{p(\cdot)}$, namely, for any $0 < \delta < \infty$ and $\tau \in \mathbb{R}$ norm inequality

$$
\|\mathcal{S}_{\delta,\tau}f\|_{p(\cdot)} \lesssim \|f\|_{p(\cdot)}
$$

holds with a positive constant depend on p.

As a corollary of Theorem [3](#page-167-0) we get

Corollary 1. Let $p \in P^{Log}$, $0 < \delta < \infty$, $f \in L^{p(\cdot)}$. If $\tau = \delta/2$ then,

$$
T_{\delta}f(x) := \mathcal{S}_{\delta,\delta/2}f(x) = \frac{1}{\delta} \int_0^{\delta} f(x+t) dt
$$

and

$$
||T_{\delta}f||_{p(\cdot)} \lesssim ||f||_{p(\cdot)}
$$

holds with a positive constant depend on p.

4. Modulus of Smoothness and K-functional

If $f \in L^{p(\cdot)}$ and $0 \leq \delta < \infty$, $r \in \mathbb{N}$, then

$$
\Omega_r(f,\delta)_{p(\cdot)} := ||(I - T_\delta)^r f||_{p(\cdot)} \lesssim ||f||_{p(\cdot)}.
$$
\n(7)

Here I is the identity operator. In what follows $W_r^{p(\cdot)}$, $r \in \mathbb{N}$, will be the class of functions $f \in L^{p(\cdot)}$ such that $f^{(r-1)}$ is absolutely continuous and $f^{(r)} \in L^{p(\cdot)}$.

Remark 1. For $p \in P^{Log}$, $f, g \in L^{p(\cdot)}$ and $0 \le \delta < \infty$, the modulus of smoothness $\left(\Omega_{r}\left(f,\delta\right)_{p\left(\cdot\right)}\right)$, has the following usual properties:

(i) Ω_r $(f, \delta)_{p(\cdot)}$ is non-negative; non-decreasing function of $\delta \geq 0$; (ii) $\Omega_r(f+g,\cdot)_{p(\cdot)} \leq \Omega_r(f,\cdot)_{p(\cdot)} + \Omega_r(g,\cdot)_{p(\cdot)};$ (iii) $\lim_{\delta \to 0^+} \Omega_r(f, \delta)_{p(\cdot)} = 0;$ (iv) Ω_r $(f, \delta)_{p(\cdot)} \lesssim \delta^r ||f^{(r)}||_{p(\cdot)}$ for $r \in \mathbb{N}$, $f \in W_r^{p(\cdot)}$ and $\delta > 0$.

Indeed: (ii) follows from definition. (iii) is follow from $\langle 7 \rangle$, (3.4) and Theorem 3.1 of $\boxed{7}$. (iv) follows from Lemma 3.2 of $\boxed{7}$. (i) follows from Lemma $\boxed{1}$ given below.

Definition 4. Define, for $f \in L^{p(\cdot)}$, $p \in P^{Log}$, and $\delta > 0$,

$$
(\Re \delta f)(\cdot) := \frac{2}{\delta} \int_{\delta/2}^{\delta} \left(\frac{1}{h} \int_{0}^{h} f(\cdot + t) dt \right) dh.
$$

Remark 2. Note that, for $0 < \delta < \infty$, $p \in P^{Log}$ we know from Corollary [1](#page-167-2) that

$$
\|\Re_\delta f\|_{p(\cdot)} \lesssim \|f\|_{p(\cdot)}
$$

and, hence, $f - \Re_{\delta} f \in L^{p(\cdot)}$ for $f \in L^{p(\cdot)}$.

We set $\mathfrak{R}_{\delta}^r f := (\mathfrak{R}_{\delta} f)^r$.

Lemma 1. Let
$$
0 < h \leq \delta < \infty
$$
, $p \in P^{Log}$ and $f \in L^{p(\cdot)}$. Then $\|(I - T_h) f\|_{p(\cdot)} \lesssim \|(I - T_\delta) f\|_{p(\cdot)}$ (8)

holds with a positive constant depend on p.

Lemma 2. Let $0 < \delta < \infty$, $p \in P^{Log}$ and $f \in L^{p(\cdot)}$. Then

$$
\left\| \left(I - \Re \delta\right) f \right\|_{p(\cdot)} \lesssim \left\| \left(I - T_{\delta}\right) f \right\|_{p(\cdot)}
$$

holds with a positive constant depend on p.

Remark 3. The function $\mathfrak{R}_{\delta} f$ is absolutely continuous and differentiable a.e. (almost everywhere) on $\mathbb R$ (see $\sqrt{43}$, (5.2) of Theorem 4...

The following lemma is obvious from definitions.

Lemma 3. Let $0 < \delta < \infty$, $p \in P^{Log}$ and $f \in W_1^{p(\cdot)}$. Then

$$
\frac{d}{dx}\Re s f = \Re s \frac{d}{dx}f \quad and \quad \frac{d}{dx}T_{\delta}f = T_{\delta}\frac{d}{dx}f \tag{9}
$$

a.e. on R.

Lemma 4. Let $0 < \delta < \infty$, $p \in P^{Log}$ and $f \in L^{p(\cdot)}$ be given. Then

$$
\delta \left\| \frac{d}{dx} \Re_{\delta} f \right\|_{p(\cdot)} \lesssim \left\| (I - T_{\delta}) f \right\|_{p(\cdot)}
$$
\n(10)

holds with a positive constant depend on p.

The following lemma can be proved using induction on r.

Lemma 5. Let $0 < \delta < \infty$, $r - 1 \in \mathbb{N}$, $p \in P^{Log}$, and $f \in L^{p(\cdot)}$ be given. Then

$$
\frac{d^r}{dx^r} \Re_\delta^r f = \frac{d}{dx} \Re_\delta \frac{d^{r-1}}{dx^{r-1}} \Re_\delta^{r-1} f.
$$

Modulus of smoothness $\Vert (I - T_{\delta})^r f \Vert_{p(\cdot)}$ and the K-functional

$$
K_r\left(f, \delta; L^{p(\cdot)}, W_r^{p(\cdot)}\right)_{p(\cdot)} := \inf_{g \in W_r^{p(\cdot)}} \left\{ \|f - g\|_{p(\cdot)} + \delta^r \left\| g^{(r)} \right\|_{p(\cdot)} \right\}
$$

are equivalent:

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Theorem 4. If $r \in \mathbb{N}$, $p \in P^{Log}$, $f \in L^{p(\cdot)}$, and $\delta > 0$, then

$$
\frac{\left\|\left(I - T_{\delta}\right)^{r} f\right\|_{p(\cdot)}}{K_{r}\left(f, \delta; L^{p(\cdot)}, W_{r}^{p(\cdot)}\right)_{p(\cdot)}} \approx 1\tag{11}
$$

holds for some positive constants depend on p, r.

5. Results on Simultaneous Approximation

Let $\mathcal{G}_{\sigma}(X)$ be the subclass of entire integral functions $f(z)$ of exponential type $\leq \sigma$ that belonging to X and

$$
A_{\sigma}(f)_X := \inf_{g} \{ \|f - g\|_X : g \in \mathcal{G}_{\sigma}(X) \}.
$$

Let $\mathcal C$ be the class of bounded uniformly continuous functions defined on $\mathbb R$. We set $\mathcal{G}_{\sigma,\infty} := \mathcal{G}_{\sigma}(\mathcal{C})$ and $\mathcal{G}_{\sigma,p(\cdot)} := \mathcal{G}_{\sigma}(L^{p(\cdot)}).$

Remark 4. ($\overline{[10]}$, definition given in (5.3)]) Let $\sigma > 0$, $1 \le p \le \infty$, $f \in L^p(\mathbb{R})$,

$$
\vartheta(x) := \frac{2}{\pi} \frac{\sin(x/2) \sin(3x/2)}{x^2}
$$

and

$$
J(f, \sigma) = \sigma \int_{\mathbb{R}} f(x - u) \,\vartheta(\sigma u) \, du
$$

be the delà Valèe Poussin operator ($\overline{10}$, definition given in (5.3)]). It is known (see (5.4)-(5.5) of $\overline{10}$) that, if $f \in L^p(\mathbb{R})$, $1 \leq p \leq \infty$, then, (i) $J(f,\sigma) \in \mathcal{G}_{2\sigma}(\overline{L^p}(\mathbb{R})),$

(*ii*) $J(g_{\sigma}, \sigma) = g_{\sigma}$ for any $g_{\sigma} \in \mathcal{G}_{\sigma} (L^p(\mathbb{R})),$

$$
(iii) \|\widetilde{J}(f,\sigma)\|_{L^p(\mathbb{R})} \leq \frac{3}{2} \|f\|_{L^p(\mathbb{R})},
$$

(iv) $(J(f,\sigma))^{(r)} = J(f^{(r)},\sigma)$ for any $r \in \mathbb{N}$ and $f \in W_r^p(\mathbb{R})$,

(v)
$$
||J(f, \frac{\sigma}{2}) - f||_{L^p(\mathbb{R})} \to 0
$$
 (as $\sigma \to \infty$) and hence

$$
\|\left(J\left(f,\frac{\sigma}{2}\right)\right)^{(k)} - f^{(k)}\|_{L^p(\mathbb{R})} \to 0 \text{ as } \sigma \to \infty,
$$

for $f \in W_r^p(\mathbb{R})$ and $1 \leq k \leq r$, (vi) $||f-J(f, \frac{\sigma}{2})||_{L^p(\mathbb{R})} \leq \frac{5\pi}{4} \frac{4^r}{\sigma^r} ||f^{(r)}||_{L^p(\mathbb{R})}$ for $f \in W_r^p(\mathbb{R})$.

Theorem 5. Let $p \in P^{Log}$, $\sigma > 0$, $r \in \mathbb{N}$ and $f \in W_r^{p(\cdot)}$. Then

$$
A_{\sigma} (f)_{p(\cdot)} \lesssim \frac{1}{\sigma^r} A_{\sigma} \left(f^{(r)} \right)_{p(\cdot)}
$$
 (12)

holds with a positive constant depend on p, r .

Theorem 6. Let $p \in P^{Log}$, $\sigma > 0$, $k \in \mathbb{N}$, $r \in \{0\} \cup \mathbb{N}$ and $f \in W_r^{p(\cdot)}$. Then

$$
A_{\sigma}(f)_{p(\cdot)} \lesssim \frac{1}{\sigma^r} \Omega_k \left(f^{(r)}, \frac{1}{\sigma} \right)_{p(\cdot)} . \tag{13}
$$

with positive constants depend on p, k, r .

Theorem 7. Let $p \in P^{Log}$, $\sigma > 0$ and $g_{\sigma} \in \mathcal{G}_{\sigma,p(\cdot)}$. Then, Bernstein's inequality

$$
\| (g_{\sigma})^{(r)} \|_{p(\cdot)} \lesssim \sigma^r \| g_{\sigma} \|_{p(\cdot)}
$$

holds with a positive constant depend on p, r .

Definition 5. [47], p.161] For $r, k \in \mathbb{N}$, $\sigma > 0$, we define

$$
g(\sigma, r, x) = \left(\frac{1}{x}\sin\frac{\sigma x}{2r}\right)^{2r}, \text{ and}
$$

$$
G(\sigma, r, k, \zeta) = \sum_{v=1}^{k} (-1)^{k-v} \frac{1}{v} {k \choose v} g\left(\sigma, r, \frac{\zeta}{v}\right).
$$

For $r \geq \frac{1}{2}(k+2)$ we set

$$
\gamma_{r,\sigma} := \int_{\mathbb{R}} \left(\frac{1}{t} \sin \frac{\sigma t}{2r} \right)^{2r} dt.
$$

Let us introduce the Bernstein's singular integral ($\sqrt{47}$, p.161)

$$
D_{\sigma,k}f(x) := \frac{(-1)^{k+1}}{\gamma_{r,\sigma}} \int_{\mathbb{R}} f(u)G(\sigma,r,k,u-x) dt \qquad (14)
$$

for $r, k \in \mathbb{N}$, $\sigma > 0$, and measurable complex valued f satisfying $\int_{\mathbb{R}} \frac{|f(u)|}{1+u^{2r}} du < \infty$.

Remark 5. It is well known that, if $r, k \in \mathbb{N}$, $\sigma \in (0, \infty)$, $r \geq \frac{1}{2}(k+2)$, then $D_{\sigma,k}f \in \mathcal{G}_{\sigma}\left(L^p\left(\mathbb{R}\right)\right)$ for $p \geq 1$. ($\left[\sqrt{4\eta}, p.161\right]$).

Lemma 6. If $r \in \mathbb{N}$, $\sigma \in (0, \infty)$, then,

(i) we have

$$
\gamma_{r,\sigma} = \frac{\sigma^{2r-1}}{(2r)^{2r-1}} \int_{\mathbb{R}} \left(\frac{\sin v}{v}\right)^{2r} dv
$$

(ii) (see, e.g.
$$
[13, (5)]
$$
)
\n
$$
\int_{\mathbb{R}} \left(\frac{\sin v}{v}\right)^{2r} dv = \frac{2\pi}{(2r-1)!2^{2r}} \left\{ (2r)^{2r-1} - {2r \choose 1} (2r-2)^{2r-1} + {2r \choose 2} (2r-4)^{2r-1} - \cdots \right\}
$$
\n(iii) and, as a result,
\n
$$
\gamma_{r,\sigma} = \frac{\sigma^{2r-1}}{(2r)^{2r-1}} b_r
$$

where b_r is the right hand side of equality in (ii), having r terms.

Define $[a] := \min\{n \in \mathbb{N} : n \ge a\}$ and $[\sigma] := \max\{n \in \mathbb{Z} : n \le \sigma\}$. We will take $r := \lceil \frac{1}{2} (k+2) \rceil$ in the next Theorems.

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Theorem 8. Let $p \in P^{Log}$, $k \in \mathbb{N}$, $\sigma > 0$, $f \in W_k^{p(\cdot)}$ $k^{(r)}$, then

$$
\|f - D_{\sigma,k}f\|_{p(\cdot)} \lesssim \frac{1}{\sigma^k} \left\|f^{(k)}\right\|_{p(\cdot)}
$$
\n(15)

holds with a positive constant depend on p, k .

Theorem 9. Let $p \in P^{Log}$, $k \in \mathbb{N}$, $\sigma > 0$. If $f \in L^{p(\cdot)}$, then

$$
||D_{\sigma,k}f||_{p(\cdot)} \lesssim ||f||_{p(\cdot)}
$$

holds with a positive constant depend on p, k .

Theorem 10. Let $p \in P^{Log}$, $k \in \mathbb{N}$, $\sigma > 0$, $f \in L^{p(\cdot)}$, then

$$
\left\|f - D_{\sigma,k}f\right\|_{p(\cdot)} \lesssim \Omega_k\left(f, \frac{1}{\sigma}\right)_{p(\cdot)}
$$

holds with a positive constant depend on p, k .

Corollary 2. By Theorem \overline{Q} , if $r, k \in \mathbb{N}$, $\sigma \in (0, \infty)$, $r \geq \frac{1}{2}(k+2)$, then $D_{\sigma,k}f \in$ $\mathcal{G}_{\sigma,p(\cdot)}$ for $p \in P^{Log}$ and $f \in L^{p(\cdot)}$.

Theorem 11. Let $r \in \mathbb{N}$, $p \in P^{Log}$, $\sigma > 0$ and $f \in W_r^{p(\cdot)}$. Then for all $k =$ $0, 1, \ldots, r$, there exist positive constants depending only on k, r and $p(\cdot)$ such that

$$
\left\| f^{(k)} - (g^*_{\sigma})^{(k)} \right\|_{p(\cdot)} \lesssim \frac{1}{\sigma^{r-k}} A_{\sigma} \left(f^{(r)} \right)_{p(\cdot)}
$$

holds for any $g_{\sigma}^* \in \mathcal{G}_{\sigma, p(\cdot)}$ satisfying $A_{\sigma}(f)_{p(\cdot)} = ||f - g_{\sigma}^*||_{p(\cdot)}$.

Theorem 12. Let $r, s \in \mathbb{N}$, $p \in P^{Log}$ and $f \in W_r^{p(\cdot)}$. Then there exists a $\Phi \in$ $\mathcal{G}_{2\sigma,p(\cdot)}$ such that for all $k = 0, 1, \ldots, r$ inequalities

$$
\left\|f^{(k)} - \Phi^{(k)}\right\|_{p(\cdot)} \lesssim \frac{1}{\sigma^{r-k}} \Omega_s \left(f^{(r)}, \frac{1}{\sigma}\right)_{p(\cdot)}
$$

are hold with a positive constant depending only on k, r and $p(\cdot)$.

Definition 6. Set $\sigma, \eta > 0$, $f \in L^1(\mathbb{R})$, $\Theta_{\eta} f(x, y) := f(x + \eta y)$ and

$$
B_{\sigma}f(x,t) := \int_{\mathbb{R}} \Theta_{\frac{2}{\sigma}} f(x,y) h(y,t) dy.
$$

Remark 6. The following theorem was poved in $\overline{31}$ for $\sigma = 2$ with three minor mistypes. For the sake of completeness here we will prove it when $\sigma > 0$.

Theorem 13. Suppose that $h(y, t)$, $y, t \in \mathbb{R}$, is positive measurable function with respect to y and

$$
\int_{\mathbb{R}} h(y, t) dy \lesssim 1, \quad \int_{\mathbb{R}} |yh'_{y}(y, t)| dy \lesssim 1
$$

with constants independent of t. If $\sigma > 0$ and $f \in L^1(\mathbb{R})$, then

$$
\sup_{t>0}|B_{\sigma}f(\cdot,t)|\lesssim Mf(\cdot)
$$

for $t > 0$ and a.e. on $\mathbb R$ where M f is the Hardy-Littlewood maximal function of f.

6. Proof of the Results

Let $C(A)$ be the class of continuous functions defined on A. For $r \in \mathbb{N}$, we define $C^{r}(A)$ consisting of every member $f \in C(A)$ such that the derivative $f^{(k)}$ exists and is continuous on A for $k = 1, ..., r$. We set $C^{\infty}(A) := \{ f \in C^{r}(A) \text{ for any } r \in \mathbb{N} \}.$ We denote by $C_c(A)$, the collection of real valued continuous functions on A and support of f is compact set in A. We define $C_c^r(A) := C^r(A) \cap C_c(A)$ for $r \in \mathbb{N}$ and $C_c^{\infty}(A) := C^{\infty}(A) \cap C_c(A)$. Let $L^p(A), 1 \leq p \leq \infty$ be the classical Lebesgue space of functions on A.

Definition 7. ($\boxed{17}$) Let $\mathbb{N} := \{1, 2, 3, \dots\}$ be natural numbers and $\mathbb{N}_0 := \mathbb{N} \cup \{0\}$. (a) A family Q of measurable sets $E \subset \mathbb{R}$ is called locally N-finite ($N \in \mathbb{N}$) if

$$
\sum_{E\in Q}\chi_{E}\left(x\right)\leq N
$$

almost everywhere in $\mathbb R$ where χ_U is the characteristic function of the set U.

(b) A family Q of open bounded sets $U \subset \mathbb{R}$ is locally 1-finite if and only if the sets $U \in Q$ are pairwise disjoint.

(c) Let $U \subset \mathbb{R}$ be a measurable set and

$$
A_U f := \frac{1}{|U|} \int\limits_U |f(t)| \, dt.
$$

(d) For a family Q of open sets $U \subset \mathbb{R}$ we define averaging operator by

$$
T_Q: L_{loc}^1 \to L^0,
$$

$$
T_Q f(x) := \sum_{U \in Q} \chi_U(x) A_U f, \quad x \in \mathbb{R},
$$

where L^0 is the set of measurable functions on \mathbb{R} .

(e) For a measurable set $A \subset \mathbb{R}$, symbol |A| will represent the Lebesgue measure of A.

Theorem 14. ($\overline{17}$) Suppose that $p \in P^{Log}$, and $f \in L^{p(\cdot)}$. If Q is 1-finite family of open bounded subsets of $\mathbb R$ having Lebesgue measure 1, then, the averaging operator T_Q is uniformly bounded in $L^{p(\cdot)}$, namely,

$$
||T_Qf||_{p(\cdot)} \le c_4 ||f||_{p(\cdot)}
$$

holds with a positive constant c_4 depending only on p.

We define $\langle f, g \rangle := \int_{\mathbb{R}} f(x)g(x)dx$ when integral exists. We will need the following Propositions.

Proposition 1. ($\boxed{17}$) Let $p \in P^{Log}$. Then

$$
\frac{1}{12\mathbf{c}_4}\left\|f\right\|_{p(\cdot)}\leq \sup_{g\in L^{p'(\cdot)}\cap C_0^\infty:\|g\|_{p'(\cdot)}\leq 1}\left\langle\left|f\right|,\left|g\right|\right\rangle\leq 2\left\|f\right\|_{p(\cdot)}
$$

holds for all $f \in L^{p(\cdot)}$.

Proposition 2. (a) $C_c(\mathbb{R})$ and $C_c^{\infty}(\mathbb{R})$ are dense subsets of $L^p(\mathbb{R})$, $1 \leq p <$ ∞.(Theorems 17.10 and 23.59 of $\frac{5}{49}$, p. 415 and p. 575]).

(b) $C_c(\mathbb{R})$ contained $L^{\infty}(\mathbb{R})$ but not dense (Remark 17.11 of $\overline{[49]}$, p.416)) in $L^{\infty}(\mathbb{R})$.

Theorem 15. Let $p \in P^{Log}$. In this case,

(a) if $f \in L^{p(\cdot)}$, then, the function $F_f := F_{f,G}$ defined in [\(5\)](#page-166-2) is a bounded, uniformly continuous function on R,

(b) if $r \in \mathbb{N}$, and $f \in W_r^{p(\cdot)}$, then, $\frac{d^k}{du^k}(F_f)$ exists and

$$
\frac{d^k}{du^k}\left(F_f\right) = F_{f^{(k)}}
$$

for $k \in \{1, ..., r\}$.

Proof. (a) Since C_0^{∞} is a dense subset of $L^{p(\cdot)}$, we consider functions $f \in C_0^{\infty}$ and its corresponding $F_{f,G}$ given in [\(5\)](#page-166-2). For any $\varepsilon > 0$, there exists $\delta := \delta(\varepsilon) > 0$ so that

$$
\left|f\left(x+u_1\right)-f\left(x+u_2\right)\right| < \frac{\varepsilon}{1+|sptG|}
$$

for any $u_1, u_2 \in \mathbb{R}$ with $|u_1 - u_2| < \delta$, where sptG is the support of the function $G \in L^{p'(\cdot)} \cap C_0^{\infty}$. Then, there holds inequality

$$
|F_{f,G}(u_1) - F_{f,G}(u_2)| \le \int_{\mathbb{R}} |f(x + u_1) - f(x + u_2)| |G(x)| dx
$$

=
$$
\int_{sptG} |f(x + u_1) - f(x + u_2)| |G(x)| dx
$$

$$
\le \sup_{x, u_1, u_2 \in sptG} |f(x + u_1) - f(x + u_2)| ||G||_{1, sptG}
$$

$$
\le \frac{\varepsilon}{1 + |sptG|} (1 + |sptG|) ||G||_{p'(\cdot)} \le \varepsilon
$$

for any $u_1, u_2 \in \mathbb{R}$ with $|u_1 - u_2| < \delta$. Thus conclusion of Theorem [15](#page-173-0) follows. For the general case $f \in L^{p(\cdot)}$ there exists an $g \in C_0^{\infty}$ so that

$$
\|f-g\|_{p(\cdot)} < \frac{\xi}{4\left(1+|sptG|\right)\mathbf{c}_0}
$$

for any $\xi > 0$. Therefore

$$
|F_{f,G}(u_1) - F_{f,G}(u_2)| = |F_{f,G}(u_1) - F_{g,G}(u_1)| + |F_{g,G}(u_1) - F_{g,G}(u_2)| +
$$

+ |F_{g,G}(u_2) - F_{f,G}(u_2)| = |F_{f-g,G}(u_1)| + \frac{\xi}{2} + |F_{g-f,G}(u_2)|

$$
\leq 2\left(1+|sptG|\right)\mathbf{c}_0\left\|f-g\right\|_{p(\cdot),\omega}+\frac{\xi}{2}<\xi.
$$

As a result $F_{f,G}$ is uniformly continuous on \mathbb{R} .

 (b) is follow from definitions.

Proof of Theorem $\boxed{1}$. Let $0 \leq f, g \in L^{p(\cdot)}$. In this case there exists a constant $C>0$ such that

$$
\|F_{f,G}\|_{C(\mathbb{R})} \le C \left\|F_{g,G}(u)\right\|_{C(\mathbb{R})} = C \left\| \int_{\mathbb{R}} g(u+x) |G(x)| dx \right\|_{C(\mathbb{R})}
$$

= $C \sup_{u \in \mathbb{R}} \int_{\mathbb{R}} g(u+x) |G(x)| dx = C \sup_{u \in sptG} \int_{sptG} g(u+x) |G(x)| dx$
 $\le C \sup_{u \in sptG} \|g(u+\cdot)\|_{1,sptG} \|G\|_{\infty} \le C (1+|sptG|) \mathbf{c}_0 \|g\|_{p(\cdot)}.$

On the other hand, for any $\varepsilon > 0$ and appropriately chosen $\tilde{G}_{\varepsilon} \in L^{p'(\cdot)}$ with

$$
\int_{\mathbb{R}} g(x) \, \tilde{G}_{\varepsilon}(x) \, dx \ge \frac{1}{12 \mathbf{c}_4} \left\| g \right\|_{p(\cdot)} - \varepsilon, \qquad \left\| \tilde{G}_{\varepsilon} \right\|_{p'(\cdot)} \le 1,
$$

(see Proposition $\vert \mathbf{1} \vert$), one can find

$$
||F_{f,G}||_{C(\mathbb{R})} \geq |F_{f,G}(0)| \geq \int_{\mathbb{R}} f(x) |G(x)| dx > \frac{1}{12c_4} ||f||_{p(\cdot)} - \varepsilon.
$$

In the last inequality we take as $\varepsilon \to 0^+$ and obtain

$$
\|F_{f,G}\|_{C(\mathbb{R})} > \frac{1}{12c_4} \|f\|_{p(\cdot)}.
$$

Combining these inequalities we get

$$
||f||_{p(\cdot)} < 12\mathbf{c}_4 ||F_{f,G}||_{C(\mathbb{R})} \le 12\mathbf{c}_4 C ||F_{g,G}||_{C(\mathbb{R})}
$$

$$
\le 12\mathbf{c}_4 C (1+|sptG|) \mathbf{c}_0 ||g||_{p(\cdot)}.
$$

For general case $f, g \in L^{p(\cdot)}$ we obtain

$$
||f||_{p(\cdot)} < 24\mathbf{c}_4 (1+|sptG|) \mathbf{c}_0 C ||g||_{p(\cdot)}
$$
\n(16)

and proof is finished. \Box

Remark 7. Note that, in $\overline{16}$ constant depend on |sptG| and $||G||_{\infty}$ but it is possible to avoid dependence on $|sptG|$ and $||G||_{\infty}$. To do so, we can change the definition of F_f with

$$
F_f(u) := \int_{\mathbb{R}} \mathcal{S}_{1,u} f(x) |G(x)| dx, \quad u \in \mathbb{R},
$$

where $G \in L^{p'(\cdot)} \cap C_0^{\infty}$ and $||G||_{p'(\cdot)} \leq 1$. Now, boundedness of $S_{1,u}f$ in $L^{p(\cdot)}$ for any $u \in \mathbb{R}$, and the same procedure give $\overline{16}$ with a constant does not depend on $|\text{sptG}|$ and $||G||_{\infty}$. Hence, constants in other results can be free of dependence on $|sptG|$ and $||G||_{\infty}$.

$$
\Box
$$

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Proof of Lemma [1](#page-168-0). Let $0 < h \le \delta < \infty$, $p \in P^{Log}$ and $f \in L^{p(\cdot)}$. Then, using (16) we get

$$
||(I - T_h) f||_{p(\cdot)} < 24\mathbf{c}_4 ||F_{(I - T_h)f, G}||_{C(\mathbb{R})} \le 24 \cdot 72\mathbf{c}_4 ||F_{(I - T_\delta)f, G}||_{C(\mathbb{R})}
$$

$$
\le 1728\mathbf{c}_4 (1 + |sptG|) \mathbf{c}_0 ||(I - T_\delta) f||_{p(\cdot)}.
$$

Proof of Lemma ². If $f \in L^{p(\cdot)}$, then, using generalized Minkowski's integral inequality and Lemma $\overline{1}$ we obtain

$$
\begin{split} \left\| (I - \Re_{\delta}) f \right\|_{p(\cdot)} &= \left\| \frac{2}{\delta} \int_{\delta/2}^{\delta} \left(\frac{1}{h} \int_{0}^{h} \left(f \left(x + t \right) - f \left(x \right) \right) dt \right) dh \right\|_{p(\cdot)} \\ &= \left\| \frac{2}{\delta} \int_{\delta/2}^{\delta} \left(T_{h} f \left(x \right) - f \left(x \right) \right) dh \right\|_{p(\cdot)} \leq \frac{2}{\delta} \int_{\delta/2}^{\delta} \left\| T_{\delta} f - f \right\|_{p(\cdot)} dh \\ &\leq 1728 \mathbf{c}_{4} \left(1 + |sptG| \right) \mathbf{c}_{0} \left\| T_{\delta} f - f \right\|_{p(\cdot)} \frac{2}{\delta} \int_{\delta/2}^{\delta} dh \\ &= 1728 \mathbf{c}_{4} \left(1 + |sptG| \right) \mathbf{c}_{0} \left\| (I - T_{\delta}) f \right\|_{p(\cdot)}. \end{split}
$$

□

Proof of Lemma [4](#page-168-2). Using

$$
\|F_{\delta(\Re\delta f)',G}\|_{C(\mathbb{R})} = \left\|\delta \left(F_{(\Re \delta f),G}\right)'\right\|_{C(\mathbb{R})} = \delta \left\| (\Re \delta(F_{f,G}))' \right\|_{C(\mathbb{R})}
$$

$$
\leq \dots \leq 2 \left(37 + 146 \ln 2^{36}\right) \left\| (I - T_{\delta})(F_{f,G}) \right\|_{C(\mathbb{R})}
$$

$$
= 2 \left(37 + 146 \ln 2^{36}\right) \left\| (F_{(I-T_{\delta})f,G}) \right\|_{C(\mathbb{R})}
$$

we conclude from Transference Result that

$$
\delta \|(\Re \delta f)' \|_{p(\cdot)} \leq \mathbf{c}_5 \left(I - T_\delta \right) f \|_{p(\cdot)}
$$

.

.

with $c_5 := 24c_4 (1 + |sptG|) c_0 (37 + 146 \ln 2^{36})$. □

Proof of Theorem [4](#page-169-0). For $r = 1, 2, 3, ...$ we consider the operator

$$
\mathcal{A}_{\delta}^r := I - (I - \mathfrak{R}_{\delta}^r)^r = \sum_{j=0}^{r-1} (-1)^{r-j+1} \binom{r}{j} \mathfrak{R}_{\delta}^{r(r-j)}
$$

From the identity $I - \mathfrak{R}_{\delta}^r = (I - \mathfrak{R}_{\delta}) \sum_{j=0}^{r-1} \mathfrak{R}_{\delta}^j$ we find

$$
\left\| \left(I - \mathfrak{R}_{\delta}^r \right) g \right\|_{p(\cdot)} \leq \left(\sum_{j=0}^{r-1} \mathbf{c}_{6}^j \right) \left\| \left(I - \mathfrak{R}_{\delta} \right) g \right\|_{p(\cdot)}
$$

with $\mathbf{c}_6 := 24\mathbf{c}_4 (1 + |sptG|) \mathbf{c}_0$. Therefore

$$
\left\| (I - \mathfrak{R}_{\delta}^{r}) g \right\|_{p(\cdot)} \leq \left(1728 \mathbf{c}_{4} \left(1 + |sptG| \right) \mathbf{c}_{0} \sum_{j=0}^{r-1} \mathbf{c}_{6}^{j} \right) \left\| (I - T_{\delta}) g \right\|_{p(\cdot)} \qquad (17)
$$

$$
= \mathbf{c}_{7} \left\| (I - T_{\delta}) g \right\|_{p(\cdot)}
$$

when $0 < \delta < \infty$, $p \in P$ and $g \in L^{p(\cdot)}$. Since $||f - \mathcal{A}_{\delta}^r f||_{p(\cdot)} = ||(I - \mathfrak{R}_{\delta}^r)^r f||_{p(\cdot)}$, recursive procedure gives

$$
||f - \mathcal{A}_{\delta}^r f||_{p(\cdot)} = ||(I - \mathfrak{R}_{\delta}^r)^r f||_{p(\cdot)} \leq \cdots \leq \mathbf{c}_7^r ||(I - T_{\delta})^r f||_{p(\cdot)}.
$$

On the other hand, using Lemmas $\overline{5}$ and $\overline{4}$,

$$
\delta^r \left\| \frac{d^r}{dx^r} \mathfrak{R}_{\delta}^r f \right\|_{p(\cdot)} = \delta^{r-1} \delta \left\| \frac{d}{dx} \mathfrak{R}_{\delta} \frac{d^{r-1}}{dx^{r-1}} \mathfrak{R}_{\delta}^{r-1} f \right\|_{p(\cdot)}
$$

$$
\leq \mathbf{c}_{5} \delta^{r-1} \left\| (I - T_{\delta}) \frac{d^{r-1}}{dx^{r-1}} \mathfrak{R}_{\delta}^{r-1} f \right\|_{p(\cdot)} \leq \cdots \leq
$$

$$
\leq \mathbf{c}_{5}^{r-1} \delta \left\| \frac{d}{dx} \mathfrak{R}_{\delta} (I - T_{\delta})^{r-1} f \right\|_{p(\cdot)} \leq \mathbf{c}_{5}^{r} \left\| (I - T_{\delta})^{r} f \right\|_{p(\cdot)}.
$$

Thus

$$
K_r\left(f, \delta; L^{p(\cdot)}, W_r^{p(\cdot)}\right)_{p(\cdot)} \leq ||f - A_\delta^r f||_{p(\cdot)} + \delta^r \left\| \frac{d^r}{dx^r} A_\delta^r f(x) \right\|_{p(\cdot)}
$$

\n
$$
\leq \mathbf{c}_7^r \left\| (I - T_\delta)^r f\right\|_{p(\cdot)} + \sum_{j=0}^{r-1} \left| \binom{r}{j} \delta^r \left\| \frac{d^r}{dx^r} \mathfrak{R}_\delta^{r(r-j)} f(x) \right\|_{p(\cdot)}
$$

\n
$$
\leq \mathbf{c}_7^r \left\| (I - T_\delta)^r f\right\|_{p(\cdot)} + \mathbf{c}_5^r \sum_{j=0}^{r-1} \left| \binom{r}{j} \right| \left\| (I - T_\delta)^r \mathfrak{R}_\delta^{r(j-j)} f \right\|_{p(\cdot)}
$$

\n
$$
\leq \mathbf{c}_7^r \left\| (I - T_\delta)^r f\right\|_{p(\cdot)} + \mathbf{c}_5^r \sum_{j=0}^{r-1} \left| \binom{r}{j} \right| \mathbf{c}_6^{r-j} \left\| (I - T_\delta)^r f\right\|_{p(\cdot)}
$$

\n
$$
\leq \mathbf{c}_8 \left\| (I - T_\delta)^r f\right\|_{p(\cdot)}
$$

where

$$
\mathbf{c}_8 := \max \left\{ \mathbf{c}_7^r, \mathbf{c}_5^r \sum_{j=0}^{r-1} \left| {r \choose j} \right| \mathbf{c}_6^{r-j} \right\}.
$$

For the reverse of the last inequality, when $g \in W_r^{p(\cdot)}$, we get

$$
\Omega_r(f, \delta)_{p(\cdot)} \le (1 + \mathbf{c}_6)^r \|f - g\|_{p(\cdot)} + \Omega_r(g, \delta)_{p(\cdot)}
$$

$$
\le (1 + \mathbf{c}_6)^r \|f - g\|_{p(\cdot)} + 2^{-r} \mathbf{c}_6^r \delta^r \|g^{(r)}\|_{p(\cdot)},
$$
 (18)

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and taking infimum on $g \in W_{p(\cdot)}^r$ in $\left(\sqrt{18}\right)$ we obtain

$$
\Omega_r(f,\delta)_{p(\cdot)} \le (1+\mathbf{c}_6)^r K_r\left(f,\delta;L^{p(\cdot)},W_r^{p(\cdot)}\right)_{p(\cdot)}.
$$

Proof of Theorem $\overline{5}$ **.** The following inequality

$$
A_{\sigma}(f)_{C(\mathbb{R})} \leq \left\| f - J\left(f, \frac{\sigma}{2}\right) \right\|_{C(\mathbb{R})} \leq \frac{5\pi}{4} \frac{4^{r}}{\sigma^{r}} \| f^{(r)} \|_{C(\mathbb{R})}, \quad \forall f \in C^{r}(\mathbb{R})
$$

known (see (vi) of Remark $\overline{4}$). Now using TR we find

$$
\left\|f-J\left(f,\frac{\sigma}{2}\right)\right\|_{p(\cdot)} \le \frac{5\pi}{2} \frac{4^r \mathbf{c}_6}{\sigma^r} \|f^{(r)}\|_{p(\cdot)}, \quad \forall f \in W_r^{p(\cdot)}.\tag{19}
$$

Let $r = 1$. Suppose that

$$
A_{\sigma}(f')_{p(\cdot)} = ||f' - g_{\sigma}^{*}(f')||_{p(\cdot)}, \quad g_{\sigma}^{*}(f') \in \mathcal{G}_{\sigma, p(\cdot)}
$$

and

$$
F(x) := \int_0^x g^*_{\sigma}(f')(t) dt.
$$

Then $\digamma \in \mathcal{G}_\sigma$ ([\[26,](#page-183-13) p.397]). Setting

$$
\varphi\left(x\right) = f\left(x\right) - F\left(x\right)
$$

one has

$$
\|\varphi'\|_{p(\cdot)} = \|f' - g^*_{\sigma}(f')\|_{p(\cdot)} = A_{\sigma}(f')_{p(\cdot)}.
$$

Thus

$$
A_{\sigma} (f)_{p(\cdot)} = A_{\sigma} (f - F)_{p(\cdot)} \stackrel{\text{(L3)}}{\leq} 10\pi \mathbf{c}_6 \frac{1}{\sigma} \left\| (f - F)' \right\|_{p(\cdot)}
$$

$$
= \frac{10\pi \mathbf{c}_6}{\sigma} \left\| f' - F' \right\|_{p(\cdot)} = \frac{10\pi \mathbf{c}_6}{\sigma} \left\| f' - g_{\sigma}^*(f') \right\|_{p(\cdot)}
$$

$$
= 10\pi \mathbf{c}_6 \frac{1}{\sigma} A_{\sigma} (f')_{p(\cdot)}.
$$

Now, result follows from the last inequality:

$$
A_{\sigma}(f)_{p(\cdot)} \leq 10\pi \mathbf{c}_6 \frac{1}{\sigma} A_{\sigma}(f')_{p(\cdot)} \leq \cdots \leq (10\pi \mathbf{c}_6)^r \frac{1}{\sigma^r} A_{\sigma}(f^{(r)})_{p(\cdot)}.
$$

Proof of Theorem $\overline{6}$. Let $p \in P^{Log}$, $\sigma > 0$, $k \in \mathbb{N}$, $r \in \{0\} \cup \mathbb{N}$ and $f \in W_r^{p(\cdot)}$. First we consider the case $r = 0$. For every $g \in W_k^{p(\cdot)}$ we find

$$
A_{\sigma}(f)_{p(\cdot)} \leq A_{\sigma}(f-g)_{p(\cdot)} + A_{\sigma}(g)_{p(\cdot)}
$$

$$
\leq ||f-g||_{p(\cdot)} + \frac{5\pi}{2} \frac{4^{k} \mathbf{c}_{6}}{\sigma^{k}} ||f^{(k)}||_{p(\cdot)}.
$$

Taking infimum on g in the last inequality

$$
A_{\sigma}(f)_{p(\cdot)} \leq \frac{5\pi}{2} 4^{k} \mathbf{c}_{6} K_{k}\left(f, \delta; L^{p(\cdot)}, W_{k}^{p(\cdot)}\right)_{p(\cdot)}.
$$

Now using (11)

$$
A_{\sigma}(f)_{p(\cdot)} \leq \mathbf{c}_8 \frac{5\pi}{2} 4^k \mathbf{c}_6 \Omega_k \left(f, \frac{1}{\sigma}\right)_{p(\cdot)}.
$$

In the second stage we consider the case $r \in \mathbb{N}$. In this case

$$
A_{\sigma} (f)_{p(\cdot)} \leq (10\pi\mathbf{c}_6)^r \frac{1}{\sigma^r} A_{\sigma} (f^{(r)})_{p(\cdot)}
$$

$$
\leq 5\pi\mathbf{c}_8 (10)^r \pi^r \mathbf{c}_6^{r+1} 2^{2k-1} \frac{1}{\sigma^r} \Omega_k (f^{(r)}, \frac{1}{\sigma})_{p(\cdot)}.
$$

Proof of Theorem [7](#page-170-0). Let $p \in P^{Log}$, $\sigma > 0$ and $g_{\sigma} \in \mathcal{G}_{\sigma,p(\cdot)}$. Then, Bernstein's inequality

$$
\| (g_{\sigma})^{(r)} \|_{C(\mathbb{R})} \leq \sigma^r \| g_{\sigma} \|_{C(\mathbb{R})}, \quad \forall g_{\sigma} \in \mathcal{G}_{\sigma, \infty}
$$

and TR gives

$$
\| (g_{\sigma})^{(r)} \|_{p(\cdot)} \leq \mathbf{c}_6 \sigma^{r} \| g_{\sigma} \|_{p(\cdot)}, \quad \forall g_{\sigma} \in \mathcal{G}_{\sigma, p(\cdot)}.
$$

Proof of Theorem ⁸. Define for $k \in \mathbb{N}$ the classical modulus of smoothness of function $f \in C(\mathbb{R})$ of step $\delta > 0$ by

$$
\omega_{k}(f,\delta)_{C(\mathbb{R})} := \sup_{|h| \leq \delta} ||\Delta_{t}^{k}f||_{C(\mathbb{R})}
$$

where $\Delta_t^k f(\cdot) := (I - \tilde{T}_h)^k f(\cdot), \tilde{T}_h f(\cdot) := f(\cdot + h)$ and I is the identity operator. From (14) , one can write

$$
\|f - D_{\sigma,k}f\|_{C(\mathbb{R})} = \left\| \frac{(-1)^k}{\gamma_{r,\sigma}} \int_{\mathbb{R}} \sum_{v=0}^k (-1)^{k-v} {k \choose v} f(x+vt) g(\sigma, r, t) dt \right\|_{C(\mathbb{R})}
$$

\n
$$
\leq \frac{1}{\sigma^{2r-1} \frac{b_r}{(2r)^{2r-1}}} \int_{\mathbb{R}} \left\| \Delta_t^k f(x) \right\|_{C(\mathbb{R})} g(\sigma, r, t) dt \leq \frac{(2r)^{2r-1}}{b_r \sigma^{2r-1}} \int_{\mathbb{R}} \omega_k (f, t)_{C(\mathbb{R})} g(\sigma, r, t) dt
$$

\n
$$
\leq \frac{(2r)^{2r-1} \sigma^k}{b_r \sigma^{2r-1}} \omega_k \left(f, \frac{1}{\sigma}\right)_{C(\mathbb{R})} \int_{\mathbb{R}} \left(t + \frac{1}{\sigma}\right)^k g(\sigma, r, t) dt
$$

\n
$$
\leq \frac{(2r)^{2r-1} \sigma^k}{b_r \sigma^{2r-1}} \frac{1}{\sigma^k} \left\| f^{(k)} \right\|_{C(\mathbb{R})} \int_{\mathbb{R}} \left(t + \frac{1}{\sigma}\right)^k g(\sigma, r, t) dt
$$

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$$
\leq \frac{(2r)^{2r-1}}{b_r\sigma^{2r-1}}\left\|f^{(k)}\right\|_{C(\mathbb{R})}\left\{\frac{2^k}{\sigma^k}\int\limits_{|t|\leq \frac{1}{\sigma}}\left|g\left(\sigma,r,t\right)\right|dt+2^k\int\limits_{|t|\geq \frac{1}{\sigma}}\left|t\right|^k\left|g\left(\sigma,r,t\right)\right|dt\right\}.
$$

Using $r = \lceil \frac{1}{2} (k+2) \rceil$

$$
\frac{(2r)^{2r-1} 2^k}{b_r \sigma^{2r-1}} \int_{|t| \ge 1/\sigma} |t|^k \left(\frac{1}{t} \sin \frac{\sigma t}{2r}\right)^{2r} dt
$$

$$
\le \frac{(2r)^{2r-1} 2^k}{b_r \sigma^{2r-1}} \int_{|t| \ge 1/\sigma} \left(\frac{1}{t} \sin \frac{\sigma t}{2r}\right)^{2r-k} dt
$$

$$
\le \frac{(2r)^{2r-1} 2^k}{b_r \sigma^{2r-1}} \frac{\sigma^{2r-k+1}}{(2r)^{2r-k+1}} \int_{\mathbb{R}} \left(\frac{\sin u}{u}\right)^2 dt = \frac{1}{\sigma^k} \frac{2^{2k} r^k}{b_r} \pi.
$$

On the other hand

$$
\frac{(2r)^{2r-1}}{b_r \sigma^{2r-1}} \frac{2^k}{\sigma^k} \int_{|t| \le 1/\sigma} \left(\frac{1}{t} \sin \frac{\sigma t}{2r}\right)^{2r} dt
$$

$$
\le \frac{(2r)^{2r-1}}{b_r \sigma^{2r-1}} \frac{2^k}{\sigma^k} \int_{\mathbb{R}} \left(\frac{1}{t} \sin \frac{\sigma t}{2r}\right)^{2r} dt
$$

$$
= \frac{(2r)^{2r-1}}{b_r \sigma^{2r-1}} \sigma^{2r-1} \frac{b_r}{(2r)^{2r-1}} = \frac{2^k}{\sigma^k}.
$$

Thus

$$
||f - D_{\sigma,k}f||_{C(\mathbb{R})} \le \left(\frac{2^{2k}r^k}{b_r} + 2^k\right) \frac{1}{\sigma^k} ||f^{(k)}||_{C(\mathbb{R})}.
$$

From this and TR we get

$$
\left\|f-D_{\sigma,k}f\right\|_{p(\cdot)}\leq\mathbf{c}_6\left(\frac{2^{2k}r^k}{b_r}+2^k\right)\frac{1}{\sigma^k}\left\|f^{(k)}\right\|_{p(\cdot)}=\mathbf{c}_6\mathbf{c}\left(k,r\right)\frac{1}{\sigma^k}\left\|f^{(k)}\right\|_{p(\cdot)}.
$$

Proof of Theorem 9. Fixed $\sigma > 0$, we find

$$
\|D_{\sigma,k}f\|_{C(\mathbb{R})} = \left\| \frac{(-1)^{k+1}}{\gamma_{r,\sigma}} \int_{\mathbb{R}} f(u)G(\sigma,r,k,u-x) du \right\|_{C(\mathbb{R})}
$$

$$
\left\| \frac{(-1)^{k+1}}{\gamma_{r,\sigma}} \int_{\mathbb{R}} \sum_{v=1}^{k} (-1)^{k-v} {k \choose v} f(u) g(\sigma,r, \frac{u-x}{v}) du \right\|_{C(\mathbb{R})}
$$

$$
\leq \left\| \frac{(-1)^{k+1}}{\gamma_{r,\sigma}} \int_{\mathbb{R}} \sum_{v=1}^{k} (-1)^{k-v} {k \choose v} f(x+vt) g(\sigma,r,t) v dt \right\|_{C(\mathbb{R})}
$$

$$
\leq \frac{k}{\gamma_{r,\sigma}} \int_{\mathbb{R}} \sum_{v=1}^{k} |{k \choose v} ||f(x+vt)||_{C(\mathbb{R})} g(\sigma,r,t) dt
$$
$$
\leq \|f\|_{C(\mathbb{R})} \sum_{v=1}^k \left| \binom{k}{v} \right| \frac{k}{\gamma_{r,\sigma}} \int_{\mathbb{R}} g\left(\sigma, r, t\right) dt \leq k 2^k \|f\|_{C(\mathbb{R})}.
$$

Now, transference result TR gives

$$
\|D_{\sigma,k}f\|_{p(\cdot)} \leq k2^k \mathbf{c}_6 \, \|f\|_{p(\cdot)} \, .
$$

Proof of Theorem [10](#page-171-0). We can write

$$
\|f - D_{\sigma,k}f\|_{p(\cdot)} = \left\|f - \mathcal{A}_{\frac{1}{\sigma}}^k f + \mathcal{A}_{\frac{1}{\sigma}}^k f - D_{\sigma,k} \mathcal{A}_{\frac{1}{\sigma}}^k f + D_{\sigma,k} \mathcal{A}_{\frac{1}{\sigma}}^k f - D_{\sigma,k} f\right\|_{p(\cdot)}
$$

\n
$$
\leq \left\|f - \mathcal{A}_{\frac{1}{\sigma}}^k f\right\|_{p(\cdot)} + \left\|\mathcal{A}_{\frac{1}{\sigma}}^k f - D_{\sigma,k} \mathcal{A}_{\frac{1}{\sigma}}^k f\right\|_{p(\cdot)} + \left\|D_{\sigma,k} (\mathcal{A}_{\frac{1}{\sigma}}^k f - f)\right\|_{p(\cdot)}
$$

\n
$$
\leq \mathbf{c}_7^k \Omega_k \left(f, \frac{1}{\sigma}\right)_{p(\cdot)} + \mathbf{c}_6 \mathbf{c}(k) \frac{1}{\sigma^k} \left\|(\mathcal{A}_{\frac{1}{\sigma}}^k f)^{(k)}\right\|_{p(\cdot)} + k2^k \mathbf{c}_6 \left\|\mathcal{A}_{\frac{1}{\sigma}}^k f - f\right\|_{p(\cdot)}
$$

\n
$$
\leq \left(\mathbf{c}_7^k + \mathbf{c}_6 \mathbf{c}(k, r) \mathbf{c}_5^k \sum_{j=0}^{k-1} \left|\binom{k}{j}\right| \mathbf{c}_6^{k-j} + 2^k k \mathbf{c}_6 \mathbf{c}_7^k \right) \Omega_k \left(f, \frac{1}{\sigma}\right)_{p(\cdot)}
$$

\n
$$
= \mathbf{c}_9 \Omega_k \left(f, \frac{1}{\sigma}\right)_{p(\cdot)}
$$

and the result follows. $\hfill \Box$

Proof of Theorem [11] Let
$$
q \in \mathcal{G}_{\sigma}
$$
 and $A_{\sigma} (f^{(k)})_{p(\cdot)} = ||f^{(k)} - q||_{p(\cdot)}$. Then
\n
$$
||f^{(k)} - (g_{\sigma}^{*})^{(k)}||_{p(\cdot)} \le ||f^{(k)} - (J(f, \sigma))^{(k)}||_{p(\cdot)} + ||(J(f, \sigma))^{(k)} - (g_{\sigma}^{*})^{(k)}||_{p(\cdot)}
$$
\n
$$
\le ||f^{(k)} - q||_{p(\cdot)} + ||q - J(f^{(k)}, \sigma)||_{p(\cdot)} + ||(J(f, \sigma) - g_{\sigma}^{*})^{(k)}||_{p(\cdot)}
$$
\n
$$
\le A_{\sigma} (f^{(k)})_{p(\cdot)} + ||J(g - f^{(k)}, \sigma)||_{p(\cdot)} + 2^{k}c_{6}\sigma^{k} ||J(f, \sigma) - g_{\sigma}^{*}||_{p(\cdot)}
$$
\n
$$
\le (1 + 3c_{6}) A_{\sigma} (f^{(k)})_{p(\cdot)} + 2^{k}c_{6}\sigma^{k} ||J(f, \sigma) - J(g_{\sigma}^{*}, \sigma)||_{p(\cdot)}
$$
\n
$$
\le (1 + 3c_{6}) \frac{2c_{6} (5\pi 4^{r-1})^{r}}{\sigma^{r-k}} A_{\sigma} (f^{(r)})_{p(\cdot)} + 3c_{6}^{2}2^{k} \frac{2c_{6} (5\pi 4^{r-1})^{r}}{\sigma^{r-k}} A_{\sigma} (f^{(r)})_{p(\cdot)}
$$
\n
$$
\le (2c_{6} (5\pi 4^{r-1})^{r}) (1 + 3c_{6} + 3c_{6}^{2}2^{k}) \frac{\sigma^{k}}{\sigma^{r}} A_{\sigma} (f^{(r)})_{p(\cdot)} = c_{10}\sigma^{k-r} A_{\sigma} (f^{(r)})_{p(\cdot)}
$$
\nand the proof of Theorem [11] is completed.

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Proof of Theorem [12](#page-171-2). Let $g_{\sigma}^* \in \mathcal{G}_{\sigma}$, $A_{\sigma}(f)_{p(\cdot)} = ||f - g_{\sigma}^*||_{p(\cdot)}$ and $\Phi = J(f, \sigma)$. Then

$$
\begin{aligned} \|f - J\left(f, \sigma\right)\|_{p(\cdot)} &\le \|f - g_{\sigma}^* + g_{\sigma}^* - J\left(f, \sigma\right)\|_{p(\cdot)} \\ &= \|f - g_{\sigma}^* + J\left(g_{\sigma}^*, \sigma\right) - J\left(f, \sigma\right)\|_{p(\cdot)} \\ &\le A_{\sigma}\left(f\right)_{p(\cdot)} + 3\mathbf{c}_6 \left\|f - g_{\sigma}^*\right\|_{p(\cdot)} = \left(1 + 3\mathbf{c}_6\right) A_{\sigma}\left(f\right)_{p(\cdot)} \end{aligned}
$$

and

$$
\|f-J(f,\sigma)\|_{p(\cdot)} \leq (1+3\mathbf{c}_6) A_{\sigma}(f)_{p(\cdot)}
$$

$$
\leq (1+3\mathbf{c}_6) 5\pi \mathbf{c}_8 (10)^r \pi^r \mathbf{c}_6^{r+1} 2^{2s-1} \frac{1}{\sigma^r} \Omega_s \left(f^{(r)}, 1/\sigma\right)_{p(\cdot)}.
$$

Now, from

$$
\left\|f-g^{*}_{\sigma}\right\|_{p(\cdot)} \leq \frac{\pi \mathbf{c}_8 \left(10\right)^r \pi^r \mathbf{c}_6^{r+1} 2^{2s-1} }{\sigma^r} \Omega_s \left(f^{(r)},\frac{1}{\sigma}\right)_{p(\cdot)}
$$

we obtain

$$
\|J(f,\sigma)-g_{\sigma}^*\|_{p(\cdot)} \leq \frac{c_{11}}{\sigma^r} \Omega_s \left(f^{(r)},\frac{1}{\sigma}\right)_{p(\cdot)}
$$

with

$$
\mathbf{c}_{11} = \pi \mathbf{c}_8 (10)^r \pi^r \mathbf{c}_6^{r+1} 2^{2s-1} ((1+3\mathbf{c}_6) 5 + 1).
$$

Hence

$$
\left\|f^{(k)} - (J(f, \sigma))^{(k)}\right\|_{p(\cdot)} \le \left\|f^{(k)} - (g_{\sigma}^*)^{(k)}\right\|_{p(\cdot)} + \left\|(J(f, \sigma))^{(k)} - (g_{\sigma}^*)^{(k)}\right\|_{p(\cdot)}
$$

$$
\le c_{10}\sigma^{k-r}A_{\sigma}\left(f^{(r)}\right)_{p(\cdot)} + 2^k c_6\sigma^k \frac{c_{11}}{\sigma^r}\Omega_s\left(f^{(r)}, \frac{1}{\sigma}\right)_{p(\cdot)}
$$

$$
\le \left(c_{10}\frac{5\pi c_8}{2}4^s c_6 + 2^k c_6 c_{11}\right)\sigma^{k-r}\Omega_s\left(f^{(r)}, 1/\sigma\right)_{p(\cdot)}
$$

and the proof is completed.

Proof of Theorem [13](#page-171-3). Given $x \in \mathbb{R}$, let

$$
\Gamma(y) := \int_0^y \Theta_{\frac{2}{\sigma}} f(x, u) du, \quad y > 0,
$$

and $a, b > 0$. Integration by parts gives

$$
\int_{-a}^{b} \Theta_{\frac{2}{\sigma}} f(x, y) h(y, t) dy = \int_{-a}^{b} h(y, t) d\Gamma(y)
$$

$$
= \Gamma(y) h(y, t) \Big|_{-a}^{b} - \int_{-a}^{b} h'_y(y, t) \Gamma(y) dy.
$$

Since $\Gamma\left(y\right) \leq \left|y\right|Mf\left(x\right)$ we obtain

$$
\left|\int_{-a}^{b} \Theta_{\frac{2}{\sigma}} f(x, y) h(y, t) dy\right| \leq M f(x) \left(\int_{-a}^{b} \left|yh_{y}'(y, t)\right| dy + h(y, t) \Big|_{-a}^{b}\right).
$$

Now

$$
\mathbf{c}_{12} \ge \int_{\mathbb{R}} h(y, t) dy \ge \int_{-a}^{b} h(y, t) dy = h(y, t) \Big|_{-a}^{b} - \int_{-a}^{b} y h'_{y}(y, t) dy
$$

gives

$$
\left| \int_{-a}^{b} \Theta_{\frac{2}{\sigma}} f(x, y) h(y, t) dy \right| \leq \left(c_{12} + 2c_{13} \right) M f(x)
$$

for any $t > 0$. The last inequality implies the result. \Box

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SPECTRAL SINGULARITIES OF AN IMPULSIVE STURM–LIOUVILLE OPERATORS

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Abstract. In this paper, we handle an impulsive Sturm–Liouville equation with complex potential on the semi axis. The objective of this work is to examine some spectral properties of this impulsive Sturm–Liouville equation. By the help of a transfer matrix B , we obtain Jost solution of this problem. Furthermore, using Jost solution, we find Green function and resolvent operator of this equation. Finally, we consider two unperturbated impulsive Sturm– Liouville operators. We examine the eigenvalues and spectral singularities of these problems.

1. INTRODUCTION

The modeling of most of the problems encountered in the fields of mathematics, physics, mechanics and engineering in daily life is done with boundary value or initial value problems in applied mathematics and spectral analysis. Operator theory is used to solve these problems in spectral theory. First, many physicists and mathematicians studied the spectral theory of differential operators. The Sturm–Liouville operator, which is the equivalent of the one dimensional Schrödinger operator, has gained a wide place in the literature. Let us shortly give information about the literature of spectral theory of Sturm–Liouville operator. Spectral analysis of the nonself-adjoint Schrödinger operator was first investigated by Naimark in 1960 $\boxed{20}$. He proved that the spectrum of this operator consists of eigenvalues, continuous spectrum and spectral singularities. Furthermore, he discovered that the spectral singularities are poles of the resolvent operator's kernel on the continuous spectrum but not the eigenvalues of the operator. Kemp extended the results obtained by

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Naimark to differential operators defined on the whole real axis [\[16\]](#page-199-1). Gasymov also extended these results to three-dimensional Schrödinger operators $[12]$. Then, Schwartz investigated the spectral singularities of a certain class of abstract linear operators in Hilbert space and proved that self-adjoint operators have no spectral singularity [\[23\]](#page-199-3). Furthermore, these equations were studied under different initial and boundary conditions by Pavlov, Guseinov and Bairamov et al. $\|\mathbf{7}\|\mathbf{9}\|\mathbf{10}\|\mathbf{14}\|\mathbf{22}\|$.

On the other hand, in some processes, instant changes are encountered due to external factors. These are short term sudden changes and can be neglected compared to the whole process. Ordinary differential equations are not sufficient to model these processes. For this reason, impulsive differential equations are used to explain these processes mathematically. Unlike the Schrödinger equation, differential equations with impulsive conditions do not have a long history in the literature. Impulsive differential systems were first studied by Myshkis and Mil'man [\[18\]](#page-199-8). After, these equations were investigated by Bainov, Simenov and Lakshmikantham $\mathbb{E}[\mathbf{A}]$. Recently, many authors have examined impulsive differential equations in detail, because impulsive differential equations have been used in many scientific phenomena such as heart beat, population dynamics, atomic physics, mathematical economics, ecology, engineering, medicine and so forth $\boxed{13, 15, 19}$ $\boxed{13, 15, 19}$ $\boxed{13, 15, 19}$. Bairamov et al, Yardimci and Erdal investigated scattering analysis and spectral theory of different kinds of impulsive Sturm–Liouville equations $\left[2, 5, 6, 8, 11, 24\right]$ $\left[2, 5, 6, 8, 11, 24\right]$ $\left[2, 5, 6, 8, 11, 24\right]$ $\left[2, 5, 6, 8, 11, 24\right]$ $\left[2, 5, 6, 8, 11, 24\right]$ $\left[2, 5, 6, 8, 11, 24\right]$ $\left[2, 5, 6, 8, 11, 24\right]$. Different from these studies, in this paper, we consider the Sturm–Liouville equation with complex valued potential and impulsive condition in matrix form. Therefore, it creates different perspective.

Let us introduce the Sturm–Liouville operator T in $L_2(0,\infty)$, generated by the equation

$$
-v'' + q(z)v = \lambda^2 v, \quad z \in [0, z_0) \cup (z_0, \infty)
$$
 (1)

with the boundary condition

$$
(\eta_0 + \eta_1 \lambda) v'(0) + (\zeta_0 + \zeta_1 \lambda) v(0) = 0
$$
\n
$$
(2)
$$

and the impulsive condition

$$
\begin{bmatrix} \upsilon (z_0^+) \\ \upsilon'(z_0^+) \end{bmatrix} = B \begin{bmatrix} \upsilon (z_0^-) \\ \upsilon'(z_0^-) \end{bmatrix}, \qquad B = \begin{bmatrix} \beta_1 & \beta_2 \\ \beta_3 & \beta_4 \end{bmatrix}, \tag{3}
$$

where β_i , η_j , ζ_j , $i = 1, 2, 3, 4$, $j = 0, 1$ are complex numbers such that $\det B \neq 0$ and $\eta_0 \zeta_1 - \eta_1 \zeta_0 \neq 0$, z_0 is a positive real constant and q is a complex valued function satisfying the following condition

$$
\int_{0}^{\infty} (1+z)|q(z)|dz < \infty.
$$
 (4)

Throughout the paper, we will show impulsive boundary value problem $\left(\prod\right)$ - $\left(\overline{3}\right)$ by ISBVP, shortly.

This paper is organized as follows: This study consists of five chapters including

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the introduction. In the next Section, we give basic solutions and definitions. Unlike other studies in the literature, we examine the effect of the impulsive condition on the Sturm–Liouville equation with complex potential in Section 3. We find the Jost solution of ISBVP (1) - (3) . In Section 4, we obtain the set of eigenvalues and spectral singularities of $\left(\frac{1}{2}\right)$. Also, we present an asymptotic equation to obtain the properties of eigenvalues. Then, we get the resolvent operator of the Sturm– Liouville operator T. Finally, we handle two different problems to apply our main results in Section 5.

2. Preliminaries

Let $S(z, \lambda^2)$ and $C(z, \lambda^2)$ be the fundamental solutions of Π in the interval $[0, z_0)$ satisfying the initial conditions

$$
S(0, \lambda^2) = 0, \quad S'(0, \lambda^2) = 1,
$$

$$
C(0, \lambda^2) = 1, \quad C'(0, \lambda^2) = 0.
$$

It is evident that the solutions $S(z, \lambda^2)$ and $C(z, \lambda^2)$ are entire functions of λ and

$$
W[S(z, \lambda^2), C(z, \lambda^2)] = -1, \qquad \lambda \in \mathbb{C},
$$

where $W[v_1, v_2]$ denotes the Wronskian of the solutions v_1 and v_2 of the equation $\[\Pi\]$. The integral representations of $S(z, \lambda^2)$ and $C(z, \lambda^2)$ are well known in the literature as

$$
S(z, \lambda^2) = \frac{\sin \lambda z}{\lambda} + \int_0^z Q(z, t) \frac{\sin \lambda t}{\lambda} dt
$$
 (5)

$$
C(z, \lambda^2) = \cos \lambda z + \int_0^z R(z, t) \cos \lambda t dt,
$$
\n(6)

where $Q(z, t)$ and $R(z, t)$ are expressed in terms of the potential function q [\[17\]](#page-199-15).

On the other hand, $e(z, \lambda)$ is bounded solution of the equation Π in the interval (z_0, ∞) fulfilling the following condition

$$
\lim_{z \to \infty} e(z, \lambda) e^{-i\lambda z} = 1, \quad \lambda \in \overline{\mathbb{C}}_+ := \{ \lambda \in \mathbb{C} : \text{Im} \lambda \ge 0 \}
$$

and it has an integral representation

$$
e(z,\lambda) = e^{i\lambda z} + \int\limits_z^\infty K(z,t)e^{i\lambda t}dt, \quad \lambda \in \overline{\mathbb{C}}_+, \tag{7}
$$

where $K(z, t)$ is defined by the potential function q $\left| \mathbf{l} \right|$. The bounded solution $e(z, \lambda)$ is analytic with respect to λ in $\mathbb{C}_+ := {\lambda \in \mathbb{C} : Im \lambda > 0}$ and continuous up to the real axis. Similarly, $e(z, -\lambda)$ is bounded solution of $\overline{1}$ in (z_0, ∞) satisfying the condition

$$
\lim_{z \to \infty} e(z, -\lambda) e^{i\lambda z} = 1, \quad \lambda \in \overline{\mathbb{C}}_- := \{ \lambda \in \mathbb{C} : \text{Im} \lambda \le 0 \}.
$$

It is well known that

$$
W\left[e\left(z,\lambda\right),e\left(z,-\lambda\right)\right]=-2i\lambda,\qquad\lambda\in\mathbb{R}\backslash\{0\}.
$$

Furthermore, $\breve{e}(z, \lambda)$ is unbounded solution of $[1]$ in (z_0, ∞) subjecting the conditions [\[21\]](#page-199-16)

$$
\lim_{z \to \infty} \breve{e}(z, \lambda) e^{i\lambda z} = 1, \qquad \lim_{z \to \infty} \breve{e}'(z, \lambda) e^{i\lambda z} = -i\lambda, \qquad \lambda \in \overline{\mathbb{C}}_+.
$$

It is clear that

$$
W\left[e\left(z,\lambda\right),\breve{e}\left(z,\lambda\right)\right]=-2i\lambda,\qquad z\in\left(z_0,\infty\right),\qquad\lambda\in\overline{\mathbb{C}}_+.
$$

3. Solutions of Impulsive Sturm–Liouville Equation

By the help of linearly independent solutions (1) , we will define the general solutions of $\left(\overline{1} \right)$ for $\lambda \in \mathbb{R} \setminus \{0\},$

$$
\Psi_1(z,\lambda) = \begin{cases} v_1^-(z,\lambda) = a^-(\lambda)S(z,\lambda^2) + b^-(\lambda)C(z,\lambda^2); & 0 \le z < z_0 \\ v_1^+(z,\lambda) = a^+(\lambda)e(z,\lambda) + b^+(\lambda)e(z,-\lambda); & z_0 < z < \infty, \end{cases}
$$
 (8)

$$
\Psi_2(z,\lambda) = \begin{cases}\nv_2^-(z,\lambda) = c^-(\lambda)S(z,\lambda^2) + d^-(\lambda)C(z,\lambda^2); & 0 \le z < z_0 \\
v_2^+(z,\lambda) = c^+(\lambda)e(z,\lambda) + d^+(\lambda)e(z,-\lambda); & z_0 < z < \infty\n\end{cases} \tag{9}
$$

and for $\lambda\in\overline{\mathbb{C}}_+\backslash\{0\},$

$$
\Psi_3(z,\lambda) = \begin{cases}\nv_3^-(z,\lambda) = f^-(\lambda)S(z,\lambda^2) + h^-(\lambda)C(z,\lambda^2) & 0 \le z < z_0 \\
v_3^+(z,\lambda) = f^+(\lambda)e(z,\lambda) + h^+(\lambda)\breve{e}(z,\lambda) & z_0 < z < \infty,\n\end{cases} \tag{10}
$$

respectively.

Using (3) and (8) , we obtain

$$
\begin{bmatrix} a^+(\lambda) \\ b^+(\lambda) \end{bmatrix} = N \begin{bmatrix} a^-(\lambda) \\ b^-(\lambda) \end{bmatrix},\tag{11}
$$

.

where

$$
N := \begin{bmatrix} N_{11}(\lambda) & N_{12}(\lambda) \\ N_{21}(\lambda) & N_{22}(\lambda) \end{bmatrix} = L^{-}BM
$$
\n(12)

such that

$$
L = \begin{bmatrix} e(z_0, \lambda) & e(z_0, -\lambda) \\ e'(z_0, \lambda) & e'(z_0, -\lambda) \end{bmatrix}
$$

and

$$
M = \begin{bmatrix} S(z_0, \lambda^2) & C(z_0, \lambda^2) \\ S'(z_0, \lambda^2) & C'(z_0, \lambda^2) \end{bmatrix}
$$

Since det $L = -2i\lambda$, in accordance with $\sqrt{12}$, we find that

$$
N_{21}(\lambda) = \frac{i}{2\lambda} \left[-e'(z_0, \lambda) \left(\beta_1 S(z_0, \lambda^2) + \beta_2 S'(z_0, \lambda^2) \right) + e(z_0, \lambda) \left(\beta_3 S(z_0, \lambda^2) + \beta_4 S'(z_0, \lambda^2) \right) \right]
$$
(13)

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$$
N_{22}(\lambda) = \frac{i}{2\lambda} \left[-e'(z_0, \lambda) \left(\beta_1 C(z_0, \lambda^2) + \beta_2 C'(z_0, \lambda^2) \right) + e(z_0, \lambda) \left(\beta_3 C(z_0, \lambda^2) + \beta_4 C'(z_0, \lambda^2) \right) \right].
$$
 (14)

Now, we shall consider the Jost solution of ISBVP (I) - (3) and denote by E. Thus, by using $\boxed{8}$, the coefficients $a^+(\lambda)$ and $b^+(\lambda)$ turn into 1 and 0, respectively. For $\lambda \in \overline{\mathbb{C}}_+$, we write the following solution of (1) - (3)

$$
E(z,\lambda) = \begin{cases} a^-(\lambda)S(z,\lambda^2) + b^-(\lambda)C(z,\lambda^2); & z \in [0,z_0) \\ e(z,\lambda); & z \in (z_0,\infty). \end{cases}
$$

By the help of (11) and (12) , we easily obtain the coefficients $a^{-}(\lambda)$ and $b^{-}(\lambda)$ as follows

$$
a^{-}(\lambda) = \frac{N_{22}(\lambda)}{\det N}, \qquad b^{-}(\lambda) = -\frac{N_{21}(\lambda)}{\det N}.
$$
 (15)

Let us consider the solution of $\left(\frac{1}{2}\right)$ satisfying the boundary condition $\left(\frac{2}{2}\right)$ and denote by F . By (2) and (9) , the following can be easily seen

$$
c^-(\lambda) = (\zeta_0 + \zeta_1 \lambda), \qquad d^-(\lambda) = (\eta_0 + \eta_1 \lambda).
$$

For $\lambda \in \mathbb{R} \setminus \{0\}$, we will consider the following solution of ISBVP [\(1\)](#page-186-0)-[\(3\)](#page-186-1)

$$
F(z,\lambda) = \begin{cases} -(\zeta_0 + \zeta_1 \lambda) S(z,\lambda^2) + (\eta_0 + \eta_1 \lambda) C(z,\lambda^2); & z \in [0, z_0) \\ c^+(\lambda)e(z,\lambda) + d^+(\lambda)e(z,-\lambda); & z \in (z_0, \infty). \end{cases}
$$

From (3) and (12) , we get

$$
c^{+}(\lambda) = -(\zeta_0 + \zeta_1 \lambda) N_{11}(\lambda) + (\eta_0 + \eta_1 \lambda) N_{12}(\lambda)
$$
 (16)

$$
d^{+}(\lambda) = -(\zeta_0 + \zeta_1 \lambda) N_{21}(\lambda) + (\eta_0 + \eta_1 \lambda) N_{22}(\lambda), \qquad (17)
$$

respectively.

Lemma 1. For $\lambda \in \mathbb{R} \setminus \{0\}$, the Wronskian of the solutions $E(z, \lambda)$ and $F(z, \lambda)$ is given by

$$
W[E(z,\lambda), F(z,\lambda)] = \begin{cases} H(\lambda); & z \in [0, z_0) \\ 2i\lambda H(\lambda) \det N; & z \in (z_0, \infty), \end{cases}
$$

where

$$
H(\lambda) := \frac{(\zeta_0 + \zeta_1 \lambda) N_{21}(\lambda) - (\eta_0 + \eta_1 \lambda) N_{22}(\lambda)}{\det N}.
$$
 (18)

Proof. Using the definition of Wronskian for $z \in [0, z_0)$, we find

$$
W\left[E\left(z,\lambda\right),F\left(z,\lambda\right)\right]=-\left(\zeta_0+\zeta_1\lambda\right)b^-\left(\lambda\right)-\left(\eta_0+\eta_1\lambda\right)a^-\left(\lambda\right).
$$

By using (15) , the following can be easily seen

$$
W\left[E\left(z,\lambda\right),F\left(z,\lambda\right)\right]=H(\lambda)
$$

for $z \in [0, z_0)$. Similarly, we write

$$
W\left[E\left(z,\lambda\right),F\left(z,\lambda\right)\right]=-2i\lambda d^+(\lambda), \quad z\in\left(z_0,\infty\right).
$$

By the help of (17) , it is clear that

$$
W\left[E\left(z,\lambda\right),F\left(z,\lambda\right)\right]=2i\lambda H\left(\lambda\right)\det N
$$

for $z \in (z_0, \infty)$.

This completes the proof. \Box

Since H is composed of $e(z, \lambda)$, $C(z, \lambda^2)$ and $S(z, \lambda^2)$, it is analytic in \mathbb{C}_+ and continuous up to the real axis.

4. Eigenvalues, Spectral Singularities And Resolvent Operator of T

From Lemma 1, a necessary and sufficient condition to investigate the eigenvalues and spectral singularities of the Sturm–Liouville operator T with impulsive condition \mathcal{B} is to investigate the zeros of the function H.

The set of eigenvalues and spectral singularities of the operator T are defined as

$$
\sigma_d(T) = {\mu = \lambda^2 : \text{Im}\lambda > 0 \text{ and } H(\lambda) = 0},
$$

$$
\sigma_{ss}(T) = {\mu = \lambda^2, \text{Im}\lambda = 0, \lambda \neq 0 \text{ and } H(\lambda) = 0},
$$

respectively.

Theorem 1. Under the condition $(\mathbf{4})$, the function H satisfies the following asymptotic equation

$$
H(\lambda) = \frac{\mu_1 \beta_2 \lambda^2}{\det N} \left(\frac{i}{4} + O\left(\frac{1}{\lambda}\right) \right), \quad \lambda \in \overline{\mathbb{C}}_+, \quad |\lambda| \to \infty,
$$

where $\mu_1 \beta_2 \neq 0$.

Proof. By means of $(\overline{5})$ - $(\overline{7})$, we easily find for $\lambda \in \mathbb{C}$

$$
S'\left(z_0, \lambda^2\right) = \cos \lambda z_0 + Q\left(z_0, z_0\right) \frac{\sin \lambda z_0}{\lambda} + \int\limits_0^{z_0} Q(z_0, t) \frac{\sin \lambda t}{\lambda} dt \tag{19}
$$

$$
C'\left(z_0, \lambda^2\right) = -\lambda \sin \lambda z_0 + R\left(z_0, z_0\right) \cos \lambda z_0 + \int_0^{z_0} R(z_0, t) \cos \lambda t dt \tag{20}
$$

and for $\lambda \in \overline{\mathbb{C}}_+$

$$
e'(z_0, \lambda) = i\lambda e^{i\lambda z_0} - K(z_0, z_0) e^{i\lambda z_0} + \int_{z_0}^{\infty} K_z(z_0, t) e^{i\lambda t} dt.
$$
 (21)

From $(5)-(7)$ $(5)-(7)$ $(5)-(7)$, we get

$$
S(z_0, \lambda^2) = \frac{e^{-i\lambda z_0}}{\lambda} \left(\frac{i}{2} + o(1)\right)
$$

\n
$$
C(z_0, \lambda^2) = e^{-i\lambda z_0} \left(\frac{1}{2} + o(1)\right)
$$

\n
$$
e(z_0, \lambda) = e^{i\lambda z_0} (1 + o(1)),
$$
\n(22)

where $\lambda\in\overline{\mathbb{C}}_+$ and $|\lambda|\to\infty.$

In a similar way, by using (19) - (21) , we obtain for $\lambda \in \overline{\mathbb{C}}_+$ and $|\lambda| \to \infty$

$$
S'(z_0, \lambda^2) = e^{-i\lambda z_0} \left(\frac{1}{2} + O\left(\frac{1}{\lambda}\right)\right)
$$

\n
$$
C'(z_0, \lambda^2) = \lambda e^{-i\lambda z_0} \left(-\frac{i}{2} + O\left(\frac{1}{\lambda}\right)\right)
$$

\n
$$
e'(z_0, \lambda) = \lambda e^{i\lambda z_0} \left(i + O\left(\frac{1}{\lambda}\right)\right).
$$
\n(23)

By means of (22) and (23) , it is obvious that $H(\lambda)$ satisfies the asymptotic equation given in Theorem $\boxed{1}$. This completes the proof. $\boxed{}$

Now, let us define another solution of $\left(1\right)$ - $\left(3\right)$

$$
G(z,\lambda) = \begin{cases} -(\zeta_0 + \zeta_1 \lambda) S(z,\lambda^2) + (\eta_0 + \eta_1 \lambda) C(z,\lambda^2); & z \in [0,z_0) \\ f^+(\lambda)e(z,\lambda) + h^+(\lambda)\breve{e}(z,\lambda); & z \in (z_0,\infty) \end{cases}
$$

for all $\lambda \in \overline{\mathbb{C}}_+\backslash \{0\}$. By the help of $\overline{\mathbb{3}}$, we obtain that

$$
\begin{bmatrix} f^{+}(\lambda) \\ h^{+}(\lambda) \end{bmatrix} = V \begin{bmatrix} -(\zeta_{0} + \zeta_{1} \lambda) \\ (\eta_{0} + \eta_{1} \lambda) \end{bmatrix},
$$
\n(24)

where

$$
V := \begin{bmatrix} V_{11}(\lambda) & V_{12}(\lambda) \\ V_{21}(\lambda) & V_{22}(\lambda) \end{bmatrix} = U^{-}BM \tag{25}
$$

with

$$
U = \begin{bmatrix} e(z_0, \lambda) & \breve{e}(z_0, \lambda) \\ e'(z_0, \lambda) & \breve{e}'(z_0, \lambda) \end{bmatrix}.
$$
 (26)

From (25) and (26) , the following equations can be found as

$$
V_{21}(\lambda) = \frac{i}{2\lambda} \left[-e'(z_0, \lambda) \left(\beta_1 S(z_0, \lambda^2) + \beta_2 S'(z_0, \lambda^2) \right) + e(z_0, \lambda) \left(\beta_3 S(z_0, \lambda^2) + \beta_4 S'(z_0, \lambda^2) \right) \right]
$$
(27)

$$
V_{22}(\lambda) = \frac{i}{2\lambda} \left[-e'(z_0, \lambda) \left(\beta_1 C(z_0, \lambda^2) + \beta_2 C'(z_0, \lambda^2) \right) + e(z_0, \lambda) \left(\beta_3 C(z_0, \lambda^2) + \beta_4 C'(z_0, \lambda^2) \right) \right].
$$
 (28)

By using (24) , the coefficients $f^+(\lambda)$ and $h^+(\lambda)$ must be as follows

$$
f^+(\lambda) = -(\zeta_0 + \zeta_1\lambda) V_{11}(\lambda) + (\eta_0 + \eta_1\lambda) V_{12}(\lambda)
$$

$$
h^+(\lambda) = -(\zeta_0 + \zeta_1\lambda) V_{21}(\lambda) + (\eta_0 + \eta_1\lambda) V_{22}(\lambda).
$$

By using (13) , (14) , (27) and (28) , it is clear that

$$
N_{21}(\lambda) = V_{21}(\lambda), \qquad N_{22}(\lambda) = V_{22}(\lambda).
$$

Therefore, using (18) , we rewrite $h^+(\lambda)$ as

$$
h^{+}(\lambda) = -H(\lambda) \det N.
$$
 (29)

In view of (29) , we obtain that

$$
W[E(z,\lambda),G(z,\lambda)] = \begin{cases} H(\lambda); & z \in [0,z_0) \\ 2i\lambda H(\lambda) \det N; & z \in (z_0,\infty) \end{cases}
$$

for $\lambda \in \overline{\mathbb{C}}_+ \backslash \{0\}.$

Theorem 2. Assume $\left(4\right)$. Then the resolvent operator of T is defined by

$$
\mathbb{R}_{\lambda}\phi = \int_{0}^{\infty} R(z, t; \lambda)\phi(t)dt,
$$

where

$$
R(z, t; \lambda) = \begin{cases} \frac{E(z, \lambda)G(t, \lambda)}{W[E(z, \lambda), G(z, \lambda)]}; & 0 \le t < z \\ \frac{G(z, \lambda)E(t, \lambda)}{W[E(z, \lambda), G(z, \lambda)]}; & z \le t < \infty \end{cases}
$$

is the Green function of \Box - \Box for $z \neq z_0$, $t \neq z_0$.

Proof. Let us consider the following equation

$$
-v'' + q(z)v - \lambda^2 v = \phi(z), \quad z \in [0, z_0) \cup (z_0, \infty).
$$
 (30)

By using the solutions $E(z, \lambda)$ and $G(z, \lambda)$, we write the solution of [\(30\)](#page-192-1)

$$
\phi(z,\lambda) = \theta_1(z)E(z,\lambda) + \theta_2(z)G(z,\lambda).
$$

Using the method of variation of parameters, we get the coefficients $\theta_1(z)$ and $\theta_2(z)$ as follows

$$
\theta_1(z) = k + \int_0^z \frac{\phi(t)G(t,\lambda)}{W[E(z,\lambda),G(z,\lambda)]} dt
$$

$$
\theta_2(z) = m + \int_z^\infty \frac{\phi(t)E(t,\lambda)}{W[E(z,\lambda),G(z,\lambda)]} dt,
$$

where k and m are real numbers. Let us write the coefficients $\theta_1(z)$ and $\theta_2(z)$ in solution $\phi(z,\lambda)$

$$
\phi(z,\lambda) = kE(z,\lambda) + \int_{0}^{z} \frac{\phi(t)G(t,\lambda)}{W[E(z,\lambda),G(z,\lambda)]} dt E(z,\lambda)
$$

$$
+ mG(z,\lambda) + \int_{z}^{\infty} \frac{\phi(t)E(t,\lambda)}{W[E(z,\lambda),G(z,\lambda)]} dtdG(z,\lambda) .
$$

Since the solution $\phi(z, \lambda)$ is in $L_2(0, \infty)$, m becomes zero. In accordance with the boundary condition \mathbb{Z} , we also find that k is equal to zero. The proof is completed. \Box

5. Unperturbated Impulsive Operators

In this section, we will investigate two unperturbated impulsive Sturm–Liouville operators.

Example 1. Now, we consider the Sturm-Liouville operator T_0 in $L^2[0,\infty)$ corresponding to the following impulsive problem

$$
-v'' = \lambda^2 v, \quad z \in [0, 1) \cup (1, \infty)
$$

$$
(\eta_0 + \eta_1 \lambda) v'(0) + (\zeta_0 + \zeta_1 \lambda) v(0) = 0
$$

$$
\begin{bmatrix} v(1^+) \\ v'(1^+) \end{bmatrix} = B \begin{bmatrix} v(1^-) \\ v'(1^-) \end{bmatrix}, \qquad B = \begin{bmatrix} \gamma_1 & 0 \\ 0 & \gamma_2 \end{bmatrix},
$$
(31)

where $\gamma_1, \gamma_2, \eta_j, \zeta_j, j = 0, 1$ are complex numbers such that $\eta_0 \zeta_1 - \eta_1 \zeta_0 \neq 0$ and $\gamma_1 \gamma_2 \neq 0$. Since $q = 0$, it is evident that

$$
e(z, \lambda) = e^{i\lambda z}
$$
, $C(z, \lambda^2) = \cos \lambda z$, $S(z, \lambda^2) = \frac{\sin \lambda z}{\lambda}$.

By using (18) , we write that

$$
H(\lambda) = \frac{ie^{i\lambda}}{2\lambda \det N} [(\eta_0 + \eta_1 \lambda)(i\gamma_1 \lambda \cos \lambda + \gamma_2 \lambda \sin \lambda) + (\zeta_0 + \zeta_1 \lambda)(\gamma_2 \cos \lambda - i\gamma_1 \sin \lambda)].
$$
\n(32)

To investigate the eigenvalues and spectral singularities of (31) , we examine the zeros of H. Let us choose $\zeta_1 = \eta_0 = 1$ and $\zeta_0 = \eta_1 = 0$ in [\(32\)](#page-193-1) for the simplicity. Therefore, we rewrite the equation (32)

$$
H(\lambda) = \frac{ie^{i\lambda}}{2 \det N} [i\gamma_1 \cos \lambda + \gamma_2 \sin \lambda - i\gamma_1 \sin \lambda + \gamma_2 \cos \lambda].
$$

We obtain that

$$
\lambda_k = -\frac{i}{2} \ln \left| \frac{1+D}{1-D} \right| + \frac{1}{2} Arg \left(\frac{1+D}{1-D} \right) + k\pi, \quad k \in \mathbb{Z},
$$

where
$$
D = \frac{\gamma_1 - i\gamma_2}{\gamma_2 - i\gamma_1}
$$
. There appear three cases:
\nCase1: Let $D = \frac{e^{i\theta} - 1}{e^{i\theta} + 1}$ such that $\theta \in \mathbb{R}$. Since $D = \frac{e^{i\theta} - 1}{e^{i\theta} + 1}$, it is easily seen that
\n $Arg\left(\frac{1+D}{1-D}\right) = \theta$ and $\left|\frac{1+D}{1-D}\right| = 1$. Then, we find that
\n $\lambda_k = \frac{\theta}{2} + k\pi$, $k \in \mathbb{Z}$.

In this case, $\lambda_k \in \mathbb{R} \setminus \{0\}$, $k \in \mathbb{Z}$ are the spectral singularities of (31) . However, there is no eigenvalues.

Case2: Let $\text{Im}D \neq 0$.

2a: Let D be purely imaginary. We obtain that

$$
\lambda_k = \frac{1}{2} \text{Arg}\left(\frac{1+D}{1-D}\right) + k\pi, \quad k \in \mathbb{Z}.
$$

In this case, similar with Case1, the ISBVP (31) has no eigenvalues. But it has spectral singularity.

2b: Assume $\text{Re}D < 0$. We get

$$
\lambda_k = -\frac{i}{2} \ln \left| \frac{1+D}{1-D} \right| + \frac{1}{2} Arg \left(\frac{1+D}{1-D} \right) + k\pi, \quad k \in \mathbb{Z}.
$$

 $Since 0 < \begin{bmatrix} \end{bmatrix}$ $1+D$ $1-D$ $< 1, \lambda_k \in \mathbb{C}_+, k \in \mathbb{Z}$ are the eigenvalues of $\boxed{31}$. However, the operator T_0 doesn't have any spectral singularity.

2c: For $0 < \text{Re}D$, the impulsive Sturm–Liouville boundary value problem $\boxed{31}$ has no eigenvalues and spectral singularity.

Case3: Let D be a real number. 3a: If $0 < D < 1$, then $1 <$ $1+D$ $1-D$ $\begin{array}{c} \begin{array}{c} \begin{array}{c} \end{array} \\ \begin{array}{c} \end{array} \end{array} \end{array}$. Similar to the Case2c, the eigenvalues and spectral singularity of (31) are not existing. 3b: For $1 < D < \infty$, we see that

$$
\lambda_k = -\frac{i}{2} \ln \left| \frac{1+D}{1-D} \right| + (2k+1)\frac{\pi}{2}, \quad k \in \mathbb{Z}.
$$

Since $\lambda_k \in \mathbb{C}_-$, there are no eigenvalues and spectral singularity. 3c: Assume $-1 < D < 0$. We obtain that

$$
\lambda_k = -\frac{i}{2} \ln \left(\frac{1+D}{1-D} \right) + k\pi, \quad k \in \mathbb{Z}.
$$

 $Since 0 < \vert$ $1+D$ $1-D$ $\begin{array}{c} \hline \rule{0pt}{2.2ex} \\ \rule{0pt}{2.2ex} \end{array}$ < 1 , there exists eigenvalues but the problem $\sqrt{31}$ has no spectral singularty.

3d: For $-\infty < D < 1$, we find that

 \lceil

$$
\lambda_k = -\frac{i}{2} \ln \left| \frac{1+D}{1-D} \right| + (2k+1)\frac{\pi}{2}, \quad k \in \mathbb{Z},
$$

where $0 < |$ $1+D$ $1-D$ $\begin{array}{c} \hline \rule{0pt}{2ex} \rule{$ $< 1.$ Hence, $\lambda_k \in \mathbb{C}_+, k \in \mathbb{Z}$ are the eigenvalues of T_0 . But this operator has no spectral singularity.

Example 2. We investigate the Sturm–Liouville operator T_1 in $L^2[0,\infty)$ created by the following ISBVP

$$
-v'' = \lambda^2 \rho(z)v, \quad z \in [0,1) \cup (1,\infty)
$$

\n
$$
(\eta_0 + \eta_1 \lambda) v'(0) + (\zeta_0 + \zeta_1 \lambda) v(0) = 0
$$

\n
$$
v(1^+)
$$

\n
$$
v'(1^+)
$$

\n
$$
= B \begin{bmatrix} v(1^-) \\ v'(1^-) \end{bmatrix}, \qquad B = \begin{bmatrix} \tau_1 & 0 \\ 0 & \tau_2 \end{bmatrix},
$$
\n(33)

where $\tau_1, \tau_2, \eta_j, \zeta_j, j = 0, 1$ are complex numbers, $\eta_0 \zeta_1 - \eta_1 \zeta_0 \neq 0, \tau_1 \tau_2 \neq 0$ and ρ is density function defined as

$$
\rho(z) = \begin{cases} \omega^2; & 0 \le z < 1 \\ 1; & 1 < z \end{cases}
$$

such that $\omega \in \mathbb{C} \setminus \{-1, 0, 1\}$. It is evident that for this example

$$
e(z, \lambda) = e^{i\lambda z}, \quad C(z, \lambda^2) = \cos(\lambda \omega z), \quad S(z, \lambda^2) = \frac{\sin(\lambda \omega z)}{\lambda \omega}.
$$

From (18) , we obtain that

$$
H(\lambda) = \frac{ie^{i\lambda}}{2\lambda \det N} [(\eta_0 + \eta_1 \lambda)(i\tau_1 \lambda \cos(\lambda \omega) + \tau_2 \lambda \omega \sin(\lambda \omega))
$$

$$
+ (\zeta_0 + \zeta_1 \lambda)(\tau_2 \cos(\lambda \omega) - i\tau_1 \frac{\sin(\lambda \omega)}{\omega})].
$$
(34)

For the simplicity on calculations, if we choose $\zeta_1 = \eta_0 = 1$ and $\zeta_0 = \eta_1 = 0$ in (34) , we get

$$
H(\lambda) = \frac{ie^{i\lambda}}{2 \det N} [i\tau_1 \cos(\lambda \omega) + \tau_2 \omega \sin(\lambda \omega) - i\tau_1 \frac{\sin(\lambda \omega)}{\omega} + \tau_2 \cos(\lambda \omega)].
$$

We easily find that

$$
\lambda_k = -\frac{i}{2\omega} \ln \left| \frac{1+P}{1-P} \right| + \frac{1}{2\omega} \left[\text{Arg} \left(\frac{1+P}{1-P} \right) + 2k\pi \right], \quad k \in \mathbb{Z},
$$

where $P = \frac{\tau_1 \omega - i \tau_2 \omega}{2}$ $\frac{m_1\omega}{\tau_2\omega^2 - i\tau_1}$. Let $\omega = m + in$. We can write the real and imaginary parts of λ_k as follows

$$
\text{Re}\lambda_k = \frac{1}{2 |\omega|^2} \left\{ m \left[\text{Arg}\left(\frac{1+P}{1-P}\right) + 2k\pi \right] - n \ln \left| \frac{1+P}{1-P} \right| \right\}
$$

and

$$
\text{Im}\lambda_k = -\frac{1}{2\left|\omega\right|^2} \left\{ m \ln \left| \frac{1+P}{1-P} \right| + n \left[\text{Arg}\left(\frac{1+P}{1-P}\right) + 2k\pi \right] \right\},\,
$$

respectively.

It is evident that if

$$
\left[m \ln \left| \frac{1+P}{1-P} \right| + n \left(\text{Arg} \left(\frac{1+P}{1-P} \right) + 2k\pi \right) \right] = 0
$$

then the operator T_1 has spectral singularities, and if

$$
\left[m \ln \left| \frac{1+P}{1-P} \right| + n \left(\text{Arg} \left(\frac{1+P}{1-P} \right) + 2k\pi \right) \right] < 0
$$

then the operator T_1 has eigenvalues.

Case1: If $P = \frac{e^{i\theta} - 1}{i\theta + 1}$ $\frac{e^{i\theta}-1}{e^{i\theta}+1}$, $\theta \in \mathbb{R}$, then $\text{Arg}\left(\frac{1+P}{1-P}\right)$ $1-F$ $\Big) = \theta$ and $\Big|$ $1 + F$ $1-F$ $\begin{array}{c} \begin{array}{c} \begin{array}{c} \end{array} \\ \begin{array}{c} \end{array} \end{array} \end{array}$ $= 1.$ We find that

$$
\lambda_k = \frac{\theta + 2k\pi}{2\omega}, \quad k \in \mathbb{Z}.
$$

1a: Assume $\omega \in \mathbb{R}$. $\lambda_k \in \mathbb{R} \setminus \{0\}$, $k \in \mathbb{Z}$ are spectral singularities of the operator T_1 but ISBVP (33) has no eigenvalues. 2a: Assume $\overline{\omega} \in \mathbb{C}$. We write

$$
\mathrm{Im}\lambda_{k} = -\frac{1}{2\left|\omega\right|^{2}}\left[n\left(\theta + 2k\pi\right)\right], \quad k \in \mathbb{Z}.
$$

If $n(\theta + 2k\pi) < 0$, then $\lambda_k \in \mathbb{C}_+$, $k \in \mathbb{Z}$ are eigenvalues of this problem [\(33\)](#page-195-1). Otherwise, the eigenvalues and spectral singularities of (33) are not existing. *Case2:* Let $\text{Im} P \neq 0$.

2a: Let P be purely imaginary. We write

$$
\lambda_k = \frac{1}{2\omega} \left[\text{Arg}\left(\frac{1+P}{1-P}\right) + 2k\pi \right], \quad k \in \mathbb{Z}.
$$

For $\omega \in \mathbb{R}, \lambda_k \in \mathbb{R} \setminus \{0\}, k \in \mathbb{Z}$ are spectral singularities of the operator T_1 . However, the problem (33) has no eigenvalues. If $\omega \in \mathbb{C}$, then we find that

$$
\text{Im}\lambda_k = -\frac{n}{2|\omega|^2} \left[\text{Arg}\left(\frac{1+P}{1-P}\right) + 2k\pi \right], \quad k \in \mathbb{Z}.
$$

It is easily seen that, for $n\left[\text{Arg}\left(\frac{1+P}{1-P}\right)\right]$ $1 - P$ $\Big\} + 2k\pi \Big] < 0$, the impulsive Sturm–Liouville boundary value problem $\boxed{33}$ has eigenvalues. Otherwise, the problem $\boxed{33}$ has no eigenvalues and spectral singularities.

2b: Assume $Re A < 0$. For $\omega \in \mathbb{R}$, we get

$$
\mathrm{Im}\lambda_k = -\frac{m}{2\left|\omega\right|^2} \left(\ln \left| \frac{1+P}{1-P} \right| \right), \quad k \in \mathbb{Z}.
$$

If $m > 0$, then the operator T_1 has eigenvalues. Otherwise, there are no eigenvalues and spectral singularities of (33) . For $\omega \in \mathbb{C}$, we obtain that

$$
\text{Im}\lambda_k = -\frac{1}{2\left|\omega\right|^2} \left\{ m \ln \left| \frac{1+P}{1-P} \right| + n \left[\text{Arg}\left(\frac{1+P}{1-P}\right) + 2k\pi \right] \right\}, \quad k \in \mathbb{Z}.
$$

If $m > 0$ and $n \left[\text{Arg} \left(\frac{1 + P}{1 - P} \right) \right]$ $1 - P$ $\Big\{+2k\pi\Big\}<0$, then $\lambda_k \in \mathbb{C}_+, k \in \mathbb{Z}$ are eigenvalues of [\(33\)](#page-195-1). However, if $m < 0$ and $n \left[\text{Arg} \left(\frac{1 + P}{1 - P} \right) \right]$ $1-F$ $+ 2k\pi$ > 0 then the operator T_1

has no eigenvalues and spectral singularities.

2c: Assume ReP > 0. Similar with case 2b, if $\omega \in \mathbb{R}$ and $m < 0$, then there exist eigenvalues of (33) . However, for $\omega \in \mathbb{R}$ and $m > 0$, there are no eigenvalues and spectral singularities of $ISBNP$ (33) .

Let $\omega \in \mathbb{C}$, it is clear that if $m < 0$ and $n \left[\text{Arg} \left(\frac{1 + P}{1 - P} \right) \right]$ $1-F$ $\Big\} + 2k\pi \Big\} < 0$, then the problem [\(33\)](#page-195-1) has eigenvalues. If $m > 0$ and $n \left[\text{Arg} \left(\frac{1 + P}{1 - P} \right) \right]$ $1 - P$ $+ 2k\pi$ > 0, then the eigenvalues and spectral singularities of (33) are not existing. Case3: Let P be a real number.

3a: For $0 < P < 1$, we find that

$$
\lambda_k = -\frac{i}{2\omega} \ln \left(\frac{1+P}{1-P} \right) + \frac{k\pi}{\omega}, \quad k \in \mathbb{Z}.
$$

Assume $\omega \in \mathbb{R}$. If $m < 0$, then the operator T_1 has eigenvalues. However, if $m > 0$, then the problem [\(33\)](#page-195-1) does not have any spectral singularity and eigenvalues. Assume $\omega \in \mathbb{C}$. If $m < 0$ and $n(2k\pi) < 0$, then $\lambda_k \in \mathbb{C}_+$, $k \in \mathbb{Z}$ are eigenvalues of ISBVP [\(33\)](#page-195-1) but if $m > 0$ and $n(2k\pi) > 0$, then the operator T_1 has no eigenvalues and spectral singularity.

3b: For $1 < P < \infty$, it is evident that

$$
\lambda_k = -\frac{i}{2\omega} \ln \left| \frac{1+P}{1-P} \right| + \frac{1}{2\omega} \left[(2k+1)\,\pi \right], \quad k \in \mathbb{Z}.
$$

Let $\omega \in \mathbb{R}$. Similar with Case3a, for $m < 0$, the problem [\(33\)](#page-195-1) has eigenvalues. Otherwise, the operator T_1 has no eigenvalues and spectral singularities.

Let $\omega \in \mathbb{C}$. If $m < 0$ and $n(2k + 1)$ $\pi < 0$, then there exists eigenvalues of [\(33\)](#page-195-1) but if $m > 0$ and $n (2k + 1) \pi > 0$, then there are no eigenvalues and spectral singularities. 3c: For $-1 < P < 0$, we obtain

$$
\lambda_k = -\frac{i}{2\omega} \ln \left(\frac{1+P}{1-P} \right) + \frac{k\pi}{2\omega}, \quad k \in \mathbb{Z}.
$$

Assume $\omega \in \mathbb{R}$. The operator T_1 has eigenvalues if and only if $m > 0$. Assume $\omega \in \mathbb{C}$. If $m > 0$ and $n (2k\pi) < 0$, then the problem [\(33\)](#page-195-1) has eigenvalues.

But if $m < 0$ and $n(2k\pi) > 0$, then ISBVP [\(33\)](#page-195-1) has no eigenvalues and spectral singularities.

3d: For $-\infty$ < P < 1, we get

$$
\lambda_k = -\frac{i}{2\omega} \ln \left| \frac{1+P}{1-P} \right| + \frac{1}{2\omega} \left[(2k+1)\,\pi \right], \quad k \in \mathbb{Z}.
$$

Let $\omega \in \mathbb{R}$. $\lambda_k \in \mathbb{C}_+$, $k \in \mathbb{Z}$ are eigenvalues of this example [\(33\)](#page-195-1) if and only if $m > 0$.

Let $\omega \in \mathbb{C}$. If $m > 0$ and $n(2k + 1)\pi < 0$, then there exists eigenvalues of [\(33\)](#page-195-1). If $m < 0$ and $n(2k+1)\pi > 0$, then the eigenvalues and spectral singularities of [\(33\)](#page-195-1) are not existing.

Case4: Let ω be purely imaginary. We easily find that

$$
\text{Im}\lambda_k = -\frac{n}{2\left|\omega\right|^2} \left[\text{Arg}\left(\frac{1+P}{1-P}\right) + 2k\pi \right], \quad k \in \mathbb{Z}.
$$

The operator T_1 has spectral singularities if and only is

$$
Arg\left(\frac{1+P}{1-P}\right) + 2k\pi = 0.
$$

The problem [\(33\)](#page-195-1) has eigenvalues if and only if

$$
n\left[\text{Arg}\left(\frac{1+P}{1-P}\right)+2k\pi\right]<0.
$$

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AN OVERVIEW TO ANALYTICITY OF DUAL FUNCTIONS

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Abstract. In this paper, the analyticity conditions of dual functions are clearly examined and the properties of the concept derivative are given in detail. Then, using the dual order relation, the dual analytic regions of dual analytic functions are constructed such that a collection of these regions forms a basis on $Dⁿ$. Finally, the equivalent of the inverse function theorem in dual space is given by a theorem and proved.

1. INTRODUCTION

In 1873, W. K. Clifford originally introduced the theory of algebra of dual numbers as a tool for his geometrical researches. Clifford showed that they constitute an algebra but not a field because only dual numbers with real part not zero have an inverse element $\boxed{1}$. An ordered pair of real numbers $\overline{x} = (x, x^*)$ is called a dual number, where x and x^* are termed by real part and dual part of the dual number, respectively. Dual numbers may be formally stated by $\bar{x} = x + \varepsilon x^*$, where $\varepsilon = (0, 1)$ is entitled by dual unit satisfying the condition that $\varepsilon^2 = 0$. The algebra of dual numbers is derived from this description. If $x = y$, $x^* = y^*$ for $\overline{x} = x + \varepsilon x^*$ and $\overline{y} = y + \varepsilon y^*, \overline{x}$ and \overline{y} are equal, and it is indicated as $\overline{x} = \overline{y}$. As for complex numbers, addition and product of two dual numbers are defined as follows, respectively:

$$
(x + \varepsilon x^*) + (y + \varepsilon y^*) = x + y + \varepsilon (x^* + y^*),
$$

$$
(x + \varepsilon x^*) \cdot (y + \varepsilon y^*) = xy + \varepsilon (xy^* + x^*y).
$$

The set of all dual numbers which is symbolized as D, i.e.,

$$
D = \{ \overline{x} = x + \varepsilon x^* \mid x, x^* \in \mathbb{R}, \ \varepsilon^2 = 0 \}
$$

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is a commutative ring over the real numbers field according to the operators $+$ and \cdot . The unit element of multiplication operation \cdot in D is the dual number $\overline{1}$ $=(1, 0) = 1 + \varepsilon$ 0. The dual number $\bar{x} = x + \varepsilon x^*$ that is divided by the dual number $\overline{y} = y + \varepsilon y^*$ providing $y \neq 0$ can be described as

$$
\frac{\overline{x}}{\overline{y}} = \frac{x + \varepsilon x^*}{y + \varepsilon y^*} = \frac{x}{y} + \varepsilon \left(\frac{x^* y - x y^*}{y^2} \right)
$$

(see $\overline{1}$ and $\overline{2}$). The dual number has a geometrical meaning which is discussed in detail in Yaglom $\boxed{3}$. It has contemporary applications within the curve design methods in computer aided geometric design and computer modeling of rigid bodies, linkages, robots, modelling human body dynamics, mechanism design, etc. [\[4\]](#page-223-4). The dual vectors were improved by A. P. Kotelnikov in the early part of the twentieth century $\overline{5}$. After W. K. Clifford, E. Study applied dual numbers and dual vectors to his study on kinematics and line geometry $[6]$. There exist several articles with regard to algebraic study of dual numbers (see $\boxed{1}$ and $\boxed{2}$). This nice notion was first performed by Kotelnikov to mechanics. Besides, the notion is often used in several fields of fundamental sciences such as astronomy, algebraic geometry, quantum mechanics and Riemannian geometry. For more details, we refer the reader to [\[3\]](#page-223-3)- [\[12\]](#page-224-0).

The set $D^n = \left\{ \frac{\overrightarrow{x}}{\overrightarrow{x}} = (\overrightarrow{x}_1, \overrightarrow{x}_2, ..., \overrightarrow{x}_n) \mid \overrightarrow{x}_i \in D, 1 \leq i \leq n \right\}$ is a module over the ring D according to the operators

$$
\overrightarrow{\overrightarrow{x}}+\overrightarrow{\overrightarrow{y}}=(\overline{x}_{1}+\overline{y}_{1},\overline{x}_{2}+\overline{y}_{2},...,\overline{x}_{n}+\overline{y}_{n})
$$

and

$$
\overrightarrow{\lambda}\overrightarrow{x}=(\overrightarrow{\lambda}\overline{x}_{1},\overrightarrow{\lambda}\overline{x}_{2},...,\overrightarrow{\lambda}\overline{x}_{n}).
$$

This module is called D−module or dual space. The elements of $Dⁿ$ are called dual vectors and a dual vector $\frac{3}{x}$ can be expressed as

$$
\overrightarrow{\overline{x}} = \overrightarrow{x} + \varepsilon \overrightarrow{x}^*,
$$

where \vec{x} and \vec{x}^* are real vectors in \mathbb{R}^n [\[1\]](#page-223-1).

The dual function
$$
\langle, \rangle_D : D^n \times D^n \to D
$$
,

$$
\left\langle \overrightarrow{x}, \overrightarrow{y} \right\rangle_{\mathcal{D}} = \left\langle \overrightarrow{x}, \overrightarrow{y} \right\rangle + \varepsilon \left(\left\langle \overrightarrow{x}, \overrightarrow{y}^* \right\rangle + \left\langle \overrightarrow{x}^*, \overrightarrow{y} \right\rangle \right)
$$

is called dual inner product function on D^n , where the notation \langle , \rangle is Euclidean inner product on \mathbb{R}^n .

Similar to dual inner product function, dual norm function $\|.\|_{\mathbb{D}} : \mathbb{D}^n \to \mathbb{D}$ is defined as follows:

$$
\left\|\overrightarrow{x}\right\|_{\mathcal{D}}=\left\{\begin{array}{cc}0&,\overrightarrow{x}=\overrightarrow{0}\\ \|\overrightarrow{x}\|+\varepsilon\frac{\left\langle\overrightarrow{x},\overrightarrow{x}^{*}\right\rangle}{\|\overrightarrow{x}\|}&,\overrightarrow{x}\neq\overrightarrow{0},\end{array}\right.
$$

where the notation $\|\cdot\|$ is Euclidean norm on \mathbb{R}^n .

Given the vectors $\overrightarrow{e}_i = (\overrightarrow{\delta}_{i1}, \overrightarrow{\delta}_{i2}, ..., \overrightarrow{\delta}_{in}),$ where

$$
\overline{\delta}_{ij} = \begin{cases} 1 + \varepsilon 0, & i = j \\ 0 + \varepsilon 0, & i \neq j \end{cases}, 1 \leq i, j \leq n,
$$

the set $\{\overrightarrow{e}_1, \overrightarrow{e}_2, ..., \overrightarrow{e}_n\}$ is standard basis of D^n . It turns out that every dual vector $\overrightarrow{x} \in D^n$ can be written in the form

$$
\overrightarrow{\overline{x}} = \overrightarrow{x_1} \overrightarrow{e}_1 + \overrightarrow{x_2} \overrightarrow{e}_2 + \dots + \overrightarrow{x_n} \overrightarrow{e}_n,
$$

where $\overrightarrow{e}_i = \overrightarrow{e}_i + \varepsilon \overrightarrow{0}$ for $1 \leq i \leq n$.

Consider that $\bar{x} = x + \varepsilon x^*$ and $\bar{y} = y + \varepsilon y^*$ are dual numbers. The relation \overline{x} \lt_{D} \overline{y} (resp. \overline{x} \leq_{D} \overline{y}) between these dual numbers is as follows (see [\[13\]](#page-224-1), [\[14\]](#page-224-2)):

1) Firstly, one compares the real parts of these dual numbers and must be $x < y$ (resp. $x < y$).

2) If the real parts of these dual numbers are the same, one compares their dual parts and must be $x^* < y^*$ (resp. $x^* \leq y^*$).

We can infer that there exist the following relations:

$$
\overline{x}
$$
 $\leq_D \overline{y} \Leftrightarrow x < y$ or $(x = y \text{ and } x^* < y^*)$

and

$$
\overline{x} \leq_{\text{D}} \overline{y} \Leftrightarrow x < y \text{ or } (x = y \text{ and } x^* \leq y^*).
$$

For the historical development of the term derivative, the expression "The derivative was first used, then discovered, and then studied and developed and finally defined." was used. The reason for using this expression is development process of the derivative starting with P. de Fermat in 1630s, continuing with I. Newton, J. L. Lagrange, G.W. Leibniz, A. L. Cauchy and reaching maturity in the 1870s with K. Weierstrass. The approaches to the derivative put forward by Leibniz and Newton were sufficient to find answers to the questions about the tangent of the curve and the velocity of the bodies. In fact, in the 19th century, this concept reached a consistent and solid foundation with the definition of derivative created by Cauchy using the term limit. It is well known that Cauchy put forward the first popularly acceptable account of the fundamental notions of the calculus. In order to prove the theorems related to the derivative, he used his own definitions. He described the derivative $\xi'(x)$ of a continuous function ξ as the limit when it exists, of the ratio $\frac{\xi(x+h)-\xi(x)}{h}$ as h went to zero. The instantaneous rate of change is entitled by the derivative. A comparison of the change in one quantity to the simultaneous change in a second quantity is expressed as a rate of change. Many of today's important problems in several fields such as engineering, biology, chemistry, physics, economics, involve finding the rate at which one quantity changes with respect to another, that is, they involve finding the derivative [\[15\]](#page-224-3).

Topology is a mathematical discipline which originated at the turn of the 20th century. On the other hand, some isolated results about topology can be traced

back several centuries. In mathematics, topology is interested in the properties of a geometric object which is preserved under continuous deformations including twisting, crumpling, stretching and bending. For many years, topology has been one of the most influential and exciting fields of research in modern mathematics. Topology is used for application fields such as physics, computer science, biology, robotics, fiber art, puzzles and games. Besides, topology has lots of applications in several branches of mathematics including differential equations, knot theory, dynamical systems, and Riemann surfaces in complex analysis. It also has some applications for describing the space-time structure of universe and analyzing many biological systems such as nanostructure and molecules, and in string theory in physics (see $[16]$ - $[29]$).

In this paper, using the order relation on dual numbers, we obtain the topology on D^n denoted by $\overline{\tau}_{\overline{d}}$. Then, how the analyticity conditions of a dual function which is often expressed in other studies are obtained is given clearly. Making use of this topology, dual analytic areas of dual analytic functions are determined. Besides, inner and external operations on the set constituted by dual analytic functions are given. With the help of these operations, some properties regarding dual analytic functions are expressed and proved. The relations between the elements of dual space and real space which will be used to define the basic concepts of differential geometry are examined. The terms dual tangent space, dual directional derivative, dual vector field and dual tangent map which are the basic tools of differential geometry are given in detail. The concepts of injective function, surjective function, inverse function and diffeomorphism in dual space is firstly expressed in this study. Some theorems related to these terms are obtained and proved. The foundation of term surface in dual space is constituted via these terms [\[13\]](#page-224-1).

2. On Dual Analytic Functions

Firstly, we shall study the concept of topology generating the basic structure of theory of curves and surfaces given by means of the expression of distance function in dual space. Previously, we talked about this basis [\[13\]](#page-224-1). After constructing a topology structure in dual space, we will determine the dual analytic regions of dual analytic functions by means of this topology.

Theorem 1. Given the sets

$$
\overline{B}(\overline{a}, \overline{r}) = \{\overline{x} = x + \varepsilon x^* \in D^n \mid ||x - a|| < r, x^* \in \mathbb{R}\}
$$

$$
\cup \{\overline{x} = x + \varepsilon x^* \in D^n \mid ||x - a|| = r \text{ and } \frac{\langle x - a, x^* - a^* \rangle}{||x - a||} < r^*\}
$$

$$
= U_1 \cup U_2
$$

$$
= U_1 \cup C_1 \cup ... \cup C_k, (k \in I = \{1, 2, ...\})
$$

and

$$
U_3 = \left\{ \overline{x} = x + \varepsilon x^* \in \mathcal{D}^n \mid x = a', \ m < x_1^* < n, \ x_{j+1}^* = c_j \in \mathbb{R}, \ m, n \in [-\infty, \infty] \right\},
$$

then a collection of all the sets $U_1, U_3, C_1, ..., C_k$ $(k \in I)$ forms a basis $\overline{\beta}$ on D^n , where $\overline{a} = a + \varepsilon a^* \in \mathbb{D}^n$, $r \in \mathbb{R}^+$, $r^* \in \mathbb{R}$ and $1 \leq j \leq n-1$.

Proof. It is enough to remark that two conditions given in definition of the term basis are satisfied.

i) It is easily seen that

$$
\bigcup_{\overline{A}\in\overline{\beta}}\overline{A}=\mathrm{D}^n
$$

.

ii) The set $\overline{A}_1 \cap \overline{A}_2$ is an arbitrary union of some sets belonging to class $\overline{\beta}$ for all $\overline{A}_1, \overline{A}_2 \in \overline{\beta}$ expect for $\overline{A}_1 \cap \overline{A}_2 = \emptyset$. Now, let us show that this expression is correct. Suppose that \overline{y} belongs to $\overline{A}_1 \cap \overline{A}_2$. Taking into account the sets $\overline{B}_1, \overline{B}_2, U'_3$ and U_3'' , the following situations hold, where

$$
\overline{B}_1(\overline{a}_1, \overline{r}_1) = \{ \overline{x} = x + \varepsilon x^* \in D^n \mid ||x - a_1|| < r_1, x^* \in \mathbb{R}^n \}
$$

$$
\cup \left\{ \overline{x} = x + \varepsilon x^* \in D^n \mid ||x - a_1|| = r_1 \text{ and } \frac{\langle x - a_1, x^* - a_1^* \rangle}{||x - a_1||} < r_1^* \right\}
$$

$$
= U_1 \cup C'_1 \cup ... \cup C'_l,
$$

$$
\overline{B}_2(\overline{a}_2, \overline{r}_2) = \{\overline{x} = x + \varepsilon x^* \in D^n \mid ||x - a_2|| < r_2, x^* \in \mathbb{R}^n\}
$$

$$
\cup \{\overline{x} = x + \varepsilon x^* \in D^n \mid ||x - a_2|| = r_2 \text{ and } \frac{\langle x - a_2, x^* - a_2^* \rangle}{||x - a_2||} < r_2^*\}
$$

$$
= U'_1 \cup C''_1 \cup ... \cup C''_l,
$$

 $U_3' = \{ \overline{x} = x + \varepsilon x^* \in D^n \mid x = b', \ m_1 < x_1^* < n_1, \ x_{j+1}^* = c_j' \in \mathbb{R}, \ m_1, n_1 \in [-\infty, \infty] \}$ and

$$
U_3'' = \{ \overline{x} = x + \varepsilon x^* \in \mathcal{D}^n \mid x = b'', \ m_2 < x_1^* < n_2, \ x_{j+1}^* = c_j'' \in \mathbb{R}, \ m_2, n_2 \in [-\infty, \infty] \}.
$$

1) Suppose that $\overline{y} \in U_3' \cap U_3''$. The following set can be written:

 $U_3' \cap U_3'' = \{ \overline{x} = x + \varepsilon x^* \in D^n \mid x = a, \ m < x_1^* < n, \ x_{j+1}^* = c_j \in \mathbb{R}, \ m, n \in [-\infty, \infty] \} \in \overline{\beta},$ where $y = b' = b'' = a$, $m < y_1^* < n$, $y_{j+1}^* = c_j' = c_j' = c_j \in \mathbb{R}$, $m = \max\{m_1, m_2\}$

and $n = \min \{n_1, n_2\}.$ 2) Assume that $\overline{y} \in U_1 \cap U_3''$. Hence, it is clear that $U_1 \cap U_3'' = U_3'' \in \overline{\beta}$.

3) Suppose that $\overline{y} \in C'_l \cap U''_3$ for any $l \in I$. In this case, the set $C'_l \cap U''_3$ can be written as

$$
U_3^j = \{ \overline{x} = x + \varepsilon x^* \in \mathcal{D}^n \mid x = a_j, \ m_j < x_1^* < n_j, \ x_{j+1}^* = c_j' \in \mathbb{R}, \ m_j, n_j \in [-\infty, \infty] \}.
$$

Therefore, $C'_l \cap U''_3 \in \overline{\beta}$.

4) Assume that $\overline{y} \in U_1 \cap U'_1$. The set $U_1 \cap U'_1$ can be written as an arbitrary union of the sets

$$
U = \{ \overline{x} = x + \varepsilon x^* \in \mathcal{D}^n \mid ||x - a|| < r, \ x^* \in \mathbb{R}^n \} \, .
$$

5) Suppose that $\overline{y} \in U_1 \cap C''_{l'}$ for any $l' \in I$. It is easy to check that $U_1 \cap C''_{l'} =$ $C''_{l'} \in \overline{\beta}.$

6) Assume that $\overline{y} \in C'_l \cap C''_{l'}$ for any $l, l' \in I$. The set $C'_l \cap C''_{l'}$ is expressed as $C_l \in \overline{\beta}$, for $l \in I$ or an arbitrary union of the sets U_3 belonging to class $\overline{\beta}$.

With these conventions, we have

$$
\overline{A}_1 \cap \overline{A}_2 = \bigcup_{\overline{A} \in \mathring{A} \subseteq \overline{\beta}} \overline{A}
$$

for all $\overline{A}_1, \overline{A}_2 \in \overline{\beta}$ expect for $\overline{A}_1 \cap \overline{A}_2 = \emptyset$, where the class \AA is a class of some sets belonging to the class $\overline{\beta}$. □

Definition 1. The class $\overline{\beta}$ given in the above mentioned theorem is called dual basis on D^n . The topology obtained from this basis is symbolized as $\overline{\tau}_{\overline{d}}$. Each element of this topology is termed by dual open set.

Theorem 2. Suppose that the class of the sets

$$
\overline{U} = \{ \overline{x} = x + \varepsilon x^* \in D^n \mid ||x - a|| < r, x^* \in \mathbb{R}^n \}
$$
\n
$$
= U \times \mathbb{R}^n
$$

belonging to the topology $\overline{\tau}_{\overline{d}}$ is symbolized as β_1 , where U is open set with respect to the standard topology of \mathbb{R}^n . Then the class $\overline{\beta}_1$ also constitutes a basis on D^n and the relationship between the topology $\bar{\tau}$ obtained from this basis and the topology $\bar{\tau}_{\bar{d}}$ $is \ \overline{\tau} \subseteq \overline{\tau}_{\overline{d}}.$

For example; let us study the topology $\overline{\tau}_{\overline{d}}$ on D. Assume that

$$
\overline{B}(\overline{a}, \overline{r}) = \left\{ \overline{x} = x + \varepsilon x^* \in D \mid |x - a| < r, x^* \in \mathbb{R} \right\}
$$
\n
$$
\cup \left\{ \overline{x} = x + \varepsilon x^* \in D \mid |x - a| = r \text{ and } \frac{(x - a)(x^* - a^*)}{|x - a|} < r^* \right\}
$$
\n
$$
= U_1 \cup U_2
$$
\n
$$
= U_1 \cup C_1 \cup C_2,
$$

where $\overline{a} = a + \varepsilon a^* \in D$, $r \in \mathbb{R}^+$, $r^* \in \mathbb{R}$ and

$$
U_2 = \{ \overline{x} = x + \varepsilon x^* \in D \mid x = a + r, \ x^* < a^* + r^* \}
$$

$$
\cup \{ \overline{x} = x + \varepsilon x^* \in D \mid x = a - r, \ x^* > a^* - r^* \}
$$

$$
= C_1 \cup C_2.
$$

Taking into consideration the set

$$
U_3=\left\{\overline{x}=x+\varepsilon x^*\in\mathcal{D}\mid x=a',\ m
$$

the collection of the sets U_1, C_1, C_2 and U_3 forms a basis on D. The topology obtained from this basis is symbolized as $\overline{\tau}_{\overline{d}}$. Besides, the collection of the sets $U_1 \cup U_2$ and U_3 is also a basis on D and the topology generated by this basis is also $\overline{\tau}_{\overline{d}}$.

Observe that

 \overline{a}

$$
B = \left\{ \tilde{x} = (x, x^*) \in \mathbb{R}^2 \mid a < x < b, \ c < x^* < d, \ a, b, c, d \in \mathbb{R} \right\}.
$$

The collection of all the sets B forms a basis on \mathbb{R}^2 . If the topology generated by this basis is symbolized as τ_1 , the relationship between $\overline{\tau}_{\overline{d}}$ and τ_1 is $\tau_1 \subseteq \overline{\tau}_{\overline{d}}$. On the other hand, if the topology derived from the collection of only the sets U_1 is symbolized as $\bar{\tau}$, then there exists the following relationship:

$$
\overline{\tau} \subseteq \tau_1 \subseteq \overline{\tau}_{\overline{d}}.
$$

Definition 2. Let $\overline{x} = x + \varepsilon x^*$ be a dual variable. The function $\overline{\xi} : D \to D$ of the dual variable $\overline{x} = x + \varepsilon x^*$ is defined as follows:

$$
\overline{\xi}(\overline{x}) = \xi(x, x^*) + \varepsilon \xi^0(x, x^*),
$$

where ξ and ξ^0 are real functions of the two real variables x and x^* .

In the following theorem, by eliminating the deficiencies in other studies, we shall discuss analyticity conditions of dual functions.

Theorem 3. The dual function $\overline{\xi}$: $\overline{U} \subseteq D \to D$, $\overline{\xi}$ ($\overline{x} = x + \varepsilon x^*$) = $\xi(x, x^*)$ + $\varepsilon \xi^0(x, x^*)$ is said to be analytic at the point $\overline{x} \in \overline{U}$ if and only if the functions ξ and ξ^0 have continuous partial derivatives ξ_x and ξ_x^0 and there exist the equalities $\xi_{x^*} = 0$ and $\xi_{x^*}^0 = \xi_x$, where $\xi_x = \frac{\partial \xi}{\partial x}$.

Proof. Firstly, let the dual function $\overline{\xi}$ be analytic at the point $\overline{x} \in \overline{U}$. Thus, this assumption permits us to write the following relation:

$$
\frac{d\overline{\xi}}{d\overline{x}} = \lim_{\overline{h}\to 0} \frac{\overline{\xi}(\overline{x} + \overline{h}) - \overline{\xi}(\overline{x})}{\overline{h}}.
$$
\n(1)

Observe that $\bar{x} = x + \varepsilon x^*$ and $\bar{h} = h + \varepsilon h^*$. By definition of dual variable functions and $\varepsilon^2 = 0$, the following equality holds:

$$
\frac{d\overline{\xi}}{d\overline{x}} = \lim_{\overline{h}\to\overline{0}} \frac{\overline{\xi}(\overline{x}+\overline{h})-\overline{\xi}(\overline{x})}{\overline{h}}
$$
\n
$$
= \lim_{(h,h^*)\to(0,0)} \frac{\xi(x+h,x^*+h^*)+\varepsilon\xi^0(x+h,x^*+h^*)-\xi(x,x^*)-\varepsilon\xi^0(x,x^*)}{h+\varepsilon h^*}
$$
\n
$$
= \lim_{(h,h^*)\to(0,0)} \frac{\xi(x+h,x^*+h^*)-\xi(x,x^*)}{h}
$$
\n
$$
+ \lim_{(h,h^*)\to(0,0)} \varepsilon \left(\frac{\frac{\xi^0(x+h,x^*+h^*)-\xi^0(x,x^*)}{h}}{\frac{h}{h^2}(\xi(x+h,x^*+h^*)-\xi(x,x^*))} \right)
$$
\n
$$
= \frac{\partial\xi}{\partial x} + \varepsilon \frac{\partial\xi^0}{\partial x}.
$$

In view of equation (II) , it is seen that the limit for $(h, h^*) \to (0, 0)$ of real part of the expression $\frac{\overline{\xi}(\overline{x}+\overline{h})-\overline{\xi}(\overline{x})}{\overline{x}}$ $\frac{\partial h}{\partial \overline{\partial}} - \xi(\overline{x})$ is $\frac{\partial \xi}{\partial x}$. Then, it is easy to check that

$$
\frac{\xi(x+h,x^*+h^*)-\xi(x,x^*)}{h} = \frac{\xi(x+h,x^*+h^*)-\xi(x,x^*+h^*)}{h} + \frac{\xi(x,x^*+h^*)-\xi(x,x^*)}{h}.
$$
 (2)

From the hypothesis and the equality (2) , we have

$$
\lim_{(h,h^*) \to (0,0)} \frac{\xi(x, x^* + h^*) - \xi(x, x^*)}{h} = 0.
$$

If this limit exists and equals to zero, it is obvious from discussion that

$$
\xi(x, x^* + h^*) - \xi(x, x^*) = 0
$$

such that $\xi(x, x^*) = \xi(x)$. Thus, the function ξ depends only on the variable x, i.e., $\frac{\partial \xi}{\partial x}$ $\frac{\partial \zeta}{\partial x^*} = 0$. It is well known from equation (II) that the limit for $(h, h^*) \to (0, 0)$ of dual part of the expression $\frac{\overline{\xi}(\overline{x}+\overline{h})-\overline{\xi}(\overline{x})}{\overline{x}}$ $\frac{\overline{h}}{\overline{h}}$ = $\frac{\overline{\xi}(\overline{x})}{\overline{h}}$ is $\frac{\partial \xi^0}{\partial x}$ $rac{\partial \mathbf{x}}{\partial x}$. By some calculations, the following equality holds:

$$
\frac{\xi^{0}(x+h,x^{*}+h^{*})-\xi^{0}(x,x^{*})}{h} - \frac{\xi(x+h,x^{*}+h^{*})-\xi(x,x^{*})}{h} \frac{h^{*}}{h}
$$
\n
$$
= \frac{\xi^{0}(x+h,x^{*}+h^{*})-\xi^{0}(x,x^{*}+h^{*})}{h} + \frac{\xi^{0}(x,x^{*}+h^{*})-\xi^{0}(x,x^{*})}{h} - \frac{\xi(x+h)-\xi(x)}{h} \frac{h^{*}}{h}.
$$
\n(3)

From the hypothesis and the equality (3) , we get

$$
\lim_{h^* \to 0} \left(\lim_{h \to 0} \frac{h\left(\xi^0(x, x^* + h^*) - \xi^0(x, x^*)\right) - h^*\left(\xi(x + h) - \xi(x)\right)}{h^2} \right) = 0. \tag{4}
$$

Since the statement

$$
\lim_{h \to 0} \frac{h\left(\xi^0\left(x, x^* + h^*\right) - \xi^0\left(x, x^*\right)\right) - h^*\left(\xi\left(x + h\right) - \xi\left(x\right)\right)}{h^2}
$$

has the indefiniteness $(\frac{0}{0})$, we write the following equality:

$$
\lim_{h^* \to 0} \left(\lim_{h \to 0} \frac{\left(\xi^0(x, x^* + h^*) - \xi^0(x, x^*)\right) - h^* \xi_x(x + h)}{2h} \right) = 0. \tag{5}
$$

From (5) , we obtain

$$
\xi^{0}(x, x^{*} + h^{*}) - \xi^{0}(x, x^{*}) = h^{*}\xi_{x}(x).
$$

Therefore, it is possible to express that

$$
\frac{\xi^{0}(x, x^{*} + h^{*}) - \xi^{0}(x, x^{*})}{h^{*}} = \xi_{x}(x),
$$

where $h^* \neq 0$. The limit of both sides of this identity for $h^* \to 0$ is $\xi_{x^*}^0 = \xi_x$.

Conversely, suppose that the functions ξ and ξ^0 have continuous partial derivatives ξ_x and ξ_x^0 and there are the equalities $\xi_{x^*} = 0$ and $\xi_{x^*}^0 = \xi_x$. The expression of dual function ξ is simplified to the following form

$$
\overline{\xi}(\overline{x}) = \xi(x) + \varepsilon \left(x^* \xi'(x) + \widetilde{\xi}(x) \right), \tag{6}
$$

where $\xi \in C^2$, $\xi \in C^1$. Given a point $\overline{x} \in \overline{U}$, we must show that the expression $\lim_{\overline{h}\to 0}$ $\overline{\xi}\left(\overline{x}+\overline{h}\right)-\overline{\xi}\left(\overline{x}\right)$ $\frac{\sqrt{h}}{\hbar}$ exists. From the equality [\(6\)](#page-208-0), the derivative of the dual function $\overline{\xi}$ with respect to dual variable \overline{x} can be expressed as follows:

$$
I = \lim_{\overline{h}\to\overline{0}} \frac{\overline{\xi}(\overline{x}+\overline{h}) - \overline{\xi}(\overline{x})}{\overline{h}} = \lim_{(h,h^*)\to(0,0)} \left[\begin{array}{c} \frac{\xi(x+h)-\xi(x)}{h} \\ +\varepsilon \left(\begin{array}{c} x^* \left(\frac{\xi'(x+h)-\xi(x)}{h} \right) + \frac{\xi(x+h)-\widetilde{\xi}(x)}{h} \\ +\frac{h^*}{h} \xi'(x+h) - \frac{\xi(x+h)-\xi(x)}{h} \frac{h^*}{h} \end{array} \right) \end{array} \right].
$$

From the hypothesis, we have

$$
I_1 = \lim_{(h,h^*) \to (0,0)} \frac{\xi(x+h) - \xi(x)}{h} = \xi'(x),
$$

\n
$$
I_2 = \lim_{(h,h^*) \to (0,0)} x^* \left(\frac{\xi'(x+h) - \xi'(x)}{h} \right) = x^* \xi''(x),
$$

\n
$$
I_3 = \lim_{(h,h^*) \to (0,0)} \frac{\tilde{\xi}(x+h) - \tilde{\xi}(x)}{h} = \tilde{\xi}'(x),
$$

\n
$$
I_4 = \lim_{(h,h^*) \to (0,0)} \frac{h^*}{h} \xi'(x+h) - \frac{\xi(x+h) - \xi(x)}{h} \frac{h^*}{h} = 0
$$

such that

$$
I = I_1 + \varepsilon (I_2 + I_3 + I_4) = \xi'(x) + \varepsilon \left(x^* \xi''(x) + \tilde{\xi}'(x) \right)
$$

Thus, this obviously completes the proof of the theorem. \Box

.

We are now ready to state the following corollaries.

Corollary 1. Theorem 3 implies that the derivative of dual function $\overline{\xi} : \overline{U} \subseteq D \to D$ with respect to dual variable \bar{x} is

$$
\frac{d\overline{\xi}}{d\overline{x}} = \lim_{\Delta \overline{x} \to 0} \frac{\overline{\xi}(\overline{x} + \Delta \overline{x}) - \overline{\xi}(\overline{x})}{\Delta \overline{x}}.
$$

This limit is independent of the ratio $\frac{\Delta x^*}{\Delta x}$ $\frac{d^{2}x}{\Delta x}$ [\[30\]](#page-224-6). Corollary 2. Taking into account Theorem 3, the analyticity conditions of dual $function \overline{\xi}: \overline{U} \subseteq D \to D, \overline{\xi}(\overline{x}) = \xi(x, x^*) + \varepsilon \xi^0(x, x^*)$ are $\frac{\partial \xi}{\partial \overline{x}}$ function $\overline{\xi} : \overline{U} \subseteq D \to D$, $\overline{\xi}(\overline{x}) = \xi(x, x^*) + \varepsilon \xi^0(x, x^*)$ are $\frac{\partial \xi}{\partial x^*} = 0$ and $\frac{\partial \xi^0}{\partial x^*} = \frac{\partial \xi}{\partial x}$.
Thus, the general representation of dual analytic functions is

$$
\overline{\xi}(\overline{x}) = \xi(x, x^*) + \varepsilon \xi^0(x, x^*) = \xi(x) + \varepsilon \left(x^* \xi'(x) + \widetilde{\xi}(x) \right),
$$

where $\xi, \widetilde{\xi}: U \subseteq \mathbb{R} \to \mathbb{R}$ and $\xi \in C^2, \widetilde{\xi} \in C^1$. In the proof of Theorem 3, it is clearly seen that the derivative of this analytic function $\overline{\xi}$ with respect to dual variable \overline{x} is

$$
\frac{d\bar{\xi}}{d\bar{x}} = \frac{\partial \xi}{\partial x} + \varepsilon \frac{\partial \xi^{0}}{\partial x} = \xi'(x) + \varepsilon \left(x^* \xi''(x) + \tilde{\xi}'(x) \right)
$$

[\[30\]](#page-224-6).

Now, based on Theorem 3, let us determine the analyticity conditions of dual function $\overline{\xi} : \overline{U} \subseteq D^n \to D$,

$$
\overline{\xi}(\overline{x}) = \xi(x_1, ..., x_n, x_1^*, ..., x_n^*) + \varepsilon \xi^0(x_1, ..., x_n, x_1^*, ..., x_n^*) = \xi + \varepsilon \xi^0.
$$

The partial derivatives of dual function $\bar{\xi}$ at any dual point $\bar{a} \in \bar{U} \subseteq D^n$ (if there exists) are

$$
\frac{\partial \overline{\xi}}{\partial \overline{x}_i}(\overline{a}) = \lim_{\Delta \overline{x}_i \to \overline{0}} \frac{\overline{\xi}(\overline{a}_1, ..., \overline{a}_i + \Delta \overline{x}_i, ..., \overline{a}_n) - \overline{\xi}(\overline{a}_1, ..., \overline{a}_n)}{\Delta \overline{x}_i}, 1 \le i \le n.
$$

The above formula is simplified to the following form:

$$
\frac{\partial \overline{\xi}}{\partial \overline{x}_i}(\overline{a}) = \frac{d}{d\overline{x}_i} \overline{\xi}(\overline{a}_1, ..., \overline{x}_i, ..., \overline{a}_n) \mid_{\overline{x}_i = \overline{a}_i} = \lim_{\overline{x}_i \to \overline{a}_i} \frac{\overline{\mu}(\overline{x}_i) - \overline{\mu}(\overline{a}_i)}{\overline{x}_i - \overline{a}_i},
$$

where $\bar{\mu}(\bar{x}_i) = \xi(\bar{a}_1, ..., \bar{x}_i, ..., \bar{a}_n)$. When Theorem 3 is taken into consideration, one can check that if this limit exists, for $1 \leq i \leq n$, then the functions ξ and ξ^0 have continuous partial derivatives ξ_{x_i} and $\xi_{x_i}^0$ at any dual point $\overline{a} \in \overline{U}$ and these relations $\frac{\partial \xi}{\partial x_i^*}$ $= 0$ and $\frac{\partial \xi^0}{\partial x_i^*}$ $=\frac{\partial \xi}{\partial x}$ $\frac{\partial \mathbf{S}}{\partial x_i}$ are satisfied. From Theorem 3, it is easy to see that the reverse exists. This result follows by proceeding as in the proof of the first assertion. Thus, these conventions permit us to write the following relation:

$$
\frac{\partial \overline{\xi}}{\partial \overline{x}_i}(\overline{a}) = \frac{\partial \xi}{\partial x_i} (a_1, ..., a_n, a_1^*, ..., a_n^*) + \varepsilon \frac{\partial \xi^0}{\partial x_i} (a_1, ..., a_n, a_1^*, ..., a_n^*).
$$

Besides, the expression $\lim_{\Delta \bar{x}_i \to 0}$ ∆ξ $\frac{\Delta \bar{\xi}}{\Delta \bar{x}_i}$ is independent of the ratio $\frac{\Delta x_i^*}{\Delta x_i}$ $\frac{\Delta x_i}{\Delta x_i}$. Note that the analyticity conditions of dual function $\overline{\xi} : \overline{U} \subseteq D^n \to D$ are $\frac{\partial \xi}{\partial x_i^*}$ $= 0$ and $\partial \xi^0$ $\overline{\partial} x^*_i$ $=\frac{\partial \xi}{\partial x}$ $\frac{\partial \mathcal{S}}{\partial x_i}$ (1 ≤ *i* ≤ *n*). In view of these equalities, we can write the following

expressions:

$$
\xi(x_1,...,x_n,x_1^*,...,x_n^*)=\xi(x_1,...,x_n)
$$

and

$$
\xi^{0}\left(x_{1},...,x_{n},x_{1}^{*},...,x_{n}^{*}\right)=\sum_{i=1}^{n}x_{i}^{*}\frac{\partial\xi}{\partial x_{i}}+\widetilde{\xi}\left(x_{1},...,x_{n}\right),
$$

where $\xi \in C^2$, $\tilde{\xi} \in C^1$. By definition of the analyticity conditions of dual function $\bar{\xi}$: $\bar{U} \subseteq D^n \to D$, the general representation of these dual analytic functions is

$$
\overline{\xi}(\overline{x}) = \xi(x_1, ..., x_n) + \varepsilon \left(\sum_{i=1}^n x_i^* \frac{\partial \xi}{\partial x_i} + \widetilde{\xi}(x_1, ..., x_n) \right). \tag{7}
$$

The partial derivatives of this function with respect to dual variables \bar{x}_i are

$$
\frac{\partial \overline{\xi}}{\partial \overline{x}_j} = \frac{\partial \xi}{\partial x_j} + \varepsilon \left(\sum_{i=1}^n x_i^* \frac{\partial^2 \xi}{\partial x_j \partial x_i} + \frac{\partial \widetilde{\xi}}{\partial x_j} \right)
$$

 $(1 \leq j \leq n)$. Throughout this paper, the functions ξ and $\tilde{\xi}$ will be considered as belonging to C^{∞} –class. Note that the sets of the topology $\overline{\tau}$ mentioned in Theorem 2 is dual analytic regions of dual analytic functions. The set of dual analytic functions is symbolized as $C(\overline{U} \subseteq D^n, D)$. Therefore, the following expression holds:

$$
C\left(\overline{U} \subseteq D^{n}, D\right) = \left\{\overline{\xi} \mid \overline{\xi} : \overline{U} \subseteq D^{n} \to D, \ \overline{\xi}(\overline{x}) = \xi(x) + \varepsilon \left(\sum_{i=1}^{n} x_{i}^{*} \frac{\partial \xi}{\partial x_{i}} + \widetilde{\xi}(x)\right)\right\}.
$$

Given the dual functions $\bar{\xi}: \bar{U} \subseteq D^n \to D^m, \bar{\xi} = (\bar{\xi}_1, ..., \bar{\xi}_m)$, we conclude that if the dual functions $\overline{\xi}_j : \overline{U} \subseteq D^n \to D$, $(1 \le j \le m)$ are dual analytic, then the dual function $\overline{\xi}$ is dual analytic. When the above information is taken into consideration, the following functions can be defined:

 $(i) +_C : C \left(\overline{U} \subseteq D^n, D \right) \times C \left(\overline{U} \subseteq D^n, D \right) \to C \left(\overline{U} \subseteq D^n, D \right), \text{for } \overline{\xi}, \overline{\mu} \in C \left(\overline{U} \subseteq D^n, D \right)$ and $\overline{x} \in \overline{U} \subseteq D^n$, we have

$$
\left(\overline{\xi} + C\,\overline{\mu}\right)(\overline{x}) = \overline{\xi}\left(\overline{x}\right) + \overline{\mu}\left(\overline{x}\right) = \xi\left(x\right) + \mu\left(x\right) + \varepsilon\left(\sum_{i=1}^{n} x_i^* \frac{\partial\left(\xi + \mu\right)}{\partial x_i} + \widetilde{\xi}\left(x\right) + \widetilde{\mu}\left(x\right)\right).
$$

 $ii)$ \cdot_C : $D \times C$ $(\overline{U} \subseteq D^n, D) \rightarrow C$ $(\overline{U} \subseteq D^n, D)$, for $\overline{\xi} \in C$ $(\overline{U} \subseteq D^n, D)$, $\overline{\lambda} =$ $\lambda + \varepsilon \lambda^* \in \mathcal{D}$ and $\overline{x} \in \overline{U} \subseteq \mathcal{D}^n$, we have

$$
\left(\overline{\lambda} \cdot_{C} \overline{\xi}\right)(\overline{x}) = \overline{\lambda} \cdot \overline{\xi}(\overline{x}) = \lambda \xi(x) + \varepsilon \left(\sum_{i=1}^{n} x_{i}^{*} \frac{\partial(\lambda \xi)}{\partial x_{i}} + \lambda \widetilde{\xi}(x) + \lambda^{*} \xi(x)\right).
$$

 $iii) \cdot_{1_C} : C \left(\overline{U} \subseteq D^n, D \right) \times C \left(\overline{U} \subseteq D^n, D \right) \to C \left(\overline{U} \subseteq D^n, D \right), \text{for } \overline{\xi}, \overline{\mu} \in C \left(\overline{U} \subseteq D^n, D \right)$ and $\overline{x} \in \overline{U} \subseteq D^n$, we have

$$
\left(\overline{\xi} \cdot_{1_C} \overline{\mu}\right)(\overline{x}) = \overline{\xi}(\overline{x}) \cdot \overline{\mu}(\overline{x}) = \xi(x) \mu(x) + \varepsilon \left(\sum_{i=1}^n x_i^* \frac{\partial(\xi \mu)}{\partial x_i} + \xi(x) \widetilde{\mu}(x) + \widetilde{\xi}(x) \mu(x)\right)
$$
\n[31].

We are interested now to some properties regarding dual analytic functions.

Proposition 1. Consider $\overline{\mu} : \overline{I} \subseteq D \to D^n$ and $\overline{\xi} : \overline{U} \subseteq D^n \to D$ are dual analytic functions, where the functions $\overline{\mu}$ and $\overline{\xi}$ are as below:

$$
\overline{\mu}\left(\overline{t}\right) = \mu\left(t\right) + \varepsilon\left(t^*\mu'\left(t\right) + \widetilde{\mu}\left(t\right)\right)
$$

and

$$
\overline{\xi}(\overline{x}) = \xi(x) + \varepsilon \left(\sum_{i=1}^{n} x_i^* \frac{\partial \xi}{\partial x_i} + \widetilde{\xi}(x) \right)
$$

such that the functions $\xi, \tilde{\xi}, \mu$ and $\tilde{\mu}$ belong to $C^{\infty}-$ class. If the functions $\overline{\xi}$ and $\overline{\mu}$ are dual analytic at the dual points $\bar{\mu}(\bar{t})$ and \bar{t} , respectively, then the composition of $\overline{\mu}$ and $\overline{\xi}$, *i.e.*, $\overline{\xi} \circ \overline{\mu}$ is dual analytic function. The derivative of this dual analytic function with respect to dual variable \bar{t} is

$$
\frac{d}{d\overline{t}}\left(\overline{\xi}\circ\overline{\mu}\right)\left(\overline{t}\right) = \left(\xi\circ\mu\right)'(t) + \varepsilon \left(t^*\left(\xi\circ\mu\right)''(t) + \left\langle\mu(t), \sum_{i=1}^n \left(\frac{\partial\xi}{\partial x_i}\circ\mu\right)(t) \overrightarrow{e}_i \right\rangle' + \left(\widetilde{\xi}\circ\mu\right)'(t) \right),
$$

where $(\xi \circ \mu)'(t) = \frac{d}{dt} (\xi \circ \mu)(t)$.

Theorem 4. Let $\overline{\xi}$: $\overline{U} \subseteq D^n \rightarrow D$ be dual analytic function. Then the following identity holds

$$
\frac{\partial^2 \overline{\xi}}{\partial \overline{x}_k \partial \overline{x}_j} = \frac{\partial^2 \overline{\xi}}{\partial \overline{x}_j \partial \overline{x}_k} \quad (1 \le j, k \le n),
$$

for any dual point of $\overline{U} \subseteq D^n$.

Proof. Let $\overline{\xi} : \overline{U} \subseteq D^n \to D$ be dual analytic function. From the equality (\mathbb{Z}) , we can write

$$
\overline{\xi}(\overline{x}) = \xi(x) + \varepsilon \left(\sum_{i=1}^{n} x_i^* \frac{\partial \xi}{\partial x_i} + \widetilde{\xi}(x) \right),
$$

where $\xi, \tilde{\xi} \in C^{\infty}$. The partial derivatives of dual function $\overline{\xi}$ with respect to dual variable \overline{x}_j are

$$
\frac{\partial \overline{\xi}}{\partial \overline{x}_j} = \frac{\partial \xi}{\partial x_j} + \varepsilon \left(\sum_{i=1}^n x_i^* \frac{\partial^2 \xi}{\partial x_j \partial x_i} + \frac{\partial \widetilde{\xi}}{\partial x_j} \right) \n= \frac{\partial \xi}{\partial x_j} + \varepsilon \left(\sum_{i=1}^n x_i^* \frac{\partial^2 \xi}{\partial x_i \partial x_j} + \frac{\partial \widetilde{\xi}}{\partial x_j} \right).
$$

The above formula are simplified to the following form

$$
\frac{\partial \overline{\xi}}{\partial \overline{x}_j} = \mu(x) + \varepsilon \left(\sum_{i=1}^n x_i^* \frac{\partial \mu}{\partial x_i} + \widetilde{\mu}(x) \right),
$$

where $\frac{\partial \xi}{\partial x_j} = \mu(x)$ and $\frac{\partial \xi}{\partial x_j} = \tilde{\mu}(x)$, i.e., $\mu, \tilde{\mu} \in C^{\infty}$. Thus, we deduce that ∂ξ $\frac{\partial \xi}{\partial \overline{x}_j} \in C\left(\overline{U} \subseteq D^n, D\right)$. In analogous to the derivative $\frac{\partial \xi}{\partial \overline{x}_j}$, the partial derivatives of dual function $\frac{\partial \xi}{\partial \overline{x}_j}$ with respect to dual variable \overline{x}_k are

$$
\frac{\partial^2 \overline{\xi}}{\partial \overline{x}_k \partial \overline{x}_j} = \frac{\partial \mu}{\partial x_k} + \varepsilon \left(\sum_{i=1}^n x_i^* \frac{\partial^2 \mu}{\partial x_k \partial x_i} + \frac{\partial \widetilde{\mu}}{\partial x_k} \right)
$$

$$
= \frac{\partial \mu}{\partial x_k} + \varepsilon \left(\sum_{i=1}^n x_i^* \frac{\partial^2 \mu}{\partial x_i \partial x_k} + \frac{\partial \widetilde{\mu}}{\partial x_k} \right),
$$

where $\frac{\partial \mu}{\partial x_k} = \frac{\partial^2 \xi}{\partial x_k \partial x}$ $\frac{\partial^2 \xi}{\partial x_k \partial x_j} = \frac{\partial^2 \xi}{\partial x_j \partial x_j}$ $\frac{\partial^2 \xi}{\partial x_j \partial x_k}$ and $\frac{\partial \widetilde{\mu}}{\partial x_k} = \frac{\partial^2 \widetilde{\xi}}{\partial x_k \partial x_k}$ $\frac{\partial^2 \widetilde{\xi}}{\partial x_k \partial x_j} = \frac{\partial^2 \widetilde{\xi}}{\partial x_j \partial x_j}$ $\frac{\partial^2 \mathbf{S}}{\partial x_j \partial x_k}$. Therefore, this yields

$$
\frac{\partial^2 \overline{\xi}}{\partial \overline{x}_k \partial \overline{x}_j} = \frac{\partial^2 \xi}{\partial x_j \partial x_k} + \varepsilon \left(\sum_{i=1}^n x_i^* \frac{\partial}{\partial x_i} \left(\frac{\partial^2 \xi}{\partial x_j \partial x_k} \right) + \frac{\partial^2 \widetilde{\xi}}{\partial x_j \partial x_k} \right). \tag{8}
$$

On the other hand, it is easy to compute

$$
\frac{\partial^2 \overline{\xi}}{\partial \overline{x}_j \partial \overline{x}_k} = \frac{\partial^2 \xi}{\partial x_j \partial x_k} + \varepsilon \left(\sum_{i=1}^n x_i^* \frac{\partial}{\partial x_i} \left(\frac{\partial^2 \xi}{\partial x_j \partial x_k} \right) + \frac{\partial^2 \widetilde{\xi}}{\partial x_j \partial x_k} \right). \tag{9}
$$

Comparing these two equations [\(8\)](#page-212-0) and [\(9\)](#page-212-1), we have $\frac{\partial^2 \bar{\xi}}{\partial \bar{\xi}}$ $\frac{\partial^2 \overline{\xi}}{\partial \overline{x}_k \partial \overline{x}_j} \,=\, \frac{\partial^2 \overline{\xi}}{\partial \overline{x}_j \partial \overline{x}_j}$ $\frac{\partial}{\partial \overline{x}_j \partial \overline{x}_k}$. Thus, this achieves the proof. \Box

Remark 1. On the set $\mathbb{R}^n \times \mathbb{R}^n = \{(x, x^*) \mid x, x^* \in \mathbb{R}^n\}$, the equality, inner operation and external operation can be defined as follows:

(i) For any (x, x^*) , $(y, y^*) \in \mathbb{R}^n \times \mathbb{R}^n$, we get

 $(x, x^*) = (y, y^*) \Leftrightarrow x = y \text{ and } x^* = y^*.$

 $(ii) +_1 : (\mathbb{R}^n \times \mathbb{R}^n) \times (\mathbb{R}^n \times \mathbb{R}^n) \to \mathbb{R}^n \times \mathbb{R}^n$, for $(x, x^*), (y, y^*) \in \mathbb{R}^n \times \mathbb{R}^n$, we get

$$
(x, x^*) +_1 (y, y^*) = (x + y, x^* + y^*).
$$

(iii) $\cdot_1 : D \times (\mathbb{R}^n \times \mathbb{R}^n) \to \mathbb{R}^n \times \mathbb{R}^n$, for $(x, x^*) \in \mathbb{R}^n \times \mathbb{R}^n$ and $\overline{\lambda} = \lambda + \varepsilon \lambda^* \in D$, we get

$$
\overline{\lambda} \cdot_1 (x, x^*) = (\lambda x, \lambda x^* + \lambda^* x).
$$

According to the above operations, the set $(\mathbb{R}^n \times \mathbb{R}^n, +_1, \cdot_1)$ constitutes a module over the set $(D, +, \cdot)$.

We are now ready to express the following theorem:

Theorem 5. Let the sets $(\mathbb{R}^n \times \mathbb{R}^n, +_1, \cdot_1)$ and $(D^n, +, \cdot)$ be modules over the set $(D, +, \cdot)$. Then the function $f : \mathbb{R}^n \times \mathbb{R}^n \to D^n$, $f(x, x^*) = x + \varepsilon x^*$ is a (module) isomorphism.

Proof. It is easy to check that f is bijective function. Now, for (x, x^*) , $(y, y^*) \in$ $\mathbb{R}^n \times \mathbb{R}^n$ and $\overline{\lambda} = \lambda + \varepsilon \lambda^* \in D$, the following equality can be written

$$
f(\overline{\lambda} \cdot_1 (x, x^*) +_1 (y, y^*)) = f(\lambda x + y, \lambda x^* + \lambda^* x + y^*)
$$

= $\lambda x + y + \varepsilon (\lambda x^* + \lambda^* x + y^*)$
= $(\lambda + \varepsilon \lambda^*) (x + \varepsilon x^*) + (y + \varepsilon y^*)$
= $\overline{\lambda} f(x, x^*) + f(y, y^*)$

such that f is a (module) linear function. In view of these conventions, we deduce that f is a (module) isomorphism. This permits us to conclude the proof. \Box

Theorem 6. The real vector space \mathbb{R}^n is isomorphic to a subset of D^n defined as $\overrightarrow{A} = \left\{ \overrightarrow{x} = \overrightarrow{x} + \varepsilon \overrightarrow{0} \mid \overrightarrow{x} \in \mathbb{R}^n \right\}$ [\[32\]](#page-225-0).

Definition 3. Let $\{x_1, ..., x_n, x_1^*, ..., x_n^*\}$ be coordinate functions of \mathbb{R}^{2n} and $\tilde{p} = (n_1, ..., n_n)^*$ $\subset \mathbb{R}^{2n}$ Then we have $(p_1, ..., p_n, p_1^*, ..., p_n^*) \in \mathbb{R}^{2n}$. Then we have

$$
\widetilde{x}_i = (x_i, x_i^*) : \mathbb{R}^{2n} \to \mathbb{R} \times \mathbb{R}, \ \widetilde{x}_i \left(\widetilde{p} \right) = (x_i \left(\widetilde{p} \right), x_i^* \left(\widetilde{p} \right)),
$$

where $x_i : \mathbb{R}^{2n} \to \mathbb{R}$, $x_i(\widetilde{p}) = p_i$ and $x_i^* : \mathbb{R}^{2n} \to \mathbb{R}$, $x_i^*(\widetilde{p}) = p_i^*$. Since the function
 $b \mapsto \mathbb{R}^n \to \mathbb{R}^{2n}$ $b \in \mathbb{R}^{2n}$ is bijective function, we can write the following diagram: $h_n: \mathbb{D}^n \to \mathbb{R}^{2n}, h_n(\overline{p}) = \widetilde{p}$ is bijective function, we can write the following diagram:

$$
\begin{array}{ccc}\nD^n & \xrightarrow{\overline{x}_i} & D \\
h_n \downarrow & & \downarrow h_1 \\
\mathbb{R}^{2n} & \xrightarrow{\overline{x}_i} & \mathbb{R} \times \mathbb{R}\n\end{array}
$$

such that dual coordinate functions \overline{x}_i can be stated by $\overline{x}_i = h_1^{-1} \circ \widetilde{x}_i \circ h_n$. Therefore,
for dual coordinate functions \overline{x}_i (1 < i < x) and obtain for dual coordinate functions \overline{x}_i $(1 \leq i \leq n)$, we obtain

$$
\overline{x}_{i}(\overline{p}) = x_{i}(\widetilde{p}) + \varepsilon x_{i}^{*}(\widetilde{p}) = p_{i} + \varepsilon p_{i}^{*} = \overline{p}_{i},
$$

where $\overline{p} = (\overline{p}_1, ..., \overline{p}_n) \in D^n$ and $\overline{p}_i = p_i + \varepsilon p_i^* \in D$.

Definition 4. Suppose that $\overline{p} \in D^n$ is a dual point and $\overrightarrow{x} \in D^n$ is a dual vector. On the set

$$
T_{\overline{p}}D^n = {\overline{p}} \times D^n = {\overline{p}} \left(\overline{p}, \overrightarrow{x} \right) | \overrightarrow{x} \in D^n,
$$

equality, inner operation and external operation can be determined as follows:

(i) For any $\left(\overline{p}, \overrightarrow{x}\right)$ and $\left(\overline{q}, \overrightarrow{y}\right)$, we have $\left(\overline{p}, \overrightarrow{x}\right) = \left(\overline{q}, \overrightarrow{y}\right) \Leftrightarrow \overline{p} = \overline{q} \text{ and } \overrightarrow{x} = \overrightarrow{y}.$ $(ii) \oplus : T_{\overline{p}}D^n \times T_{\overline{p}}D^n \to T_{\overline{p}}D^n$, for $(\overline{p}, \overrightarrow{x})$, $(\overline{p}, \overrightarrow{y}) \in T_{\overline{p}}D^n$, we have $\left(\overrightarrow{p},\overrightarrow{x}\right)\oplus\left(\overrightarrow{p},\overrightarrow{y}\right)=\left(\overrightarrow{p},\overrightarrow{x}+\overrightarrow{y}\right).$ $(iii) \odot : D \times T_{\overline{p}}D^n \to T_{\overline{p}}D^n$ for $(\overline{p}, \overrightarrow{x}) \in T_{\overline{p}}D^n$ and $\overline{\lambda} \in D$, we have $\overline{\lambda} \odot (\overline{p}, \overrightarrow{\overline{x}}) = (\overline{p}, \overrightarrow{\lambda} \overrightarrow{\overline{x}}).$

Corollary 3. Taking into account the operations \oplus and \odot defined on the set $T_{\overline{p}}D^n = {\overline{p}} \times D^n = {\{\overline{p}, \overrightarrow{x}\} \mid \overrightarrow{x} \in D^n\},\$ this set generates a module over the set $(D, +, \cdot)$. This module $(T_p D^n, \oplus, (D, +, \cdot), \odot)$ is called dual tangent space and every element of this module is entitled by dual tangent vector.

Corollary 4. When above defined operations \oplus and \odot is taken into consideration, every element $\overrightarrow{x}_{\overline{p}} = (\overrightarrow{p}, \overrightarrow{x})$ of $T_{\overline{p}}D^n$ can be expressed by

$$
\overrightarrow{\overline{x}}_{\overline{p}} = (\overline{p}, \overrightarrow{x} + \varepsilon \overrightarrow{0}) \oplus \varepsilon \odot (\overline{p}, \overrightarrow{x}^* + \varepsilon \overrightarrow{0}),
$$

where $\overrightarrow{x} = \overrightarrow{x} + \varepsilon \overrightarrow{x}^* \in D^n$.

Corollary 5. Let us define the sets

$$
\Phi = \left\{ \left(\overline{p},\overrightarrow{x}+\varepsilon \overrightarrow{0} \right) \mid \overline{p} \in \mathcal{D}^n, \overrightarrow{x} \in \mathbb{R}^n \right\}
$$

and

$$
\Psi = \{(\widetilde{p}, (x_1, ..., x_n, 0, ..., 0)) \mid \widetilde{p} \in \mathbb{R}^{2n}, x_i \in \mathbb{R}\}.
$$

The inner operation on the set Φ (resp. Ψ) is

$$
\left(\overline{p},\overrightarrow{x}+\varepsilon\overrightarrow{0}\right)+_{\Phi}\left(\overline{p},\overrightarrow{y}+\varepsilon\overrightarrow{0}\right) = \left(\overline{p},\overrightarrow{x}+\overrightarrow{y}+\varepsilon\overrightarrow{0}\right),
$$

$$
\left(\widetilde{p},\left(x_1,...,x_n,0,...,0\right)\right)+_{\Psi}\left(\widetilde{p},\left(y_1,...,y_n,0,...,0\right)\right) = \left(\widetilde{p},\left(x_1+y_1,...,x_n+y_n,0,...,0\right)\right)
$$

and for $\lambda \in \mathbb{R}$, the external operation on the set Φ (resp. Ψ) is

$$
\lambda \cdot_{\Phi} \left(\overline{p}, \overrightarrow{x} + \varepsilon \overrightarrow{0} \right) = \left(\overline{p}, \lambda \overrightarrow{x} + \varepsilon \overrightarrow{0} \right),
$$

$$
\lambda \cdot_{\Psi} \left(\widetilde{p}, (x_1, ..., x_n, 0, ..., 0) \right) = \left(\widetilde{p}, (\lambda x_1, ..., \lambda x_n, 0, ..., 0) \right)
$$

such that the sets $(\Phi, +\Phi, \cdot\Phi)$ and $(\Psi, +\Psi, \cdot\Psi)$ are n-dimensional vector spaces over the field $(\mathbb{R}, +, \cdot).$

With these conventions, the following theorem can be given:

Theorem 7. Consider that

$$
\Phi = \left\{ \left(\overline{p},\overrightarrow{x}+\varepsilon \overrightarrow{0} \right) \mid \overline{p} \in \mathcal{D}^n, \overrightarrow{x} \in \mathbb{R}^n \right\}
$$

and

$$
\Psi = \{(\widetilde{p}, (x_1, ..., x_n, 0, ..., 0)) \mid \widetilde{p} \in \mathbb{R}^{2n}, x_i \in \mathbb{R}\}.
$$

Then the function $g : (\Phi, +\Phi, \cdot\Phi) \to (\Psi, +\Psi, \cdot\Psi), g\left(\overline{p}, \overrightarrow{x} + \varepsilon \overrightarrow{0}\right) = (\widetilde{p}, (x_1, ..., x_n, 0, ..., 0))$ is a isomorphism.

Corollary 6. From Theorem 7 and $\vec{x} = (x_1, ..., x_n) \tilde{=} (x_1, ..., x_n, 0, ..., 0),$ every dual vector $\overrightarrow{x}_{\overline{p}} = (\overrightarrow{p}, \overrightarrow{x}) \in T_{\overline{p}}D^n$ can be written as

$$
\begin{array}{rcl}\n\overrightarrow{x}_{\overrightarrow{p}} & = & \left(\overrightarrow{p}, \overrightarrow{\overrightarrow{x}}\right) \\
& = & \left(\overrightarrow{p}, \overrightarrow{x} + \varepsilon \overrightarrow{0}\right) \oplus \varepsilon \odot \left(\overrightarrow{p}, \overrightarrow{x}^* + \varepsilon \overrightarrow{0}\right) \\
& = & \left(\widetilde{p}, \overrightarrow{x}\right) \oplus \varepsilon \odot \left(\widetilde{p}, \overrightarrow{x}^*\right) \\
& = & \overrightarrow{x}_{\overrightarrow{p}} \oplus \varepsilon \odot \overrightarrow{x}_{\overrightarrow{p}}^*.\n\end{array}
$$

For simplicity, throughout this paper, the operations $+$ and \cdot is used instead of the operations \oplus and \odot , respectively. Thus, this means that

$$
\overrightarrow{\overline{x}}_{\overline{p}} = \overrightarrow{x}_{\widetilde{p}} + \varepsilon \overrightarrow{x}_{\widetilde{p}}^*
$$

.

Also, it is possible to write the following equality:

$$
\overrightarrow{\overline{x}}_{\overline{p}} = (\overline{p}, \overrightarrow{\overline{x}}) = (\overline{p}, \overline{x}_1 \overrightarrow{\overline{e}}_1 + \dots + \overline{x}_n \overrightarrow{\overline{e}}_n) = \overline{x}_1 \overrightarrow{\overline{e}}_{1\overline{p}} + \dots + \overline{x}_n \overrightarrow{\overline{e}}_{n\overline{p}},
$$

where $\vec{\vec{e}}_{i\overline{p}} = (\overline{p}, \overrightarrow{e}_i + \varepsilon \overrightarrow{0})$. Moreover, the equation $\overline{\lambda}_1 \vec{\vec{e}}_{1\overline{p}} + ... + \overline{\lambda}_n \vec{\vec{e}}_{n\overline{p}} =$ −→ $\overline{0}$ $_{\overline{p}}$ can only be satisfied by $\overline{\lambda}_i = \overline{0}$ for $1 \leq i \leq n$. Thus, the set $\left\{ \overrightarrow{e}_{1\overline{p}}, ..., \overrightarrow{e}_{n\overline{p}} \right\}$ forms a basis of dual tangent space $T_{\overline{n}}D^n$.

Theorem 8. Assume that $\overline{\xi} \in C(\overline{U} \subseteq D^n, D)$ and $\overrightarrow{x}_{\overline{p}} \in T_{\overline{p}}D^n$. The derivative of dual analytic function $\overline{\xi}$ in the direction of dual tangent vector $\overrightarrow{x}_{\overline{p}}$ is

$$
\begin{aligned}\n\overrightarrow{x}_{\overline{p}}\left[\overline{\xi}\right] &= \frac{d}{d\overline{t}}\overline{\xi}\left(\overline{p} + \overline{t}\overrightarrow{\overline{x}}\right)_{|\overline{t} = \overline{0}} = \overrightarrow{x}_{\widetilde{p}}\left[\xi\right] + \varepsilon \left(\sum_{i=1}^{n} p_{i}^{*} \overrightarrow{x}_{\widetilde{p}}\left[\xi_{x_{i}}\right] + \overrightarrow{x}_{\widetilde{p}}\left[\widetilde{\xi}\right] + \overrightarrow{x}_{\widetilde{p}}^{*}\left[\xi\right]\right), \\
where \overrightarrow{x}_{\widetilde{p}}\left[\xi\right] &= \sum_{i=1}^{n} \frac{\partial \xi}{\partial x_{i}}\left(x\left(\widetilde{p}\right)\right)x_{i}.\n\end{aligned}
$$

Proof. The proof can be easily made using definitions of dual tangent vector and composition of dual analytic functions. □
Theorem 9. For $\overline{\xi}, \overline{\mu} \in C$ $(\overline{U} \subseteq D^n, D)$, $\overrightarrow{\overline{x}}_{\overline{p}}, \overrightarrow{\overline{y}}_{\overline{p}} \in T_{\overline{p}}D^n$ and $\overline{\lambda} \in D$, the following equalities exist:

$$
\begin{array}{ll}\n\text{(i)} & \overrightarrow{x}_{\overline{p}} \left[\overline{\xi} + \overline{\mu} \right] = \overrightarrow{x}_{\overline{p}} \left[\overline{\xi} \right] + \overrightarrow{x}_{\overline{p}} \left[\overline{\mu} \right] \\
\text{(ii)} & \overrightarrow{x}_{\overline{p}} \left[\overline{\lambda} \cdot \overline{\xi} \right] = \overrightarrow{\lambda} \overrightarrow{x}_{\overline{p}} \left[\overline{\xi} \right] \\
\text{(iii)} & \overrightarrow{x}_{\overline{p}} \left[\overline{\xi} \cdot \overline{\mu} \right] = \overrightarrow{x}_{\overline{p}} \left[\overline{\xi} \right] \overrightarrow{\mu} \left(\overline{p} \right) + \overline{\xi} \left(\overline{p} \right) \overrightarrow{x}_{\overline{p}} \left[\overline{\mu} \right] \\
\text{(iv)} & \left(\overrightarrow{x}_{\overline{p}} + \overrightarrow{y}_{\overline{p}} \right) \left[\overline{\xi} \right] = \overrightarrow{x}_{\overline{p}} \left[\overline{\xi} \right] + \overrightarrow{y}_{\overline{p}} \left[\overline{\xi} \right].\n\end{array}
$$

Definition 5. A dual vector field \overline{X} on D^n is a function that assigns to each dual **Definition 5.** There is the clear function of the condition $\overrightarrow{X}_{\overrightarrow{p}}$ to D^n at \overrightarrow{p} , i.e., $\overrightarrow{X}: D^n \to TD^n$,

$$
\overline{X}(\overline{p}) = \overrightarrow{\overline{X}}_{\overline{p}} = \overrightarrow{X}_{\widetilde{p}} + \varepsilon \overrightarrow{X}_{\widetilde{p}}^* ,
$$

where −→ $\overrightarrow{X} = \overrightarrow{X} + \varepsilon \overrightarrow{X}^*$. Suppose that $\overline{a}_i : \overline{U} \subseteq D^n \to D$, $\overline{a}_i = a_i + \varepsilon a_i^0$ $(1 \leq i \leq n)$ are dual analytic functions. When the dual vector field can be written in the form $\overline{X}(\overline{x}) = (\overline{a}_1(\overline{x}), ..., \overline{a}_n(\overline{x}))$, the equality can be rearranged as follows:

$$
\overline{X}(\overline{x}) = X(x) + \varepsilon \left(\sum_{j=1}^{n} x_j^* X_{x_j} + \widetilde{X}(x) \right),
$$

where $X(x) = (a_1(x), ..., a_n(x)), \ \tilde{X}(x) = (\tilde{a}_1(x), ..., \tilde{a}_n(x))$ and the functions a_i and \widetilde{a}_i belong to $C^{\infty}-class$ for $1 \leq i \leq n$. The set of dual analytic vector fields is symbolized as $\chi(D^n)$. Hence, it is possible to write below expression:

$$
\chi(D^n) = \left\{ \overline{X} \mid \overline{X} : D^n \to TD^n, \ \overrightarrow{X}_{\overline{p}} = \overrightarrow{X}_{\widetilde{p}} + \varepsilon \overrightarrow{X}_{\widetilde{p}}^* \right\}.
$$

We are now ready to introduce that the inner and external operations on $\chi(\mathbf{D}^n)$ is described as below: \Rightarrow \Rightarrow

$$
(i) +: \chi(D^n) \times \chi(D^n) \to \chi(D^n), \text{ for } \overline{X}, \overline{Y} \in \chi(D^n) \text{ and } \overline{p} \in D^n, \text{ we have}
$$

$$
(\overline{X} + \overline{Y}) (\overline{p}) = \overline{\overline{X}}_{\overline{p}} + \overline{\overline{Y}}_{\overline{p}}.
$$

$$
(ii) : D \times \chi(D^n) \to \chi(D^n), \text{ for } \overline{\overline{X}} \in \chi(D^n), \overline{\lambda} \in D \text{ and } \overline{p} \in D^n, \text{ we have}
$$

$$
(\overline{\lambda} \cdot \overline{X}) (\overline{p}) = \overline{\lambda} \cdot \overline{X} (\overline{p}) = \overline{\lambda} \cdot \overline{X}_{\overline{p}}.
$$

In view of above mentioned operations, the set $(\chi(D^n), +, \cdot)$ forms a module over the set $(D, +, \cdot)$.

Now, suppose that $\bar{\xi} \in C(\bar{U} \subseteq D^n, D)$. The derivative of dual analytic function $\overline{\xi}$ in the direction of dual analytic vector field \overline{X} is

$$
\overline{X}\left[\ \overline{\xi}\ \right]=X\left[\xi\right]+\varepsilon\left(\sum_{j=1}^n x_j^*\left(X\left[\xi\right]\right)_{x_j}+X\left[\widetilde{\xi}\right]+\widetilde{X}\left[\xi\right]\right),\,
$$

where $X[\xi] = \sum_{n=1}^{\infty}$ $i=1$ ∂ξ $\frac{\partial \zeta}{\partial x_i} a_i$, such that $\overline{X} \left[\overline{\xi} \right] \in C \left(\overline{U} \subseteq D^n, D \right)$. We can infer that for $\overline{p} \in D^n$,

$$
\overline{X}_{\overline{p}}\left[\ \overline{\xi}\ \right]=X_{\widetilde{p}}\left[\xi\right]+\varepsilon\left(\sum_{j=1}^nx_j^*\left(\widetilde{p}\right)\left(X\left[\xi\right]\right)_{x_j}\left(\widetilde{p}\right)+X_{\widetilde{p}}\left[\widetilde{\xi}\right]+\widetilde{X}_{\widetilde{p}}\left[\xi\right]\right).
$$

Definition 6. Suppose that $\overline{\xi}$: $\overline{U} \subseteq D^n \to D^m$, $\overline{\xi} = (\overline{\xi}_1, ..., \overline{\xi}_m)$ is a dual analytic function. The function $\overline{\xi}_{*\overline{p}}: T_{\overline{p}}\overline{U} \to T_{\overline{\xi}(\overline{p})}D^m$ is called a dual tangent map of the function $\overline{\xi}$ at the dual point \overline{p} , where

$$
\overline{\xi}_{*\overline{p}}\left(\overrightarrow{x}_{\overline{p}}\right) = \left(\overrightarrow{x}_{\overline{p}}\left[\overline{\xi}_{1}\right], \dots, \overrightarrow{x}_{\overline{p}}\left[\overline{\xi}_{m}\right]\right)|_{\overline{q} = \overline{\xi}(\overline{p})}
$$
\n
$$
= \xi_{*\widetilde{p}}\left(\overrightarrow{x}_{\widetilde{p}}\right) + \varepsilon \left(\sum_{j=1}^{n} p_{j}^{*}\xi_{x_{j}*\widetilde{p}}\left(\overrightarrow{x}_{\widetilde{p}}\right) + \widetilde{\xi}_{*\widetilde{p}}\left(\overrightarrow{x}_{\widetilde{p}}\right) + \xi_{*\widetilde{p}}\left(\overrightarrow{x}_{\widetilde{p}}^{*}\right)\right)
$$

and

$$
\xi_{\ast\widetilde{p}}\left(\overrightarrow{x}_{\widetilde{p}}\right)=\left(\overrightarrow{x}_{\widetilde{p}}\left[\xi_{1}\right],...,\overrightarrow{x}_{\widetilde{p}}\left[\xi_{m}\right]\right).
$$

In that case, it turns out that $\overline{\xi}_* : \chi({\bf D}^n) \to \chi({\bf D}^m)$,

$$
\overline{\xi}_{*}\left(\overline{X}\right) = \xi_{*}\left(X\right) + \varepsilon \left(\sum_{j=1}^{n} x_{j}^{*}\left(\xi_{*}\left(X\right)\right)_{x_{j}} + \widetilde{\xi}_{*}\left(X\right) + \xi_{*}\left(\widetilde{X}\right)\right),
$$

where $\xi_*(X) = (X [\xi_1], ..., X [\xi_m]).$

Theorem 10. $\bar{\xi}_{*\bar{p}}: T_{\bar{p}}D^n \to T_{\bar{\xi}(\bar{p})}D^m$ is a (module) linear map and the matrix (dual Jacobian matrix) corresponding to this linear map with respect to the bases $\left\{\overrightarrow{e}_{1\overrightarrow{p}},...,\overrightarrow{e}_{n\overrightarrow{p}}\right\}$ and $\left\{\overrightarrow{e}_{1\overrightarrow{q}},...,\overrightarrow{e}_{m\overrightarrow{q}}\right\}$ is

$$
\overline{J}(\overline{\xi},\overline{p}) = \begin{bmatrix} \frac{\partial \xi_1}{\partial x_1}(\widehat{p}) & \cdots & \frac{\partial \xi_1}{\partial x_n}(\widehat{p}) \\ \frac{\partial \xi_2}{\partial x_1}(\widehat{p}) & \cdots & \frac{\partial \xi_2}{\partial x_n}(\widehat{p}) \\ \vdots & \ddots & \vdots \\ \frac{\partial \xi_m}{\partial x_1}(\widehat{p}) & \cdots & \frac{\partial \xi_m}{\partial x_n}(\widehat{p}) \end{bmatrix} + \varepsilon \begin{bmatrix} \frac{\partial \xi_1^0}{\partial x_1}(\widehat{p}) & \cdots & \frac{\partial \xi_1^0}{\partial x_n}(\widehat{p}) \\ \frac{\partial \xi_2^0}{\partial x_1}(\widehat{p}) & \cdots & \frac{\partial \xi_m}{\partial x_n}(\widehat{p}) \\ \vdots & \ddots & \vdots \\ \frac{\partial \xi_m}{\partial x_1}(\widehat{p}) & \cdots & \frac{\partial \xi_m}{\partial x_n}(\widehat{p}) \end{bmatrix}
$$

\n
$$
= J(\xi,\widehat{p}) + \varepsilon \left(\sum_{j=1}^n p_j^* J(\xi_{x_j},\widehat{p}) + J(\widetilde{\xi},\widehat{p}) \right),
$$

\nwhere $\frac{\partial \xi_j^0}{\partial x_i} = \sum_{k=1}^n x_k^* \frac{\partial^2 \xi_j}{\partial x_i \partial x_k} + \frac{\partial \widetilde{\xi}_j}{\partial x_i}.$

Remark 2. Assume that $\overline{\xi}$: $\overline{U} \subseteq D^n \to D$ is dual analytic function, where $\overline{U} =$ $U \times \mathbb{R}^n$. Then we know that

$$
\overline{\xi}(\overline{x}) = \xi(x_1, ..., x_n) + \varepsilon \left(\sum_{i=1}^n x_i^* \frac{\partial \xi}{\partial x_i}(x_1, ..., x_n) + \widetilde{\xi}(x_1, ..., x_n) \right).
$$

The value of this function at the dual point $\overline{x} = \overline{p}$ is

$$
\overline{\xi}(\overline{p}) = \xi(x_1(\widetilde{p}), ..., x_n(\widetilde{p})) + \varepsilon \left(\sum_{i=1}^n x_i^*(\widetilde{p}) \frac{\partial \xi}{\partial x_i}(x_1(\widetilde{p}), ..., x_n(\widetilde{p})) + \widetilde{\xi}(x_1(\widetilde{p}), ..., x_n(\widetilde{p})) \right)
$$

$$
= \xi(p_1, ..., p_n) + \varepsilon \left(\sum_{i=1}^n p_i^* \frac{\partial \xi}{\partial x_i}(p_1, ..., p_n) + \widetilde{\xi}(p_1, ..., p_n) \right).
$$

As a result, the functions ξ and $\tilde{\xi}$ can be reduced to the functions defined from $U \subseteq \mathbb{R}^n$ to $\mathbb R$ such that these functions belong to $C^{\infty}-class$.

Definition 7. Assume that $\overline{\xi}$: $\overline{U} \subseteq D \to D$, $\overline{\xi}(\overline{x}) = \xi(x) + \varepsilon(x^* \xi'(x) + \widetilde{\xi}(x))$ is a dual analytic function and $\xi'(x)$ is not zero for all $x \in U \subseteq \mathbb{R}$. If the equality $\overline{\xi}(\overline{x}_1) = \overline{\xi}(\overline{x}_2)$ requires the equality $\overline{x}_1 = \overline{x}_2$ for all $\overline{x}_1, \overline{x}_2 \in \overline{U} \subseteq D$, then the function ξ is called injective function.

Theorem 11. Assume that $\overline{\xi}$: $\overline{U} \subseteq D \to D$, $\overline{\xi}(\overline{x}) = \xi(x) + \varepsilon \left(x^* \xi'(x) + \widetilde{\xi}(x)\right)$ is a dual analytic function and $\xi'(x)$ is not zero for all $x \in U \subseteq \mathbb{R}$. Then ξ is injective function if and only if the dual analytic function $\overline{\xi}$ is injective function.

Proof. Suppose that $\overline{\xi} : \overline{U} \subseteq D \to D$, $\overline{\xi}(\overline{x}) = \xi(x) + \varepsilon(x^* \xi'(x) + \widetilde{\xi}(x))$ is a dual analytic function, $\xi'(x)$ is not zero for all $x \in U \subseteq \mathbb{R}$ and ξ is injective function. Assume that there exists the equality $\overline{\xi}(\overline{x}_1) = \overline{\xi}(\overline{x}_2)$ for all $\overline{x}_1, \overline{x}_2 \in \overline{U} \subseteq D$. From the definition of dual analytic functions, the following equality can be written:

$$
\overline{\xi}(\overline{x}_1) = \xi(x_1) + \varepsilon \left(x_1^* \xi'(x_1) + \widetilde{\xi}(x_1) \right) = \xi(x_2) + \varepsilon \left(x_2^* \xi'(x_2) + \widetilde{\xi}(x_2) \right) = \overline{\xi}(\overline{x}_2)
$$
\nwhich implies

which implies

$$
\xi(x_1) = \xi(x_2)
$$

and

$$
x_1^* \xi'(x_1) + \widetilde{\xi}(x_1) = x_2^* \xi'(x_2) + \widetilde{\xi}(x_2).
$$

From the hypothesis, since ξ is an injective function, it is clear that $x_1 = x_2$. On the other hand, we have $\tilde{\xi}(x_1) = \tilde{\xi}(x_2)$, since $\tilde{\xi}$ is a well-defined function. Hence, since $\xi'(x)$ is not zero for all $x \in U \subseteq \mathbb{R}$ and $\tilde{\xi}(x_1) = \tilde{\xi}(x_2)$, it is easily seen that $x_1^* = x_2^*$. That is to say, $\overline{x}_1 = \overline{x}_2$. Therefore, $\overline{\xi}$ is an injective function.

Conversely, we shall prove this part of theorem by means of contrapositive method. Assume that ξ is not an injective function. That is to say, the equality $\xi(x_1) = \xi(x_2)$ requires the inequality $x_1 \neq x_2$ for at least $x_1, x_2 \in U \subseteq \mathbb{R}$. We must show that dual analytic function $\bar{\xi}$ is not an injective function. It is enough to remark that the equality $\overline{\xi}(\overline{x}_1) = \overline{\xi}(\overline{x}_2)$ requires $\overline{x}_1 \neq \overline{x}_2$ for at least $\overline{x}_1, \overline{x}_2 \in \overline{U} \subseteq D$. Suppose that there exists the equality $\overline{\xi}(\overline{x}_1) = \overline{\xi}(\overline{x}_2)$ for at least $\overline{x}_1, \overline{x}_2 \in \overline{U} \subseteq D$. Thus, this gives rise to the relation

$$
\overline{\xi}(\overline{x}_1) = \xi(x_1) + \varepsilon \left(x_1^* \xi'(x_1) + \widetilde{\xi}(x_1) \right) = \xi(x_2) + \varepsilon \left(x_2^* \xi'(x_2) + \widetilde{\xi}(x_2) \right) = \overline{\xi}(\overline{x}_2),
$$

i.e.,

$$
\xi(x_1) = \xi(x_2)
$$

and

$$
x_1^* \xi'(x_1) + \widetilde{\xi}(x_1) = x_2^* \xi'(x_2) + \widetilde{\xi}(x_2).
$$

As already known, since ξ is not an injective function, the equality $\overline{\xi}(\overline{x}_1) = \overline{\xi}(\overline{x}_2)$ requires the expression $\overline{x}_1 \neq \overline{x}_2$ for at least $\overline{x}_1, \overline{x}_2 \in \overline{U} \subseteq D$, that is, dual analytic function $\overline{\xi}$ is not an injective function. Therefore, the proof is completed.

Definition 8. Assume that $\overline{\xi}$: $\overline{U} \subseteq D \to \overline{V} \subseteq D$, $\overline{\xi}(\overline{x}) = \xi(x) + \varepsilon(x^* \xi'(x) + \widetilde{\xi}(x))$ is a dual analytic function and $\xi'(x)$ is not zero for all $x \in \mathcal{U} \subseteq \mathbb{R}$. If there exists at least one $\overline{x} = x + \varepsilon x^* \in \overline{U} \subseteq D$ satisfying the equality $\overline{y} = \overline{\xi}(\overline{x})$ for all $\overline{y} = y + \varepsilon y^* \in \overline{V} \subseteq D$, then the dual analytic function $\overline{\xi}$ is called surjective function. That is to say, if ξ is surjective function and there exists $x^* \in \mathbb{R}$ satisfying the equality $x^* = \frac{y^* - \tilde{\xi}(x)}{\tilde{\xi}(x)}$ $\frac{-g(x)}{\xi'(x)}$ for all $y^* \in \mathbb{R}$, then dual analytic function $\overline{\xi}$ is called surjective function.

Definition 9. Assume that $\overline{\xi} : \overline{U} \subseteq D \to \overline{V} \subseteq D, \overline{\xi}(\overline{x}) = \xi(x) + \varepsilon(x^* \xi'(x) + \widetilde{\xi}(x))$ is a dual analytic function and $\xi'(x)$ is not zero for all $x \in U \subseteq \mathbb{R}$. If dual analytic function $\overline{\xi}$ is bijective function, there is only one function $\overline{\mu} : \overline{V} \subseteq D \to \overline{U} \subseteq D$ satisfying the equalities $(\overline{\mu} \circ \overline{\xi}) (\overline{x}) = \overline{I} (\overline{x})$ and $(\overline{\xi} \circ \overline{\mu}) (\overline{y}) = \overline{I} (\overline{y})$, where \overline{I} is dual unit function. The function $\overline{\mu}$ is called inverse function of dual function $\overline{\xi}$ and the inverse function is symbolized as $\overline{\mu} = \overline{\xi}^{-1}$.

Theorem 12. Assume that $\overline{\xi}$: $\overline{U} \subseteq D \to \overline{\xi}$ $(\overline{U}) \subseteq D$, $\overline{\xi}$ $(\overline{x}) = \xi(x) + \varepsilon \left(x * \xi'(x) + \widetilde{\xi}(x)\right)$ is a dual analytic function and $\xi'(x)$ is not zero for all $x \in U \subseteq \mathbb{R}$. If there exists inverse of dual function $\overline{\xi}$, it is expressed as

$$
\overline{\xi}^{-1}(\overline{y}) = \xi^{-1}(y) + \varepsilon \left(y^* \left(\xi^{-1} \right)'(y) - \left(\widetilde{\xi} \circ \xi^{-1} \right) (y) \cdot \left(\xi^{-1} \right)'(y) \right),
$$

where ξ^{-1} is inverse of real function ξ .

Proof. Suppose that $\overline{\xi} : \overline{U} \subseteq D \to \overline{\xi}(\overline{U}) \subseteq D, \overline{\xi}(\overline{x}) = \xi(x) + \varepsilon(x^* \xi'(x) + \widetilde{\xi}(x))$ is a dual analytic function, $\xi'(x)$ is not zero for all $x \in U \subseteq \mathbb{R}$ and there exists inverse of dual analytic function $\bar{\xi}$. Since the function ξ is bijective function, there

is inverse of function ξ , i.e., ξ^{-1} such that $(\xi^{-1})'(y) = \frac{1}{\xi'(x)}$ for all $x \in U \subseteq \mathbb{R}$. Thus, from the hypothesis, we have

$$
\begin{array}{rcl}\n\left(\overline{\xi}^{-1} \circ \overline{\xi}\right)(\overline{x}) & = & \left(\xi^{-1} \circ \xi\right)(x) + \varepsilon \left(x^* \left(\left(\xi^{-1} \circ \xi\right)'(x)\right) + \widetilde{\xi}(x) \left(\xi^{-1}\right)'(\xi(x))\right) \\
& = & x + \varepsilon x^* \\
& = & \overline{I}(\overline{x}),\n\end{array}
$$

where $\overline{I}: D \to D$, $\overline{I}(\overline{x}) = x + \varepsilon x^*$. Similarly, we get $(\overline{\xi} \circ \overline{\xi}^{-1})(\overline{y}) = \overline{I}(\overline{y})$. From the definition of equality in functions, we can write

$$
\overline{\xi}^{-1} \circ \overline{\xi} = \overline{\xi} \circ \overline{\xi}^{-1} = \overline{I}.
$$

Hence, this achieves the proof. \Box

Corollary 7. If there exists the inverse of dual analytic function $\overline{\xi}$: $\overline{U} \subseteq D \rightarrow$ $\overline{\xi}\left(\overline{U}\right) \subseteq \mathrm{D}, \overline{\xi}\left(\overline{x}\right) = \xi\left(x\right) + \varepsilon\left(x^*\xi'\left(x\right) + \widetilde{\xi}\left(x\right)\right),$ the inverse function is a dual analytic function expressed as follows:

$$
\overline{\xi}^{-1}(\overline{x}) = \xi^{-1}(x) + \varepsilon \left(x^* \left(\xi^{-1} \right)'(x) - \left(\widetilde{\xi} \circ \xi^{-1} \right) (x) \left(\xi^{-1} \right)'(x) \right).
$$

The derivative of this dual analytic function with respect to dual variable \bar{x} is

$$
\frac{d\overline{\xi}^{-1}}{d\overline{x}} = (\xi^{-1})'(x) + \varepsilon \left(\begin{array}{c} x^* \left(\xi^{-1} \right)''(x) - \left(\widetilde{\xi} \circ \xi^{-1} \right) (x) \left(\xi^{-1} \right)''(x) \\ -\widetilde{\xi}' \left(\xi^{-1} (x) \right) \left(\left(\xi^{-1} \right)'(x) \right)^2 \end{array} \right).
$$

Definition 10. Let $\overline{\xi} : \overline{U} \subseteq D^n \to \overline{V} \subseteq D^n$ be a dual analytic function, where $\overline{\xi}(\overline{x}) = (\overline{\xi}_1(\overline{x}), ..., \overline{\xi}_n(\overline{x}))$

$$
= (\xi_1(x), ..., \xi_n(x)) + \varepsilon \left(\left(\sum_{i=1}^n x_i^* \frac{\partial \xi_1}{\partial x_i}, ..., \sum_{i=1}^n x_i^* \frac{\partial \xi_n}{\partial x_i} \right) + \left(\tilde{\xi}_1(x), ..., \tilde{\xi}_n(x) \right) \right).
$$

If there exists the inverse function $\overline{\xi}^{-1}$ being dual analytic function, then the dual analytic function $\overline{\xi}$ is called dual diffeomorphism.

Theorem 13. Assume that $\overline{\xi}: D^n \to D^n$, $\overline{\xi}(\overline{x}) = (\overline{\xi}_1(\overline{x}), ..., \overline{\xi}_n(\overline{x})) = \xi + \varepsilon \xi^0$ is a dual analytic function. If $rank J(\xi, q) = n$ for all $q \in U$, where $\overline{q} = q + \varepsilon q^* \in D^n$ and U is open set in terms of standard topology of \mathbb{R}^n , then there is at least one dual open set $\overline{U} \in \overline{\tau}$ in D^n covering point $\overline{q} \in D^n$ such that $\overline{\xi} \mid_{\overline{U}} : \overline{U} \to \overline{\xi}(\overline{U})$ is dual diffeomorphism.

Proof. Assume that $\overline{\xi}: D^n \to D^n$, $\overline{\xi}(\overline{x}) = (\overline{\xi}_1(\overline{x}), ..., \overline{\xi}_n(\overline{x})) = \xi + \varepsilon \xi^0$ is a dual analytic function and $rank J(\xi, q) = n$ for all $q \in U$. We know that the functions ξ and $\tilde{\xi}$ can be reduced to the functions defined from $U \subseteq \mathbb{R}^n$ to $\mathbb R$ such that these

functions belong to C^{∞} –class. Hence, the function $\bar{\xi}: D^n \to D^n$ can be expressed as

$$
\overline{\xi}(\overline{x}) = \xi(x) + \varepsilon \left(\sum_{j=1}^{n} x_j^* \frac{\partial \xi}{\partial x_j}(x) + \widetilde{\xi}(x) \right).
$$

Since the function $\xi : \mathbb{R}^n \to \mathbb{R}^n$ belongs to C^{∞} –class and rank $J(\xi, q) = n$ for all $q \in U$, $\xi |_{U}: U \to \xi(U)$ is a diffeomorphism.

Suppose that there is the equality $\overline{\xi}(\overline{p}) = \overline{\xi}(\overline{q})$ for all $\overline{p}, \overline{q} \in \overline{U} \subseteq D^n$ $(p, q \in U \subseteq \mathbb{R}^n)$. Hence, we can write

$$
\xi(p) + \varepsilon \left(\sum_{j=1}^n p_j^* \frac{\partial \xi}{\partial x_j}(p) + \widetilde{\xi}(p) \right) = \xi(q) + \varepsilon \left(\sum_{j=1}^n q_j^* \frac{\partial \xi}{\partial x_j}(q) + \widetilde{\xi}(q) \right),
$$

which implies

$$
\xi(p) = \xi(q)
$$

and

$$
\sum_{j=1}^{n} p_j^* \frac{\partial \xi}{\partial x_j} (p) + \tilde{\xi} (p) = \sum_{j=1}^{n} q_j^* \frac{\partial \xi}{\partial x_j} (q) + \tilde{\xi} (q).
$$
 (10)

Since $\xi |_U$ is injective function, we have $p = q$. From the equation [\(10\)](#page-221-0), we get the following equality:

$$
(p_1^* - q_1^*) \frac{\partial \xi}{\partial x_1} + \dots + (p_n^* - q_n^*) \frac{\partial \xi}{\partial x_n} = (0, \dots, 0).
$$

Since the set $\frac{\partial \xi}{\partial x}$ $\frac{\partial \xi}{\partial x_1},...,\frac{\partial \xi}{\partial x_n}$ ∂x_n is linearly independent, we have $p_i^* = q_i^*$ for $1 \leq i \leq n$, i.e., $p^* = q^*$. That is to say, we can write $\overline{p} = p + \varepsilon p^* = q + \varepsilon q^* = \overline{q}$ such that the dual analytic function $\xi |_{\overline{U}}$ is injective.

Now, let us show that there exists at least one $\overline{q} \in \overline{U} \subseteq D^n$ satisfying the equality $\overline{p} = \overline{\xi}(\overline{q})$ for all $\overline{p} \in \overline{\xi}(\overline{U}) \subseteq D^n$. The equality

$$
\overline{p} = p + \varepsilon p^* = \xi(q) + \varepsilon \left(\sum_{j=1}^n q^*_j \frac{\partial \xi}{\partial x_j} (q) + \widetilde{\xi}(q) \right) = \overline{\xi}(\overline{q})
$$

allows us to write

$$
p = \xi(q)
$$

and

$$
p^* = \sum_{j=1}^n q_j^* \frac{\partial \xi}{\partial x_j} (q) + \tilde{\xi} (q).
$$
 (11)

Since $\xi |_{U}$ is bijective function, there exists $q = \xi^{-1}(p) \in U \subseteq \mathbb{R}^{n}$. Expanding the equation ($\boxed{11}$), it is seen that the following linear equation system is obtained:

$$
q_1^* \frac{\partial \xi_1}{\partial x_1} (q) + \dots + q_n^* \frac{\partial \xi_1}{\partial x_n} (q) = p_1^* - \tilde{\xi}_1 (q)
$$

$$
q_1^* \frac{\partial \xi_2}{\partial x_1} (q) + \dots + q_n^* \frac{\partial \xi_2}{\partial x_n} (q) = p_2^* - \tilde{\xi}_2 (q)
$$

...

$$
q_1^* \frac{\partial \xi_n}{\partial x_1} (q) + \ldots + q_n^* \frac{\partial \xi_n}{\partial x_n} (q) = p_n^* - \widetilde{\xi}_n (q).
$$

The matrix form of this linear equation system is

$$
\begin{bmatrix}\n\frac{\partial \xi_1}{\partial x_1}(q) & \frac{\partial \xi_1}{\partial x_2}(q) & \cdots & \frac{\partial \xi_1}{\partial x_n}(q) \\
\frac{\partial \xi_2}{\partial x_1}(q) & \frac{\partial \xi_2}{\partial x_2}(q) & \cdots & \frac{\partial \xi_2}{\partial x_n}(q) \\
\vdots & \vdots & \ddots & \vdots \\
\frac{\partial \xi_n}{\partial x_1}(q) & \frac{\partial \xi_n}{\partial x_2}(q) & \cdots & \frac{\partial \xi_n}{\partial x_n}(q)\n\end{bmatrix}\n\begin{bmatrix}\nq_1^* \\
q_2^* \\
\vdots \\
q_n^*\n\end{bmatrix}\n=\n\begin{bmatrix}\np_1^* - \tilde{\xi}_1(q) \\
p_2^* - \tilde{\xi}_2(q) \\
\vdots \\
p_n^* - \tilde{\xi}_n(q)\n\end{bmatrix}.
$$

If we denote $[A]_{n \times n} = \left[\frac{\partial \xi_i}{\partial x}\right]$ $\frac{\partial \xi_i}{\partial x_j}(q)\bigg]$ and $[B]_{n\times 1} = [p_i^* - \tilde{\xi}_i(q)]$
 $\underset{1 \leq i,j \leq n}{\text{and }} [B]_{n\times 1} = [p_i^* - \tilde{\xi}_i(q)]$ $\sum_{1 \leq i \leq n}$, the above matrix form can be rewritten as

$$
[A]_{n \times n} [q^*]_{n \times 1} = [B]_{n \times 1}.
$$
 (12)

Since $rank J(\xi, q) = n$ for all $q \in U \subseteq \mathbb{R}^n$, there exists an inverse of the matrix $[A]_{n \times n}$ such that $[q^*]_{n \times 1} = [A^{-1}]_{n \times n} [B]_{n \times 1}$. Therefore, there exists dual point $\overline{q} = q + \varepsilon q^* \in \overline{U} \subseteq D^n$, that is, dual analytic function $\overline{\xi} \mid_{\overline{U}}$ is surjective.

With these conventions, there exists the inverse of dual analytic function $\xi|_{\overline{U}}$ and this inverse function is $\overline{\mu} : \overline{\xi}(\overline{U}) \subseteq D^n \to \overline{U} \subseteq D^n$,

$$
\overline{\mu}(\overline{y}) = \mu(y) + \varepsilon \left(\sum_{j=1}^{n} y_j^* \frac{\partial \mu}{\partial y_j} + \widetilde{\mu}(y) \right) = \mu + \varepsilon \mu^0,
$$

where $\mu = (\xi |_{U})^{-1}$ and $\widetilde{\mu}_{i}(y) = \left\langle -\widetilde{\xi}(\mu(y)), \nabla \mu_{i}(y) \right\rangle$ for $1 \leq i \leq n$. This fact can be verified as follows:

$$
\left(\overline{\xi} \mid_{\overline{U}} \circ \overline{\mu}\right)(\overline{y}) = \left(\xi \mid_{U} \circ \mu\right)(y) + \varepsilon \left(\begin{array}{c} y_{1}^{*}\left(\frac{\partial \mu_{1}}{\partial y_{1}} \frac{\partial \xi}{\partial x_{1}}\left(\mu\left(y\right)\right) + \ldots + \frac{\partial \mu_{n}}{\partial y_{1}} \frac{\partial \xi}{\partial x_{n}}\left(\mu\left(y\right)\right)\right) \\ + \ldots + \\ y_{n}^{*}\left(\frac{\partial \mu_{1}}{\partial y_{n}} \frac{\partial \xi}{\partial x_{1}}\left(\mu\left(y\right)\right) + \ldots + \frac{\partial \mu_{n}}{\partial y_{n}} \frac{\partial \xi}{\partial x_{n}}\left(\mu\left(y\right)\right)\right) \\ + \tilde{\mu}_{1}\left(y\right) \frac{\partial \xi}{\partial x_{1}}\left(\mu\left(y\right)\right) + \ldots + \tilde{\mu}_{n}\left(y\right) \frac{\partial \xi}{\partial x_{n}}\left(\mu\left(y\right)\right) \\ + \tilde{\xi}\mid_{U}\left(\mu\left(y\right)\right)\end{array}\right)
$$

$$
= (\xi |_{U} \circ \mu)(y) + \varepsilon \begin{pmatrix} y_{1}^{*} \frac{\partial (\xi \circ \mu)}{\partial y_{1}} + ... + y_{n}^{*} \frac{\partial (\xi \circ \mu)}{\partial y_{n}} \\ -\tilde{\xi}_{1} (\mu(y)) \frac{\partial (\xi \circ \mu)}{\partial y_{1}} - ... -\tilde{\xi}_{n} (\mu(y)) \frac{\partial (\xi \circ \mu)}{\partial y_{n}} \\ +\tilde{\xi} |_{U} (\mu(y)) \end{pmatrix}
$$

= $y + \varepsilon y^{*}$
= $\overline{I}(\overline{y}).$

 \setminus

 $\begin{array}{c} \hline \end{array}$

In analogous to $(\overline{\xi} \mid_{\overline{U}} \circ \overline{\mu}) (\overline{y}) = \overline{I} (\overline{y}),$ it is easy to check that $(\overline{\mu} \circ \overline{\xi} \mid_{\overline{U}}) (\overline{x}) = \overline{I} (\overline{x}).$ On the other hand, the dual function $\bar{\mu} = (\bar{\xi}|_{\bar{U}})^{-1}$ is a dual analytic function, since $\frac{\partial \mu}{\partial x}$ $\overline{\partial y^*_i}$ $= 0$ and $\frac{\partial \mu^0}{\partial y_i^*}$ $=\frac{\partial \mu}{\partial x}$ $\frac{\partial \mu}{\partial y_i}$ for $1 \leq i \leq n$ and the functions μ and $\tilde{\mu}$ belong to C^{∞} –class. That is to say, $\overline{\xi}$ | \overline{U} is dual diffeomorphism. is dual diffeomorphism. \Box

3. CONCLUSION

The relation between some sets of the topology constituted in dual space and regions where dual analytic functions are analytic is explained in this paper. Besides, we can assert that it is possible to construct the concept of dual surface via the expression of the inverse function theorem in dual space.

Author Contribution Statements The authors have made substantial contributions to the analysis and interpretation of the results and they read and approved the final manuscript.

Declaration of Competing Interests The authors declare that they have no known competing financial interest or personal relationships that could have appeared to influence the work reported in this paper.

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(U, V) -LUCAS POLYNOMIAL COEFFICIENT RELATIONS OF THE BI-UNIVALENT FUNCTION CLASS

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Abstract. In geometric function theory, Lucas polynomials and other special polynomials have recently gained importance. In this study, we develop a new family of bi-univalent functions. Also we examined coefficient inequalities and Fekete-Szegö problem for this new family via these polynomials.

1. INTRODUCTION

Let $\mathfrak A$ denote the family of all functions $\theta(\xi)$ that are analytic in the unit disc $\mathfrak{U} = \{\xi : \xi \in C, |\xi| < 1\}$ normalized by the conditions $\theta(0) = \theta'(0) - 1 = 0$. Such a function $\theta(\xi)$ takes the form

$$
\theta(\xi) = \xi + \sum_{r=2}^{\infty} n_r \xi^r \quad (\xi \in \mathfrak{U}).
$$
\n(1)

Assume that S be the subclass of $\mathfrak A$ compose of univalent functions.

As a subclass of \mathfrak{A} , the class of bi-univalent functions was first presented by Lewin [\[18\]](#page-238-0). He indicated that $|n_2| \leq 1.15$. After that, a lot of studies have been made about coefficient estimates. See for example $\frac{4}{10}$, $\frac{1}{11}$, $\frac{1}{4}$, $\frac{1}{15}$, $\frac{27}{30}$, $\frac{30}{11}$. According to the Koebe 1/4 theorem (see [\[12\]](#page-238-5)), the range of every function $\theta \in \mathcal{S}$ contains the disc $d_{\omega} = {\omega : |\omega| < 0.25}$, thus, for all $\theta \in S$ with its inverse θ^{-1} , such that $\theta^{-1}(\theta(\xi)) = \xi \quad (\xi \in \mathfrak{U}) \text{ and } \theta(\theta^{-1}(\omega)) = \omega, \quad (\omega : |\omega| < r_0(\theta); r_0(\theta) \geq 0.25) \text{ where}$ $\theta^{-1}(\omega)$ is expressed as

$$
\vartheta(\omega) = \omega - n_2 \omega^2 + (2n_2^2 - n_3)\omega^3 - (5n_2^3 - 5n_2n_3 + n_4)\omega^4 + \cdots
$$
 (2)

Keywords. (U, V) -Lucas polynomial, bi-univalent analytic function, subordination.

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Thus, a function $\theta \in \mathfrak{A}$ is said that bi-univalent in \mathfrak{U} , if both $\theta(\xi)$ and $\theta^{-1}(\omega)$ are univalent in U. Let we show the class of holomorphic and bi-univalent functions in U by B.

It is known that some similar functions $\theta \in \mathcal{S}$ for instance the Koebe function $\kappa(\xi) = \xi/(1-\xi)^2$, its rotation function $\kappa_{\varsigma}(\xi) = \xi/(1-e^{i\varsigma}\xi)^2$, $\theta(\xi) = \xi - \xi^2/2$ and $\theta(\xi) = \xi/(1-\xi^2)$ are in **B**. Also some functions $\theta \in (\mathcal{S} \cap \mathfrak{B})$ contains $\theta(\xi) = \xi$, $\theta(\xi) = 1/2 \log[(1-\xi)/(1-\xi)], \xi/(1-\xi).$

For the functions $h, H \in \mathfrak{A}$, The function h is said to be subordinate to H or H is said to be superordinate to h, if there exists a function η , analytic in \mathfrak{U} , with $\eta(0) = 0$ and $|\eta(z)| < 1$ and such that $h(\xi) = H(\eta(\xi))$. In such a case we write $h \prec H$ or $h(\xi) \prec H(\xi)$. If h is one-to-one, this $h \prec H$ iff $h(0) = H(0)$ and $h(U) \subset H(U)$. Babalola $\boxed{9}$ studied the class $\mathfrak{L}_{\sigma}(\varphi)$ of σ -pseudo- starlike functions of order $\varphi(0 \leq \varphi < 1)$ which is own geometric conditions fulfill

$$
\Re\left(\frac{\xi(\theta'(\xi))^{\sigma}}{\theta(\xi)}\right)>\varphi.
$$

He discover that every pseudo-starlike functions are Bazilevič of type $\left(1 - \frac{1}{\sigma}\right)$ order $\varphi^{\frac{1}{\sigma}}$ and univalent in \mathfrak{U} .

In recent years, theory and applications of Dickson, Fibonacci, Lucas, Chebyshev, Lucas-Lehmer polynomials in modern science have emerged as a very current subject. These polynomials are important in mathematics due to the fact that they can applicable to number theory, numerical analysis, combinatorics, and other fields. Nowadays, these polyinomials have been studied and different generalizations have been made by many authors: see $[1-\frac{3}{5}]-8$. Also see $[13,17,19,-26,28,29,43,46]$ $[13,17,19,-26,28,29,43,46]$ $[13,17,19,-26,28,29,43,46]$ $[13,17,19,-26,28,29,43,46]$ $[13,17,19,-26,28,29,43,46]$ $[13,17,19,-26,28,29,43,46]$ $[13,17,19,-26,28,29,43,46]$ $[13,17,19,-26,28,29,43,46]$ $[13,17,19,-26,28,29,43,46]$.

We recall some important properties interested in which we use to construct our new class. Assume that polynomials with real coefficients are written by $U(x)$ and $V(x)$. By using the recurrence relation, the (U, V) -Lucas polynomials $\mathcal{L}_{U, V, t}(x)$ are explained [\[17\]](#page-238-10) as

$$
\mathcal{L}_{U,V,t}(x) = U(x)\mathcal{L}_{U,V,t-1}(x) + V(x)\mathcal{L}_{U,V,t-2}(x) \ (t \ge 2).
$$
 (3)

Also

$$
\mathcal{L}_{U,V,0}(x) = 2,\n\mathcal{L}_{U,V,1}(x) = U(x),\n\mathcal{L}_{U,V,2}(x) = U^2(x) + 2V(x),\n\mathcal{L}_{U,V,3}(x) = U^3(x) + 3U(x)V(x).
$$
\n(4)

The generating function of the (U, V) -Lucas polynomial sequence $\mathcal{L}_{U, V, t}(x)$ is expressed by [\[17\]](#page-238-10)

$$
\mathcal{K}_{\{\mathcal{L}_t(x)\}}(\xi) = \sum_{t=0}^{\infty} \mathcal{L}_{U,V,t}(x)\xi^t = \frac{2 - U(x)\xi}{1 - U(x)\xi - V(x)\xi^2}.
$$
\n(5)

In the next section, using this polynomials as a tool, we define the family $\mathfrak{H}^{\mathfrak{B},\beta}(\gamma,\sigma;x)$ as follow:

Definition 1. For $\beta \geq 0$, $\sigma \geq 1$, $|\gamma| \leq 1$ but $\gamma \neq 1$, a function $\theta \in \mathfrak{B}$ is called in the family $\mathfrak{H}^{\mathfrak{B},\beta}(\gamma,\sigma;x)$ if the following subordinations are satisfied:

$$
\frac{((1-\gamma)\xi)^{1-\beta}(\theta'(\xi))^{\sigma}}{(\theta(\xi)-\theta(\gamma\xi))^{1-\beta}} \prec \mathcal{K}_{\{\mathcal{L}_{U,V,t}(x)\}}(\xi)-1
$$
\n(6)

and

$$
\frac{((1-\gamma)\omega)^{1-\beta}(\vartheta'(w))^{\sigma}}{(\vartheta(\omega)-\vartheta(\gamma\omega))^{1-\beta}} \prec \mathcal{K}_{\{\mathcal{L}_{U,V,t}(x)\}}(\omega)-1.
$$
\n(7)

Taking special values for β, γ and σ , the class $\mathfrak{H}^{\mathfrak{B},\beta}(\gamma, \sigma; x)$ reduces some exciting new families:

Remark 1. For $\sigma = 1$, we get the new family $\mathfrak{H}^{\mathfrak{B},\beta}(\gamma,1;x)$. If $\theta \in \mathfrak{B}$, is in $\mathfrak{H}^{\mathfrak{B},\beta}(\gamma,1;x)$ then following condition fulfilles

$$
\frac{((1-\gamma)\xi)^{1-\beta}\theta'(\xi)}{(\theta(\xi)-\theta(\gamma\xi))^{1-\beta}} \prec \mathcal{K}_{\{\mathcal{L}_{U,V,t}(x)\}}(\xi)-1
$$
\n(8)

and

$$
\frac{((1-\gamma)\omega)^{1-\beta}\vartheta'(w)}{(\vartheta(\omega)-\vartheta(\gamma\omega))^{1-\beta}} \prec \mathcal{K}_{\{\mathcal{L}_{U,V,t}(x)\}}(\omega) - 1.
$$
\n(9)

Remark 2. For $\beta = 0$, we obtain the new class

$$
\mathfrak{H}^{\mathfrak{B},0}(\gamma,\sigma;x) = \mathfrak{H}^{\mathfrak{B}}(\gamma,\sigma;x).
$$

If $\theta \in \mathfrak{B}$ is in $\mathfrak{H}^{\mathfrak{B}}(\gamma, \sigma; x)$, then following condition fulfilles

$$
\frac{\xi(1-\gamma)(\theta'(\xi))^{\sigma}}{\theta(\xi)-f(\gamma\xi)} \prec \mathcal{K}_{\{\mathcal{L}_{U,V,t}(x)\}}(\xi)-1
$$
\n(10)

and

$$
\frac{\omega(1-\gamma)(\vartheta'(w))^{\sigma}}{\vartheta(\omega)-\vartheta(\gamma\omega)} \prec \mathcal{K}_{\{\mathcal{L}_{U,V,t}(x)\}}(\omega) - 1.
$$
\n(11)

Also,

(1) Choosing $\sigma = 1$ in the class $\mathfrak{H}^{\mathfrak{B}}(\gamma, \sigma; x)$ we have new family $\mathfrak{H}^{\mathfrak{B}}(\gamma, 1; x) =$ $\mathfrak{H}^{\mathfrak{B}}(\gamma; x)$. The class $\mathfrak{H}^{\mathfrak{B}}(\gamma; x)$ consists of the function $f \in \mathfrak{B}$ fulfilling

$$
\frac{\xi(1-\gamma)\theta'(\xi)}{\theta(\xi)-\theta(\gamma\xi)} \prec \mathcal{K}_{\{\mathcal{L}_{U,V,t}(x)\}}(\xi) - 1
$$
\n(12)

$$
\frac{\omega(1-\gamma)\vartheta'(w)}{\vartheta(\omega)-\vartheta(\gamma\omega)} \prec \mathcal{K}_{\{\mathcal{L}_{U,V,t}(x)\}}(\omega) - 1.
$$
\n(13)

(2) Choosing $\gamma = 0$ in the class $\mathfrak{H}^{\mathfrak{B}}(\gamma, \sigma; x)$ we have the class $\mathfrak{H}^{\mathfrak{B}}(0, \sigma; x) =$ $\mathfrak{H}^{\mathfrak{B}}(\sigma; x) = \mathcal{L}_{\Sigma}(\mathfrak{U}; x)$. The class $\mathcal{L}_{\Sigma}(\mathfrak{U}; x)$ was studied by Murugusundaramoor-thy and Yalçin [\[20\]](#page-238-12). This class consists of the function $\theta \in \mathfrak{B}$ satisfying

$$
\frac{\xi(\theta'(\xi))^{\sigma}}{\theta(\xi)} \prec \mathcal{K}_{\{\mathcal{L}_{U,V,t}(x)\}}(\xi) - 1 \tag{14}
$$

and

$$
\frac{\omega(\vartheta'(w))^{\sigma}}{\vartheta(\omega)} \prec \mathcal{K}_{\{\mathcal{L}_{U,V,t}(x)\}}(\omega) - 1.
$$
\n(15)

(3) Choosing $\sigma = 2$ in the class $\mathfrak{H}^{\mathfrak{B}}(\sigma; x)$ we have the class

$$
\mathfrak{H}^{\mathfrak{B}}(2;x) = \mathfrak{H}^{\mathfrak{B}}(x).
$$

The class consists of the function $f \in \mathfrak{B}$ satisfying

$$
\theta'(\xi) \frac{\xi \theta'(\xi)}{\theta(\xi)} \prec \mathcal{K}_{\{\mathcal{L}_{U,V,t}(x)\}}(\xi) - 1 \tag{16}
$$

and

$$
\vartheta'(\omega) \frac{\omega \vartheta'(w)}{\vartheta(\omega)} \prec \mathcal{K}_{\{\mathcal{L}_{U,V,t}(x)\}}(\omega) - 1.
$$
\n(17)

Remark 3. For $\beta = 1$, we have the new class $\mathfrak{H}^{\mathfrak{B},1}(\sigma; x)$. If $\theta \in \mathfrak{B}$, is in $\mathfrak{H}^{\mathfrak{B},1}(\sigma;x)$, then following condition fulfilles

$$
(\theta'(\xi))^{\sigma} \prec \mathcal{K}_{\{\mathcal{L}_{U,V,t}(x)\}}(\xi) - 1 \tag{18}
$$

and

$$
(\vartheta'(\omega))^{\sigma} \prec \mathcal{K}_{\{\mathcal{L}_{U,V,t}(x)\}}(\omega) - 1.
$$
 (19)

Also,

(1) Choosing $\sigma = 1$ in the class $\mathfrak{H}^{\mathfrak{B},1}(\sigma; x)$ we have the class

 $\mathfrak{H}^{\mathfrak{B},1}(1;x).$

This class consists of the function $\theta \in \mathfrak{B}$ satisfying

$$
\theta'(\xi) \prec \mathcal{K}_{\{\mathcal{L}_{U,V,t}(x)\}}(\xi) - 1 \tag{20}
$$

$$
\vartheta'(\omega) \prec \mathcal{K}_{\{\mathcal{L}_{U,V,t}(x)\}}(\omega) - 1. \tag{21}
$$

2. MAIN THEOREMS FOR THE CLASS $\mathfrak{H}^{\mathfrak{B},\beta}(\gamma,\sigma;x)$

Theorem 1. Let $\theta(\xi) \in \mathfrak{H}^{\mathfrak{B},\beta}(\gamma,\sigma;x)$. Then

$$
|n_2| \leq \frac{|U(x)|\sqrt{|U(x)|}}{\sqrt{\left[U^{2}(x) \left[(\beta - 1) \left[\frac{(\beta - 2)(1 + \gamma)^2}{2} + 2\sigma(1 + \gamma) + (1 + \gamma + \gamma^2) \right] + \sigma(2\sigma + 1) \right] - [2\sigma + (\beta - 1)(1 + \gamma)]^2 \right] - 2V(x)[2\sigma + (\beta - 1)(1 + \gamma)]^2}}
$$
(22)

$$
|n_3| \le \frac{U^2(x)}{[2\sigma + (\beta - 1)(1 + \gamma)]^2} + \frac{|U(x)|}{[3\sigma + (\beta - 1)(1 + \gamma + \gamma^2)]},
$$
(23)

where $\beta \geq 0$, $\sigma \geq 1$ and $|\gamma| \leq 1$ but $\gamma \neq 1$.

Proof. Let $\theta(\xi) \in \mathfrak{H}^{\mathfrak{B},\beta}(\gamma,\sigma;x)$. Then, according to the Definition \mathfrak{I} for some holomorphic functions Φ , Υ such that $\Upsilon(0) = \Phi(0) = 0$, $|\Upsilon(\omega)| < 1$, $|\Phi(\xi)| < 1$, $(\xi, \omega \in \mathfrak{U})$, we can write

$$
\frac{((1-\gamma)\xi)^{1-\beta}(\theta'(\xi))^{\sigma}}{(\theta(\xi)-\theta(\gamma\xi))^{1-\beta}} = \mathcal{K}_{\{\mathcal{L}_{U,V,t}(x)\}}(\Phi(\xi)) - 1
$$

and

$$
\frac{((1-\gamma)\omega)^{1-\beta}(\vartheta'(w))^{\sigma}}{(\vartheta(\omega)-\vartheta(\gamma\omega))^{1-\beta}}=\mathcal{K}_{\{\mathcal{L}_{U,V,t}(x)\}}(\Upsilon(\omega))-1,
$$

by equivalence

$$
\frac{((1-\gamma)\xi)^{1-\beta}(\theta'(\xi))^{\sigma}}{(\theta(\xi)-\theta(\gamma\xi))^{1-\beta}} = -1 + \mathcal{L}_{U,V,0}(x) + \mathcal{L}_{U,V,1}(x)\Phi(\xi) + \mathcal{L}_{U,V,2}(x)\Phi^2(\xi) + \cdots (24)
$$

and

$$
\frac{((1-\gamma)\omega)^{1-\beta}(\vartheta'(w))^{\sigma}}{(\vartheta(\omega)-\vartheta(\gamma\omega))^{1-\beta}} = -1 + \mathcal{L}_{U,V,0}(x) + \mathcal{L}_{U,V,1}(x)\Upsilon(\omega) + \mathcal{L}_{U,V,2}(x)\Upsilon^{2}(\omega) + \cdots
$$
\n(25)

From (24) and (25) , yields

$$
\frac{((1-\gamma)\xi)^{1-\beta}(\theta'(\xi))^{\sigma}}{(\theta(\xi)-\theta(\gamma\xi))^{1-\beta}} = 1 + \mathcal{L}_{U,V,1}(x)y_1\xi + \left[\mathcal{L}_{U,V,1}(x)y_2 + \mathcal{L}_{U,V,2}(x)y_1^2\right]\xi^2 + \cdots (26)
$$

$$
\frac{((1-\gamma)\omega)^{1-\beta}(\vartheta'(w))^{\sigma}}{(\vartheta(\omega)-\vartheta(\gamma\omega))^{1-\beta}} = 1 + \mathcal{L}_{U,V,1}(x)\mu_1\omega + \left[\mathcal{L}_{U,V,1}(x)\mu_2 + \mathcal{L}_{U,V,2}(x)\mu_1^2\right]\omega^2 + \cdots
$$
\n(27)

for $\xi, \omega \in \mathfrak{U}$, it known before that if

$$
|\Phi(\xi)| = \left|\sum_{j=1}^{\infty} y_j \xi^j\right| < 1
$$

and

$$
|\Upsilon(\omega)| = \left| \sum_{j=1}^{\infty} \mu_j \omega^j \right| < 1,
$$
\n
$$
|y_j| < 1
$$
\n
$$
(28)
$$

also

thus

$$
|\mu_j| < 1 \tag{29}
$$

where $j \in \mathfrak{N} = \{1, 2, 3, \cdots\}$. If we compare corresponding coefficients in (26) and (27) , then we have

$$
[2\sigma + (\beta - 1)(1 + \gamma)]n_2 = \mathcal{L}_{U,V,1}(x)y_1,
$$
\n(30)

$$
[3\sigma + (\beta - 1)(1 + \gamma + \gamma^2)]n_3 + \left[\frac{(\beta - 1)(\beta - 2)}{2}(1 + \gamma)^2 + 2\sigma(\sigma - 1 + (\beta - 1)(1 + \gamma)) \right] n_2^2
$$

= $\mathcal{L}_{U,V,1}(x)y_2 + \mathcal{L}_{U,V,2}(x)y_1^2,$ (31)

$$
-[2\sigma + (\beta - 1)(1 + \gamma)]n_2 = \mathcal{L}_{U,V,1}(x)\mu_1,
$$
\n(32)

$$
\left[2(\beta-1)(1+\gamma+\gamma^2)+\frac{(\beta-1)(\beta-2)}{2}(1+\gamma)^2+2\sigma(\sigma+2+(\beta-1)(1+\gamma))\right]n_2^2
$$

$$
-\left[3\sigma+(\beta-1)(1+\gamma+\gamma^2)\right]n_3=\mathcal{L}_{U,V,1}(x)\mu_2+\mathcal{L}_{U,V,2}(x)\mu_1^2. \quad (33)
$$

From (30) and (32)

$$
y_1 = -\mu_1,\tag{34}
$$

$$
2[2\sigma + (\beta - 1)(1 + \gamma)]^2 n_2^2 = \mathcal{L}_{U,V,1}^2(x)(y_1^2 + \mu_1^2).
$$
 (35)

Summation of (31) and (33) gives

$$
\[2(\beta - 1)\left[\frac{(\beta - 2)(1 + \gamma)^2}{2} + 2\sigma(1 + \gamma) + (1 + \gamma + \gamma^2)\right] + 2\sigma(2\sigma + 1)\]n_2^2
$$

= $\mathcal{L}_{U,V,1}(x)(y_2 + \mu_2) + \mathcal{L}_{U,V,2}(x)(y_1^2 + \mu_1^2).$ (36)

Applying [\(35\)](#page-231-4) in [\(36\)](#page-231-5), yields

$$
\left\{ \mathcal{L}_{U,V,1}^{2}(x) \left[2(\beta - 1) \left[\frac{(\beta - 2)(1 + \gamma)^{2}}{2} + 2\sigma(1 + \gamma) + (1 + \gamma + \gamma^{2}) \right] + 2\sigma(2\sigma + 1) \right] \right\}
$$

$$
-2\mathcal{L}_{U,V,2}(x)[2\sigma + (\beta - 1)(1 + \gamma)]^{2}\bigg\}n_{2}^{2} = \mathcal{L}_{U,V,1}^{3}(x)(y_{2} + \mu_{2}), \quad (37)
$$

$$
\left[U^2(x)\left[2(\beta-1)\left[\frac{(\beta-2)(1+\gamma)^2}{2}+2\sigma(1+\gamma)+(1+\gamma+\gamma^2)\right]+2\sigma(2\sigma+1)\right]\right]
$$

$$
-2[2\sigma+(\beta-1)(1+\gamma)]^2\right]-4[2\sigma+(\beta-1)(1+\gamma)]^2V(x)\Bigg]n_2^2=\mathcal{L}_{U,V,1}^3(x)(y_2+\mu_2)
$$

which gives desired result given by (1) .

Hence, (31) minus (33) gives us

$$
2[3\sigma + (\beta - 1)(1 + \gamma + \gamma^2)]n_3 + 2[3\sigma + (\beta - 1)(1 + \gamma + \gamma^2)]n_2^2 = \mathcal{L}_{U,V,1}(x)(y_2 - \mu_2). \tag{38}
$$

Then, by using (34) and (35) in (38) , we get

$$
n_3 = n_2^2 + \frac{\mathcal{L}_{U,V,1}(x)(y_2 - \mu_2)}{2[3\sigma + (\beta - 1)(1 + \gamma + \gamma^2)]}
$$
\n(39)

$$
n_3 = \frac{\mathcal{L}_{U,V,1}^2(x)(y_1^2 + \mu_1^2)}{2[2\sigma + (\beta - 1)(1 + \gamma)]^2} + \frac{\mathcal{L}_{U,V,1}(x)(y_2 - \mu_2)}{2[3\sigma + (\beta - 1)(1 + \gamma + \gamma^2)]}.
$$
(40)

Applying (4) , we have

$$
|n_3| \le \frac{U^2(x)}{[2\sigma + (\beta - 1)(1 + \gamma)]^2} + \frac{|U(x)|}{[3\sigma + (\beta - 1)(1 + \gamma + \gamma^2)]}.
$$

Thus, the proof of our main theorem is completed. \Box

3. Corollaries

By specializing the parameters γ , β , σ , in Theorem [1,](#page-230-5) we get the following consequences.

Corollary 1. Let $\theta(\xi) \in \mathfrak{H}^{\mathfrak{B},\beta}(\gamma,1;x)$. Then

$$
|n_2| \leq \frac{|U(x)|\sqrt{|U(x)|}}{\sqrt{\left|\frac{U^2(x)\left[(\beta-1)\left[\frac{(\beta-2)(1+\gamma)^2}{2}+3(1+\gamma)+\gamma^2\right]+3-[2+(\beta-1)(1+\gamma)]^2\right]}{-2V(x)[2+(\beta-1)(1+\gamma)]^2}\right|}}
$$
(41)

$$
|n_3| \le \frac{U^2(x)}{[2+(\beta-1)(1+\gamma)]^2} + \frac{|U(x)|}{[3+(\beta-1)(1+\gamma+\gamma^2)]}.
$$
 (42)

 $\overline{}$ $\overline{}$ $\overline{}$ $\overline{}$ $\overline{}$ Corollary 2. Let $\theta(\xi) \in \mathfrak{H}^{38,0}(\gamma,\sigma;x) = \mathfrak{H}^{38}(\gamma,\sigma;x)$. Then

$$
|n_2| \le \frac{|U(x)|\sqrt{|U(x)|}}{\sqrt{\left|U^2(x)\left[(2\sigma - 1)(\sigma - \gamma) - [2\sigma - (1 + \gamma)]^2\right] - 2V(x)[2\sigma - (1 + \gamma)]^2\right|}} (43)
$$

$$
|n_3| \le \frac{U^2(x)}{[2\sigma - (1+\gamma)]^2} + \frac{|U(x)|}{|3\sigma - (1+\gamma+\gamma^2)|}.
$$
 (44)

Corollary 3. Let $\theta(\xi) \in \mathfrak{H}^{B,0}(\gamma, 1; x) = \mathfrak{H}^{B}(\gamma, 1; x)$. Then

$$
|n_2| \le \frac{|U(x)|\sqrt{|U(x)|}}{\sqrt{\left|U^2(x)(\gamma - \gamma^2) - 2V(x)(\gamma^2 - 2\gamma + 1)\right|}},\tag{45}
$$
\n
$$
|n_3| \le \frac{U^2(x)}{(1 - \gamma)^2} + \frac{|U(x)|}{|2 - \gamma(\gamma + 1)|}.\tag{46}
$$

Corollary 4. Choosing $\beta = 0$ and $\gamma = 0$ in Theorem $\overline{\beta}$, that is if $\theta(\xi) \in \mathfrak{H}^{\mathfrak{B}}(\sigma; x)$, the results which we obtain reduce to Theorem 2.1 in $[20]$.

$$
|n_2| \le \frac{|U(x)|\sqrt{|U(x)|}}{\sqrt{|U^2(x)(-2\sigma^2 - 1 + 3\sigma) - 2V(x)(2\sigma - 1)^2|}},\tag{47}
$$

$$
|n_3| \le \frac{U^2(x)}{[2\sigma - 1]^2} + \frac{|U(x)|}{|3\sigma - 1|}.\tag{48}
$$

Corollary 5. Choosing $\beta = 0$, $\gamma = 0$ and $\sigma = 2$ in Theorem $\overline{1}$, $\theta(\xi) \in \mathfrak{H}^{\mathfrak{B}}(2; x)$, our corollary coincides with the corollary 2.3 of Theorem 2.1 in $\boxed{20}$.

$$
|n_2| \le \frac{|U(x)|\sqrt{|U(x)|}}{\sqrt{3|U^2(x) + 6V(x)|}},\tag{49}
$$

$$
|n_3| \le \frac{U^2(x)}{9} + \frac{|U(x)|}{5}.\tag{50}
$$

Corollary 6. Choosing $\beta = 0$, $\gamma = 0$ and $\sigma = 1$ in Theorem $\overline{1}$, $\theta(\xi) \in \mathfrak{H}^{\mathfrak{B}}(1; x)$, our corollary coincides with the corollary 2.2 of Theorem 2.1 in $\sqrt{44}$.

$$
|n_2| \le \frac{|U(x)|\sqrt{|U(x)|}}{\sqrt{|U^2(x)|}} = \sqrt{|U(x)|},\tag{51}
$$

$$
|n_3| \le U^2(x) + \frac{|U(x)|}{2}.\tag{52}
$$

Corollary 7. Let $\theta(\xi) \in \mathfrak{H}^{\mathfrak{B},1}(\sigma; x)$. Then

$$
|n_2| \le \frac{|U(x)|\sqrt{|U(x)|}}{\sqrt{\sigma |U^2(x)(1 - 2\sigma) - 8\sigma V(x)|}},
$$
\n(53)

$$
|n_3| \le \frac{U^2(x)}{4\sigma^2} + \frac{|U(x)|}{3\sigma}.
$$
\n(54)

Corollary 8. Let $\theta(\xi) \in \mathfrak{H}^{\mathfrak{B},1}(1;x)$. Then

$$
|n_2| \le \frac{|U(x)|\sqrt{|U(x)|}}{\sqrt{|U^2(x) + 8V(x)|}},\tag{55}
$$

$$
|n_3| \le \frac{U^2(x)}{4} + \frac{|U(x)|}{3}.\tag{56}
$$

Theorem 2. For $\beta \geq 0$, $\sigma \geq 1$, $|\gamma| \leq 1$ but $\gamma \neq 1$, let $\theta \in \mathfrak{A}$ be in the class $\mathfrak{H}^{\mathfrak{B},\beta}(\gamma,\sigma;x)$. Then

$$
\left|n_3-\chi n_2^2\right|\leq \left\{\begin{array}{cc} \frac{|U(x)|}{3\sigma+(\beta-1)(1+\gamma+\gamma^2)}, & |\chi-1|\leq K\\ \\ \frac{|1-\chi|\cdot |U^3(x)|}{|U^2(x)\Delta-2V(x)[2\sigma+(\beta-1)(1+\gamma)]^2|}, & |\chi-1|\geq K. \end{array}\right.
$$

Where

$$
K = \frac{1}{|3\sigma + (\beta - 1)(1 + \gamma + \gamma^{2})|} \left| \Delta - 2[2\sigma + (\beta - 1)(1 + \gamma)]^{2} \frac{V(x)}{U^{2}(x)} \right|
$$

$$
\Delta = (\beta - 1) \left[\frac{(\beta - 2)(1 + \gamma)^2}{2} + 2\sigma(1 + \gamma) + 1 + \gamma + \gamma^2 \right] + \sigma(2\sigma + 1) - [2\sigma + (\beta - 1)(1 + \gamma)]^2.
$$

Proof. From (37) and (38) , we get

$$
n_3 - \chi n_2^2 = \mathcal{L}_{U,V,1}(x) \left[\left(\zeta(\chi;x) + \frac{1}{2[3\sigma + (\beta - 1)(1 + \gamma + \gamma^2)]} \right) y_2 + \left(\zeta(\chi;x) - \frac{1}{2[3\sigma + (\beta - 1)(1 + \gamma + \gamma^2)]} \right) \mu_2 \right]
$$

where

$$
\zeta(\chi; x) = \frac{\mathcal{L}_{U, V, 1}^2(x)(1 - \chi)}{\mathcal{L}_{U, V, 1}^2(x) \left[2(\beta - 1)\left[\frac{(\beta - 2)(1 + \gamma)^2}{2} + 2\sigma\right]\right]}
$$

 $(1+\gamma) + (1+\gamma+\gamma^2) + 2\sigma(2\sigma+1) - 2\mathcal{L}_{U,V,2}(x)[2\sigma + (\beta-1)(1+\gamma)]^2$. Thus, according to (4) , we have

$$
\left|n_3 - \chi n_2^2\right| \le \begin{cases} \frac{|U(x)|}{3\sigma + (\beta - 1)(1 + \gamma + \gamma^2)}, & 0 \le |\zeta(\chi; x)| \le \frac{1}{2[3\sigma + (\beta - 1)(1 + \gamma + \gamma^2)]} \\ 2|\zeta(\chi; x)| \cdot |U(x)|, & |\zeta(\chi; x)| \ge \frac{1}{2[3\sigma + (\beta - 1)(1 + \gamma + \gamma^2)]} \end{cases}
$$

hence, after some calculations, gives

$$
|n_3 - \chi n_2^2| \le \begin{cases} \frac{|U(x)|}{3\sigma + (\beta - 1)(1 + \gamma + \gamma^2)}, & |x - 1| \le K\\ \frac{|1 - \chi| \cdot |U^3(x)|}{|U^2(x)\Delta - 2V(x)[2\sigma + (\beta - 1)(1 + \gamma)]^2|}, & |x - 1| \ge K. \end{cases}
$$

By choosing special values for the parameters γ , β , σ , in Theorem [2,](#page-234-0) we get the following corollaries:

Corollary 9. For $\sigma = 1$, let $\theta \in \mathfrak{H}^{\mathfrak{B},\beta}(\gamma,1;x)$. Then

$$
\left|n_3 - \chi n_2^2\right| \le \begin{cases} \frac{|U(x)|}{3 + (\beta - 1)(1 + \gamma + \gamma^2)}, & |\chi - 1| \le K_1\\ \frac{|1 - \chi| \cdot |U^3(x)|}{|U^2(x)\Delta_1 - 2V(x)[2 + (\beta - 1)(1 + \gamma)]^2|}, & |\chi - 1| \ge K_1. \end{cases}
$$

Where

$$
K_1 = \frac{1}{|3 + (\beta - 1)(1 + \gamma + \gamma^2)|} \left| \Delta_1 - 2[2 + (\beta - 1)(1 + \gamma)]^2 \frac{V(x)}{U^2(x)} \right|
$$

$$
\Delta_1 = (\beta - 1) \left[\frac{(\beta - 2)(1 + \gamma)^2}{2} + \gamma^2 + 3\gamma + 3 \right] + 3 - [2 + (\beta - 1)(1 + \gamma)]^2.
$$

Corollary 10. For $\beta = 0$, let $\theta \in \mathfrak{H}^{\mathfrak{B},0}(\gamma,\sigma;x)$. Then

$$
\left|n_3 - \chi n_2^2\right| \le \begin{cases} \frac{|U(x)|}{3\sigma - (1+\gamma+\gamma^2)}, & |\chi - 1| \le K_2\\ \frac{|1-\chi| \cdot |U^3(x)|}{|U^2(x)\Delta_2 - 2V(x)[2\sigma - (1+\gamma)]^2|}, & |\chi - 1| \ge K_2. \end{cases}
$$

Where

$$
K_2 = \frac{1}{|3\sigma - (1 + \gamma + \gamma^2)|} \left| \Delta_2 - 2[2\sigma - (1 + \gamma)]^2 \frac{V(x)}{U^2(x)} \right|
$$

$$
\Delta_2 = (2\sigma - 1)(\sigma - \gamma) - [2\sigma - (1 + \gamma)]^2.
$$

Corollary 11. For $\sigma = 1$, $\beta = 0$, let $\theta \in \mathfrak{H}^{38,0}(\gamma, 1; x)$. Then

$$
\left|n_3 - \chi n_2^2\right| \le \begin{cases} \frac{|U(x)|}{|2 - \gamma(\gamma + 1)|}, & |\chi - 1| \le K_3\\ \frac{|1 - \chi| \cdot |U^3(x)|}{|U^2(x)\Delta_3 - 2V(x)|1 - \gamma|^2|}, & |\chi - 1| \ge K_3. \end{cases}
$$

Where

$$
K_3 = \frac{1}{|2 - \gamma(\gamma + 1)|} \left| \Delta_3 - 2[1 - \gamma]^2 \frac{V(x)}{U^2(x)} \right|
$$

$$
\Delta_3=\gamma(1-\gamma).
$$

Corollary 12. For $\beta = 0$, $\gamma = 0$, let $\theta \in \mathfrak{H}^{\mathfrak{B},0}(0,\sigma;x)$. Then

$$
|n_3 - \chi n_2^2| \le \begin{cases} \frac{|U(x)|}{|3\sigma - 1|}, & |\chi - 1| \le K_4\\ \frac{|1 - \chi| \cdot |U^3(x)|}{|U^2(x)\Delta_4 - 2V(x)(2\sigma - 1)^2|}, & |\chi - 1| \ge K_4. \end{cases}
$$

Where

$$
K_4 = \frac{1}{|3\sigma - 1|} \left| \Delta_4 - 2[2\sigma - 1]^2 \frac{V(x)}{U^2(x)} \right|.
$$

$$
\Delta_4 = (2\sigma - 1)(1 - \sigma)
$$

Corollary 13. For $\sigma = 2$, let $\theta \in \mathfrak{H}^{\mathfrak{B},0}(0,2;x)$. Then

$$
|n_3 - \chi n_2^2| \le \begin{cases} \frac{|U(x)|}{5}, & \left(|\chi - 1| \le \frac{3}{5} \middle| 1 + 6\frac{V(x)}{U^2(x)} \middle| \right) \\ & \\ \frac{|1 - \chi| \cdot |U^3(x)|}{3|U^2(x) + 6V(x)|}, & \left(|\chi - 1| \ge \frac{3}{5} \middle| 1 + 6\frac{V(x)}{U^2(x)} \middle| \right). \end{cases}
$$

Corollary 14. $\overline{[44]}$ $\overline{[44]}$ $\overline{[44]}$ For $\sigma \geq 1$, let $\theta \in \mathfrak{A}$ be in the class $\mathfrak{H}^{\mathfrak{B},0}(0,1;x) = \mathfrak{H}^{\mathfrak{B}}(x)$. Then

$$
|n_3 - \chi n_2^2| \le \begin{cases} \frac{|U(x)|}{2}, & \left(|\chi - 1| \le \frac{|V(x)|}{|U^2 x|}\right) \\ & \frac{|1 - \chi| \cdot |U^3(x)|}{2|V(x)|}, & \left(|\chi - 1| \ge \frac{|V(x)|}{|U^2 x|}\right). \end{cases}
$$

Corollary 15. For $\beta = 1$, let $\theta \in \mathfrak{H}^{\mathfrak{B},1}(\sigma; x)$. Then

$$
\left|n_3 - \chi n_2^2\right| \le \begin{cases} \frac{|U(x)|}{3\sigma}, & \left(|\chi - 1| \le K_5\right) \\ & \\ \frac{|1 - \chi| \cdot |U^3(x)|}{|U^2(x)\Delta_5 - 8\sigma^2 V(x)|}, & \left(|\chi - 1| \ge K_5\right). \end{cases}
$$

Where

$$
K_5 = \frac{1}{|3\sigma|} \left| \Delta_5 - 8\sigma^2 \frac{V(x)}{U^2(x)} \right|
$$

$$
\Delta_5 = \sigma(1 - 2\sigma)
$$

Corollary 16. $\sqrt{8}$ For $\sigma = 1$, $\beta = 1$, let $\theta \in \mathfrak{H}^{\mathfrak{B},1}(1;x)$. Then

$$
\left|n_3 - \chi n_2^2\right| \le \begin{cases} \frac{|U(x)|}{3}, & \left(|\chi - 1| \le \frac{1}{3} \left|1 + 8\frac{V(x)}{U^2(x)}\right|\right) \\ & \\ \frac{|1 - \chi| \cdot |U^3(x)|}{|U^2(x) + 8V(x)|}, & \left(|\chi - 1| \ge \frac{1}{3} \left|1 + 8\frac{V(x)}{U^2(x)}\right|\right). \end{cases}
$$

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APPROXIMATION PROPERTIES OF THE FRACTIONAL q-INTEGRAL OF RIEMANN-LIOUVILLE INTEGRAL TYPE SZÁSZ-MIRAKYAN-KANTOROVICH OPERATORS

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ABSTRACT. In the present paper, we introduce the fractional q -integral of Riemann-Liouville integral type Szász-Mirakyan-Kantorovich operators. Korovkin-type approximation theorem is given and the order of convergence of these operators are obtained by using Lipschitz-type maximal functions, second order modulus of smoothness and Peetre's K-functional. Weighted approximation properties of these operators in terms of modulus of continuity have been investigated. Then, for these operators, we give a Voronovskaya-type theorem. Moreover, bivariate fractional q - integral Riemann-Liouville fractional integral type Szász-Mirakyan-Kantorovich operators are constructed.The last section is devoted to detailed graphical representation and error estimation results for these operators.

1. INTRODUCTION

Approximation theory is a subject that serves as an important bridge between applied and pure mathematics. The approximation of functions by positive linear operators is an important research area in general mathematics. Especially, it plays an important role in mathematical analysis problems and in many fields of science. One of its most important advantages is that it provides powerful tools for application areas such as computer aided geometric design and numerical analysis. One of the best known of these operators is the Sz \acute{a} sz - Mirakyan operator (see $\vert \Omega \vert$) and [\[10\]](#page-271-2)), which is generalizations of Bernstein polynomials to the infinite interval and defined as

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$$
S_n(f;x) = \sum_{k=0}^n s_{n,k}(x) f\left(\frac{k}{n}\right),\,
$$

where $n \in \mathbb{N}$, $x \in [0, \infty)$ and $s_{n,k}(x) = e^{-nx} \frac{(nx)^k}{k!}$ $\frac{f(x)}{k!}$. In literature, there are a lot of studies that involve Szàsz operators, Szàsz-Kantorovich operators and their generalizations. For instance, see $\boxed{1}$ - $\boxed{8}$ and $\boxed{14}$ - $\boxed{22}$. Due to the rapid development

of the q-calculus, various generalizations of Szàsz Mirakyan operators involving q integers have been introduced and approximation properties have been studied. Several researchers introduced and studied different generalizations of the q -Szász-Mirakjan operators in recent years $([28], [29], [19], [30], [41])$ $([28], [29], [19], [30], [41])$ $([28], [29], [19], [30], [41])$ $([28], [29], [19], [30], [41])$ $([28], [29], [19], [30], [41])$ $([28], [29], [19], [30], [41])$ $([28], [29], [19], [30], [41])$ $([28], [29], [19], [30], [41])$ $([28], [29], [19], [30], [41])$. In $[28],$ Mahmudov introduced and studied the following q -Szász-Mirakjan operators.

$$
S_{n,q}(f;x) = \sum_{k=0}^{n} s_{n,k}(q;x) f\left(\frac{[k]_q}{q^{k-2} [n]_q}\right),
$$

where $s_{n,k}(q;x) = \frac{1}{E_q([n]_q x)} q^{\frac{k(k-1)}{2} \frac{[n]_q^k x^k}{[k]_q!}}$.

About contributions on Kantorovich type modication modified many times q -Szász-Mirakjan operators, so we refer to the papers $[31]$ - $[34]$. Recently, Fractional calculus and its applications have been paid more and more attention. fractional calculus deals with the study of fractional degree derivative and integral operators on complex or real fields and their applications (see $\boxed{23}$ - $\boxed{27}$). Mahmudov and Kara, introduced and discussed the fractional integral of Riemann-Liouville integral type Szász Mirakyan-Kantorovich operators as follows:

$$
K_n^{(\alpha)}(f;x) = \sum_{k=0}^{\infty} \alpha s_{n,k}(x) \int_0^1 \frac{f\left(\frac{k+t}{n}\right)}{(1-t)^{1-\alpha}} dt,
$$
\n(1)

where $s_{n,k}(x) = e^{-nx} \frac{(nx)^k}{k!}$ $\frac{f(x)}{k!}$. The aim of the present paper is to construct the frac-

tional q-integral of Riemann-Liouville type Szász-Mirakyan-Kantorovich operators and discuss their approximation properties. The fractional q -integral of Riemann-Liouville type ($\boxed{35}$) is given by $\left(I_q^0 f\right)(t) = f(t)$ and

$$
(I_q^{\alpha}f)(x)=\frac{1}{\Gamma_q(\alpha)}\int\limits_0^x\frac{f\left(t\right)}{\left(x-qt\right)^{\left(1-\alpha\right)}}d_qt\qquad \left(\ \alpha>0\right).
$$

We start by reminding the basic concepts and notations about fractional q -calculus.

2. Preliminaries

For $q \in (0, 1)$,

$$
[m]_q = \frac{1 - q^m}{1 - q}, \quad m \in \mathbb{R}
$$

The q-analog of the power function $(n - m)^{(k)}$ with $k \in \mathbb{N}_0 := \{0, 1, 2, ...\}$ is

$$
(n-m)^{(0)} = 1
$$
, $(n-m)^{(k)} = \prod_{i=0}^{k-1} (n-mq^i)$, $k \in \mathbb{N}, n, m \in \mathbb{R}$.

More generally, if $\gamma \in \mathbb{R}$, then

$$
(n-m)^{(\gamma)}=\prod_{i=0}^{\infty}\frac{n-mq^i}{n-mq^{\gamma+i}},\ \ n\neq 0.
$$

Note if $m = 0$, then $(n)^{(\gamma)} = n^{\gamma}$. We also use the natation $0^{(\gamma)} = 0$ for $\gamma > 0$. The q-gamma function is defined by

$$
\Gamma_q(t) = \frac{(1-q)^{(t-1)}}{(1-q)^{t-1}}, \quad t \in \mathbb{R} \setminus \{0, -1, -2, \dots\}.
$$

Obviously, $\Gamma_q(t+1) = [t]_q \Gamma_q(t)$.

For any $s, t > 0$, the q-beta function is defined by

$$
B_q(s,t) = \int_0^1 u^{(s-1)} (1 - qu)^{(t-1)} d_q u.
$$

The q -beta function can be expressed by using the q -gamma function as follows:

$$
B_q(s,t) = \frac{\Gamma_q(s)\Gamma_q(t)}{\Gamma_q(s+t)}.
$$

The q-integral definition of the function h on the interval $[0, b]$ is given as:

$$
(I_qh)(t) = \int_0^t h(s)d_q s = t(1-q)\sum_{i=0}^\infty h(tq^i)q^i, \quad t \in [0, b].
$$

In q-calculus (see $\boxed{36}$) the following functions are well known as analogues of the exponential function:

$$
e_q(x)=\sum_{k=0}^\infty \frac{x^k}{[k]_q!},\qquad |x|<\frac{1}{1-q},\quad |q|<1,
$$

$$
E_q(x) = \sum_{k=0}^{\infty} \frac{x^k}{[k]_q!} q^{\frac{k(k-1)}{2}}, \quad |q| < 1.
$$

$$
S_{n,q}(t^4;x) = \frac{x^4}{q^2} + \left(3q+2+\frac{1}{q}\right)\frac{x^3}{[n]_q} + \left(3q^3+3q^2+q\right)\frac{x^2}{[n]_q^2} + \frac{q^4x}{[n]_q^3}.
$$

Definition 1. Let $q \in (0,1)$ and $\alpha > 0$. For $f \in C[0,\infty)$, Fractional q-integral of Riemann-Liouville type Szász-Mirakyan-Kantorovich operators can be defined by

$$
K_{n,q}^{(\alpha)}(f;x) = \sum_{k=0}^{\infty} [\alpha]_q s_{n,k}(q;x) \int_0^1 f\left(\frac{q^{1-k} [k]_q + t}{[n]_q}\right) (1 - qt)^{(\alpha - 1)} d_q t,
$$
 (2)

where $s_{n,k}(q;x) = \frac{1}{E_q([n]_q x)} q^{\frac{k(k-1)}{2}} \frac{[n]_q^k x^k}{[k]_q!}$ $\frac{[n]_q^nx^n}{[k]_q!}$ and $n \in \mathbb{N}$

if $\alpha = 1$ and $q = 1$, then the operator [\(2\)](#page-244-0) reduces to classical Szász-Mirakyan -Kantorovich operators.

Due to the moments of the $K_{n,q}^{(\alpha)}$ operators plays significant role in our main results, we derive the following formula to obtain them.

Lemma 2. Let $q \in (0,1)$ and $\alpha > 0$. Then for $x \in [0,\infty)$, we have

$$
K_{n,q}^{(\alpha)}(t^m;x) = \sum_{j=0}^m {m \choose j} \frac{[\alpha]_q [n]_q^j B_q(m-j+1,\alpha)}{q^j [n]_q^m} S_{n,q}(t^j;x),
$$
(3)

where

$$
S_{n,q}(f;x) = \sum_{k=0}^{n} s_{n,k}(q;x) f\left(\frac{[k]_q}{q^{(k-2)}[n]_q}\right)
$$

and

$$
B_q(a,b) = \int_0^1 x^{a-1} (1-qx)^{b-1} d_qx, \quad a, b > 0.
$$

Proof. From (2) , we can write

$$
K_{n,q}^{(\alpha)}(t^m; x) = \sum_{k=0}^{\infty} [\alpha]_q s_{n,k}(q; x) \int_0^1 \left(\frac{q^{1-k} [k]_q + t}{[n]_q}\right)^m (1 - qt)^{(\alpha - 1)} d_q t
$$

$$
= \sum_{k=0}^{\infty} [\alpha]_q s_{n,k}(q; x) \sum_{j=0}^m {m \choose j} \frac{q^{(1-k)j} [k]_q^j}{[n]_q^m} \int_0^1 t^{(m-j)} (1 - qt)^{(\alpha - 1)} d_q t
$$

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$$
= \sum_{j=0}^{m} {m \choose j} \frac{\left[\alpha\right]_q \left[n\right]_q^j B_q(m-j+1, \alpha)}{q^j \left[n\right]_q^m} \sum_{k=0}^{\infty} s_{n,k}(q;x) \frac{\left[k\right]_q^j}{q^{(k-2)j} \left[n\right]_q^j}
$$

$$
= \sum_{j=0}^{m} {m \choose j} \frac{\left[\alpha\right]_q \left[n\right]_q^j B_q(m-j+1, \alpha)}{q^j \left[n\right]_q^m} S_{n,q}(t^j;x).
$$

□

For $j = 0, 1, 2, 3, 4$ $(K_{n,q}^{\alpha}(t^j; x))$, the following can be written immediately. **Lemma 3.** Let $q \in (0,1), \alpha > 0$ and $n \in \mathbb{N}$. Then for $x \in [0,\infty)$, we have

(i)
$$
K_{n,q}^{(\alpha)}(1;x) = 1
$$
,
\n(ii) $K_{n,q}^{(\alpha)}(t;x) = x + \frac{1}{[n]_q [\alpha + 1]_q}$
\n(iii) $K_{n,q}^{(\alpha)}(t^2;x) = \frac{[2]_q}{[\alpha + 1]_q [\alpha + 2]_q [n]_q^2} + \frac{(2 + [\alpha + 1]_q)}{[\alpha + 1]_q [n]_q} x + \frac{x^2}{q}$,
\n(iv) $K_{n,q}^{(\alpha)}(t^3;x) = \frac{[3]_q [2]_q}{[\alpha + 1]_q [\alpha + 2]_q [\alpha + 3]_q [n]_q^3} + \left(\frac{3 \cdot [2]_q}{[\alpha + 1]_q [\alpha + 2]_q [n]_q^2} + \frac{3}{[\alpha + 1]_q [n]_q^2} + 1\right) x + \left(\frac{3}{q [n]_q [\alpha + 1]_q} + \frac{2q^2 + q}{q^3 [n]_q}\right) x^2 + \frac{x^3}{q^3}$,

$$
\begin{split} (v)K_{n,q}^{(\alpha)}(t^4;x)&=\frac{[4]_q!}{[\alpha+1]_q\,[\alpha+2]_q\,[\alpha+3]_q\,[\alpha+4]_q\,[n]_q^4}\\ &+\left(\frac{4\,[3]_q!+6\,[\alpha+3]_q\,[2]_q!+[\alpha+2]_q\,[\alpha+3]_q\left\{4+[\alpha+1]_q\right\}}{[\alpha+1]_q\,[\alpha+2]_q\,[\alpha+3]_q\,[n]_q^3}\right)x\\ &+\left(\frac{6\,[2]_q!}{q\,[\alpha+1]_q\,[\alpha+2]_q\,[n]_q^2}+\frac{4\,\left(2q^2+q\right)}{q^3\,[\alpha+1]_q\,[n]_q^2}+\frac{\left(3q^3+3q^2+q\right)}{q^4\,[n]_q^2}\right)x^2\\ &+\left(\frac{4}{q^3\,[n]_q\,[\alpha+1]_q}+\frac{3q+2+\frac{1}{q}}{q^4\,[n]_q}\right)x^3+\frac{x^4}{q^6}. \end{split}
$$

Proof. Since they have the same proof technique, we only give for $K_{n,q}^{(\alpha)}(t^2; x)$. Using recurrence formula (3) and Lemma $\overline{1}$, we get

$$
K_{n,q}^{(\alpha)}(t^2;x)=\frac{[\alpha]_q\,B_q(3,\alpha)}{[n]_q^2}S_{n,q}(1;x)+\frac{2\,[n]_q\,[\alpha]_q\,B_q(2,\alpha)}{q\,[n]_q^2}S_{n,q}(t;x)
$$

$$
+\frac{\left[n\right]_{q}^{2}\left[\alpha\right]_{q}B_{q}(1,\alpha)}{q^{2}\left[n\right]_{q}^{2}}S_{n,q}(t^{2};x)
$$
\n
$$
=\frac{\left[2\right]_{q}}{\left[\alpha+1\right]_{q}\left[\alpha+2\right]_{q}\left[n\right]_{q}^{2}}+\frac{2}{\left[\alpha+1\right]_{q}\left[n\right]_{q}}x+\left(\frac{x^{2}}{q}+\frac{x}{\left[n\right]_{q}}\right)
$$
\n
$$
=\frac{\left[2\right]_{q}}{\left[\alpha+1\right]_{q}\left[\alpha+2\right]_{q}\left[n\right]_{q}^{2}}+\frac{\left(2+\left[\alpha+1\right]_{q}\right)}{\left[\alpha+1\right]_{q}\left[n\right]_{q}}x+\frac{x^{2}}{q}.
$$

We are now ready to present the central moments of the operators $K_{n,q}^{(\alpha)}$. **Lemma 4.** Let $q \in (0,1)$ and $\alpha > 0$. For every $x \in [0,\infty)$, there holds

$$
K_{n,q}^{(\alpha)}(t-x;x) = \frac{1}{[n]_q [\alpha + 1]_q},
$$
\n
$$
K_{n,q}^{(\alpha)}((t-x)^2;x) = \frac{[2]_q}{[\alpha + 1]_q [\alpha + 2]_q [n]_q^2} + \frac{x}{[n]_q} + x^2 \left(\frac{1}{q} - 1\right),
$$
\n
$$
K_{n,q}^{(\alpha)}((t-x)^4;x)
$$
\n
$$
= \frac{[4]_q!}{[\alpha + 1]_q [\alpha + 2]_q [\alpha + 3]_q [\alpha + 4]_q [n]_q^4}
$$
\n
$$
+ \left(\frac{6 [2]_q}{[\alpha + 1]_q [\alpha + 2]_q [n]_q^3} + \frac{4}{[\alpha + 1]_q [n]_q^3} + \frac{1}{[n]_q^3}\right)x
$$
\n
$$
+ \left(\frac{\frac{6 [2]_q}{q [\alpha + 1]_q [\alpha + 2]_q [n]_q^2} + \frac{4(2q^2 + q)}{[\alpha + 1]_q q^3 [n]_q^2}}{q^4 [n]_q^2} - \frac{6 [2]_q - 12[\alpha + 2]_q}{[\alpha + 1]_q [\alpha + 2]_q [n]_q^2} - 4}\right)x^2
$$
\n
$$
+ \left(\frac{\frac{4}{q^3 [\alpha + 1]_q [n]_q}}{q^3 [n]_q} + \frac{3q + 2 + \frac{1}{q}}{q^4 [n]_q} - \frac{12}{q[\alpha + 1]_q [n]_q}}{q[\alpha + 1]_q [n]_q}\right)x^3
$$
\n
$$
+ \left(\frac{1}{q^6} - \frac{4}{q^3} + \frac{6}{q} - 3\right)x^4.
$$

Proof. Since they have the same proof technique, we only give for $K_{n,q}^{(\alpha)}((t-x)^2; x)$. From the linearity property of $K_{n,q}^{(\alpha)}(t;x)$ and Lemma $\overline{3}$, we get

$$
K_{n,q}^{(\alpha)}((t-x)^2;x)
$$

=
$$
\frac{[2]_q}{[\alpha+1]_q [\alpha+2]_q [n]_q^2} + \frac{\left(2+[\alpha+1]_q\right)}{[\alpha+1]_q [n]_q}x + \frac{x^2}{q} - 2x\left(x + \frac{1}{[n]_q [\alpha+1]_q}\right) + x^2.
$$

□

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Lemma 5. Assume that the sequence (q_n) satisfy $0 < q_n \leq 1$ such that $q_n \to 1$ and $q_n^n \to b \in [0,1]$ as $n \to \infty$. For every $\alpha > 0$ and $x \in [0,\infty)$, there holds

$$
\lim_{n \to \infty} [n]_{q_n} K_{n,k}^{(\alpha)}((t-x);x) = \frac{1}{\alpha+1},
$$
\n(4)

$$
\lim_{n \to \infty} [n]_{q_n} K_{n,k}^{(\alpha)}((t-x)^2; x) = x + x^2 (1-b), \tag{5}
$$

and

$$
\lim_{n \to \infty} [n]_{q_n} K_{n,k}^{(\alpha)}((t-x)^4; x) = 0.
$$
\n(6)

Proof. Using explicit formula for moments (Lemma \overline{A}), we obtain as

$$
\lim_{n \to \infty} [n]_{q_n} K_{n,k}^{(\alpha)}((t-x);x) = \frac{1}{(\alpha+1)},
$$

$$
\lim_{n \to \infty} [n]_{q_n} K_{n,k}^{(\alpha)}((t-x)^2; x)
$$
\n
$$
= \lim_{n \to \infty} [n]_{q_n} \left(\frac{[2]_{q_n}}{[\alpha+1]_{q_n} [\alpha+2]_{q_n} [n]_{q_n}^2} + \frac{x}{[n]_{q_n}} + x^2 \left(\frac{1}{q_n} - 1 \right) \right)
$$
\n
$$
= x + x^2 (1 - b)
$$

and

$$
\lim_{n \to \infty} [n]_{q_n} K_{n,k}^{(\alpha)}((t-x)^4; x) = 0.
$$

In [\[28\]](#page-272-2), Mahmudov gave the following formula for the moments of $S_{n,q}(t^m; x)$, which is a q-analogue of result of Beker $\boxed{37}$.

Lemma 6. $[29]$ For $0 < q < 1$ and $m \in \mathbb{N}$, there holds

$$
S_{n,q}(t^m; x) = \sum_{j=1}^{m} a_{m,j}(q) \frac{x^j}{[n]_q^{m-j}}
$$
(7)

where

$$
a_{m+1,j}(q) = \frac{[j]_q a_{m,j}(q) + a_{m,j-1}(q)}{q^{j-2}}, \qquad m \ge 0, j \ge 1,
$$

$$
a_{0,0}(q) = 1, a_{m,0}(q) = 0, \quad m > 0, \quad a_{m,j}(q) = 0, \qquad m < j.
$$

In particular $S_{n,q}(t^m;x)$ is a polynomial of degree m without a constant term.

Now we additionally need to give the following definitions for our main results:

- 1. $B_m [0, \infty) = \{ f : [0, \infty) \to \mathbb{R}; |f(x)| \le M_f (1 + x^m) \}$, where M_f is constant depending on the function f .
- 2. $C_m(0,\infty) = B_m(0,\infty) \cap C(0,\infty)$.

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3.
$$
C_m^* [0, \infty) = \left\{ f : C_m [0, \infty) : \lim_{|x| \to \infty} \frac{|f(x)|}{1 + x^m} < \infty \right\}
$$

The norm on the space C_m^* [0, ∞) is showed as $||f(x)||_m = \sup_{x \in [0,\infty)}$ $\frac{|f(x)|}{1+x^m}$.

Lemma 7. Let $m \in \mathbb{N} \cup \{0\}$, $0 < q < 1$ and $\alpha > 0$ be fixed. Then, we have

$$
\left\|K_{n,q}^{(\alpha)}(1+x^m;x)\right\|_m \le C_{m,j}(q,\alpha), \ n \in \mathbb{N},\tag{8}
$$

where $C_{m,j}(q,\alpha)$ is a positive constant. Morever, we have

$$
\left\|K_{n,q}^{(\alpha)}(f;x)\right\|_{m} \le C_{m,j}(q,\alpha) \left\|f\right\|_{m}, \quad n \in \mathbb{N},\tag{9}
$$

where $f \in C_m^*$ $[0, \infty)$. Thus, for any $m \in \mathbb{N} \cup \{0\}$, $K_{n,q}^{(\alpha)} : C_m^*$ $[0, \infty) \to C_m^*$ $[0, \infty)$ is a linear positive operator.

Proof. For $m = 0$, inequality $\boxed{8}$ is obvious.

For $m \geq 1$, combining Lemma $\boxed{3}$ and inequality $\boxed{7}$, we obtain as

$$
\frac{1}{x^m+1} K_{n,q}^{(\alpha)}(1+t^m; x)
$$
\n
$$
= \frac{1}{x^m+1} + \frac{1}{x^m+1} K_{n,q}^{(\alpha)}(t^m; x)
$$
\n
$$
\frac{1}{x^m+1} + \frac{1}{x^m+1} \sum_{j=0}^m \frac{[\alpha]_q [n]_q^j B_q(m-j+1, \alpha)}{q^j [n]_q^m} \sum_{j_0=1}^j a_{j,j_0}(q) \frac{x^{j_0}}{n^{j-j_0}}
$$
\n
$$
\leq 1 + k_{m,j}(q, \alpha) = C_{m,j}(q, \alpha).
$$

 $C_{m,j}(q,\alpha)$ is a positive constant with depend on q, m, j and α . Moreover,

$$
\left\|K_{n,q}^{(\alpha)}(f;x)\right\|_{m} \le \|f\|_{m} \left\|K_{n,q}^{(\alpha)}(1+t^{m};x)\right\|_{m}
$$
\n(10)

for every $f \in C_m^* [0, \infty)$. Therefore, from $\boxed{8}$, we get

$$
\left\|K_{n,q}^{(\alpha)}(f;x)\right\|_{m} \leq C_{m,j}(q,\alpha) \left\|f\right\|_{m}.
$$

4. Direct Results

Let C_B [0, ∞) denote the space of all real-valued continuous and bounded functions f on $[0, \infty)$. The norm on the space $C_B[0, \infty)$ is showed as

$$
||f||_{C_B[0,\infty)} = \sup_{x \in [0,\infty)} |f(x)|.
$$

Then, the modulus of continuity of $f \in C_B[0,\infty)$ is given by

$$
w(f,\delta) = \sup_{0
$$

Further, Peetre's K-functional is defined by

$$
K_2(f; \delta) = \inf_{g \in \omega^2} \left\{ \|f - g\| + \delta \left\| g'' \right\| \right\} \quad \delta > 0,
$$

where $w^2 := \left\{ g \in C_B [0, \infty) : g', g'' \in C_B [0, \infty) \right\}$. By Theorem 2.4 in [\[11\]](#page-271-5), there exists an absolute constant $L > 0$ such that

$$
K_2(f; \delta) \le L\omega_2\left(f; \sqrt{\delta}\right). \tag{11}
$$

where $\delta > 0$ are absolute constant.

Here, $\omega_2(f; \delta)$ is the second order modulus of smoothness of $f \in C_B[0, \infty)$ and defined as

$$
\omega_2(f; \delta) = \sup_{0 < h \le \delta} \sup_{x \in [0, \infty)} |f(x + 2h) - 2f(x + h) + f(x)|
$$

Lemma 8. Let $f \in C_B[0,\infty)$, $0 < q < 1$ and $\alpha > 0$. Consider the operators

$$
{}^{*}K_{n,q}^{(\alpha)}(f;x) = K_{n,q}^{(\alpha)}(f;x) + f(x) - f\left(x + \frac{1}{[n]_q [\alpha + 1]_q}\right). \tag{12}
$$

Then, for all $g \in w^2$, we have

$$
\begin{aligned}\n\left| \n\begin{array}{l}\n\ast K_{n,q}^{(\alpha)}(g;x) - g(x) \\
\leq \left(\frac{[2]_q}{[\alpha+1]_q [\alpha+2]_q [n]_q^2} + \frac{x}{[n]} + x^2 \left(\frac{1}{q} - 1 \right) + \left(\frac{1}{[n]_q [\alpha+1]_q} \right)^2 \right) \|g''\|.\n\end{array}\n\right\end{aligned} \tag{13}
$$

Proof. From (12) we have

$$
{}^{*}K_{n,q}^{(\alpha)}((t-x);x) = K_{n,q}^{(\alpha)}((t-x);x) - \left(x + \frac{1}{[n]_{q} [\alpha + 1]_{q}} - x\right)
$$

$$
= K_{n,q}^{(\alpha)}(t;x) - xK_{n,q}^{(\alpha)}(1;x) - \left(x + \frac{1}{[n]_{q} [\alpha + 1]_{q}}\right) + x = 0.
$$
 (14)

Let $x \in [0, \infty)$ and $g \in w^2$. Using the Taylor's formula,

$$
g(t) - g(x) = (t - x)g'(x) + \int_{x}^{t} (t - u)g''(u)du,
$$
\n(15)

Applying $K_{n,q}^{(\alpha)}$ and using (14) , we can get

$$
{}^{*}K_{n,q}^{(\alpha)}(g;x) - g(x) = {}^{*}K_{n,q}^{(\alpha)}((t-x)g^{'}(x);x) + {}^{*}K_{n,q}^{(\alpha)}\left(\int_{x}^{t} (t-u)g^{''}(u)du;x\right)
$$

$$
= g'(x)^* K_{n,q}^{(\alpha)}((t-x);x) + K_{n,q}^{(\alpha)} \left(\int_x^t (t-u)g^{''}(u) du; x \right)
$$

$$
- \int_x^{x + \frac{1}{\lfloor n \rfloor_q \lfloor \alpha + 1 \rfloor_q}} \left(x + \frac{1}{\lfloor n \rfloor_q \lfloor \alpha + 1 \rfloor_q} - u \right) g^{''}(u) du
$$

$$
= K_{n,q}^{(\alpha)} \left(\int_x^t (t-u)g^{''}(u) du; x \right)
$$

$$
- \int_x^{x + \frac{1}{\lfloor n \rfloor_q \lfloor \alpha + 1 \rfloor_q}} \left(x + \frac{1}{\lfloor n \rfloor_q \lfloor \alpha + 1 \rfloor_q} - u \right) g^{''}(u) du.
$$

On the other hand, since

$$
\int_{x}^{t} |t - u| \left| g''(u) \right| du \le \left\| g'' \right\| \int_{x}^{t} |t - u| du \le (t - x)^2 \left\| g'' \right\|
$$

and

$$
\begin{aligned}\n\left| \int_{x}^{x + \frac{1}{\lfloor n \rfloor_q (\alpha + 1)_q}} \left(x + \frac{1}{\lfloor n \rfloor_q [\alpha + 1]_q} - u \right) g''(u) du \right| \\
&\leq \left(\frac{1}{\lfloor n \rfloor_q [\alpha + 1]_q} \right)^2 \left\| g'' \right\|, \n\end{aligned}
$$

we conclude that

$$
\begin{aligned}\n&\left|{^*K}_{n,q}^{(\alpha)}(g;x) - g(x)\right| \\
&= \left| K_{n,q}^{(\alpha)}\left(\int\limits_x^t (t-u)g^{''}(u)du;x\right) - \int\limits_x^{x + \frac{1}{[n]_q[\alpha + 1]_q}} \left(x + \frac{1}{[n]_q[\alpha + 1]_q} - u\right)g^{''}(u)du\right| \\
&\leq \left\|g^{''}\right\|K_{n,q}^{(\alpha)}\left((t-x)^2;x\right) + \left(\frac{1}{[n]_q[\alpha + 1]_q}\right)^2 \left\|g^{''}\right\|. \n\end{aligned}
$$

Finally, from Lemma $\frac{1}{4}$, we can write

$$
\left|^* K_{n,q}^{(\alpha)}(g;x) - g(x) \right|
$$

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$$
\leq \left(\frac{[2]_q}{[\alpha+1]_q [\alpha+2]_q [n]_q^2} + \frac{x}{[n]_q} + x^2 \left(\frac{1}{q} - 1 \right) + \left(\frac{1}{[n]_q [\alpha+1]_q} \right)^2 \right) ||g''||.
$$

Theorem 1. Let $f \in C_B[0,\infty)$, $0 < q < 1$ and $\alpha > 0$. Then, for every $x \in [0,\infty)$, there exists a constant $M > 0$ such that

$$
\left| K_{n,q}^{(\alpha)}(f;x) - f(x) \right| \le M \omega_2 \left(f; \sqrt{\delta_n^{(\alpha)}(x)} \right) + \omega \left(f; \beta_n^{(\alpha)}(x) \right)
$$

where

$$
\delta_n^{(\alpha)}(x) = \left(\frac{[2]_q}{[\alpha+1]_q \left[\alpha+2\right]_q \left[n\right]_q^2} + \frac{x}{[n]_q} + x^2 \left(\frac{1}{q}-1\right) + \left(\frac{1}{[n]_q \left[\alpha+1\right]_q}\right)^2\right) \left\|g^{''}\right\|
$$

and

$$
\beta^{(\alpha)}(x) = \frac{1}{\beta+1} \left\|g^{(\alpha)}(x)\right\|_q^2
$$

$$
\beta_n^{(\alpha)}(x) = \left| \frac{1}{[n]_q [\alpha + 1]_q} \right|.
$$

Proof. It follows from Lemma \circledR , that

$$
\begin{split} &\left|K_{n,q}^{(\alpha)}(f;x)-f(x)\right| \\ &\leq \left|{^*K_{n,q}^{(\alpha)}(f;x)-f(x)}\right|+\left|f(x)-f\left(x+\frac{1}{[n]_q\left[\alpha+1\right]_q}\right)\right| \\ &\leq \left|{^*K_{n,q}^{(\alpha)}(f-g;x)-(f-g)(x)}\right| \\ &+\left|f(x)-f\left(x+\frac{1}{[n]_q\left[\alpha+1\right]_q}\right)\right|+\left|{^*K_{n,q}^{(\alpha)}(g;x)-g(x)}\right| \\ &\leq \left|{^*K_{n,q}^{(\alpha)}(f-g;x)}\right|+|(f-g)(x)| \\ &+\left|f(x)-f\left(x+\frac{1}{[n]_q\left[\alpha+1\right]_q}\right)\right|+\left|{^*K_{n,q}^{(\alpha)}(g;x)-g(x)}\right|. \end{split}
$$

Since boundedness of the $K_{n,q}^{(\alpha)}$ and using inequality [\(13\)](#page-249-3), we get

$$
\begin{aligned}\n&\left| K_{n,q}^{(\alpha)}(f;x) - f(x) \right| \\
&\le 4 \left\| f - g \right\| + \left| f(x) - f\left(x + \frac{1}{[n]_q \left[\alpha + 1 \right]_q} \right) \right| \\
&+ \left(\frac{[2]_q}{[\alpha + 1]_q \left[\alpha + 2 \right]_q [n]_q^2} + \frac{x}{[n]_q} + x^2 \left(\frac{1}{q} - 1 \right) + \left(\frac{1}{[n]_q \left[\alpha + 1 \right]_q} \right)^2 \right) \left\| g'' \right\| \n\end{aligned}
$$
$$
\leq 4\left\|f-g\right\|+\omega\left(f;\left|\frac{1}{\left[n\right]_q\left[\alpha+1\right]_q}\right|\right)+\delta_n^{(\alpha)}(x)\left\|g^{''}\right\|.
$$

Now, taking infimum on the right hand side over all $g \in w^2$ and using the property of Peetre's K-functional($\boxed{11}$, we can get

$$
\left| K_{n,q}^{(\alpha)}(f;x) - f(x) \right| \le 4K_2 \left(f; \delta_n^{(\alpha)}(x) \right) + \omega \left(f; \beta_n^{(\alpha)}(x) \right)
$$

$$
\le M \omega_2 \left(f; \sqrt{\delta_n^{(\alpha)}(x)} \right) + \omega \left(f; \beta_n^{(\alpha)}(x) \right).
$$

Corollary 1. Let $0 < q_n < 1$, $\alpha > 0$. For any $A > 0$ and $f \in C_B [0, \infty)$, then $K_{n,q_n}^{(\alpha)}(f;x)$ converges to uniformly f on $[0, A]$ if and only if $q_n \to 1$ as $n \to \infty$.

Theorem 2. Let $K_{n,q}^{(\alpha)}$ be the operators defined by $[\alpha]$, $0 < q < 1$, $\alpha > 0$, $\rho \in (0,1]$ and D be any subset of the interval $[0, \infty)$. if $f \in C_B [0, \infty)$ is locally $Lip(\rho)$ on D, i.e., if f satisfies the following inequality:

$$
|f(t) - f(x)| \le C_{f,\rho} |t - x|^\rho, \qquad t \in D \text{ and } x \in [0, \infty), \tag{16}
$$

then for each $x \in [0, \infty)$, we have

$$
\left| K_{n,q}^{(\alpha)}(f;x) - f(x) \right| \leq C_{f,\rho} \left\{ \left(K_{n,q}^{(\alpha)} \left(\left(t - x \right)^2 ; x \right) \right)^{\frac{\rho}{2}} + 2d^{\rho}(x,D) \right\}
$$

where $C_{f,\rho}$ is constant depending on f and ρ and $d(x, D)$ is the distance between x and D defined by

$$
d(x, D) = \inf \left\{ |t - x| : t \in D \right\}.
$$

Proof. Let \overline{D} denote the closure of D. Due to the features of infimum, there is at least a point $t_0 \in \overline{D}$ such that $d(x, D) = |x - t_0|$. By the triangle inequality

$$
|f(t) - f(x)| \le |f(t) - f(t_0)| + |f(x) - f(t_0)|.
$$

Applying $K_{n,q}^{(\alpha)}$ to the above inequality and using (16) , we can get

$$
\begin{split} &\left| K_{n,q}^{(\alpha)}(f;x) - f(x) \right| \\ &\leq K_{n,q}^{(\alpha)} \left(|f(t) - f(t_0)| \, ; x \right) + K_{n,q}^{(\alpha)} \left(|f(x) - f(t_0)| \, ; x \right) \\ &\leq C_{f,\rho} \left\{ K_{n,q}^{(\alpha)} \left(|t - t_0|^{\rho} \, ; x \right) + |x - t_0|^{\rho} \right\} \\ &\leq C_{f,\rho} \left\{ K_{n,q}^{(\alpha)} \left(|t - x|^{\rho} + |x - t_0|^{\rho} \, ; x \right) + |x - t_0|^{\rho} \right\} \\ &\quad= C_{f,\rho} \left\{ K_{n,q}^{(\alpha)} \left(|t - x|^{\rho} \, ; x \right) + 2 \left| x - t_0 \right|^{\rho} \right\}. \end{split}
$$

Choosing $a_1 = \frac{2}{\rho}$ and $a_2 = \frac{2}{2-\rho}$ and applying Hölder inequality, we have:

 $\overline{}$ $\overline{}$ \mid

$$
\Big| K_{n,q}^{(\alpha)}(f;x) - f(x)
$$

,

$$
\leq C_{f,\rho} \left\{ \left(K_{n,q}^{(\alpha)} \left(|t-x|^{a_1 \rho}; x \right) \right)^{\frac{1}{a_1}} \left(K_n^{(\alpha)} \left(1^{a_2}; x \right) \right)^{\frac{1}{a_2}} + 2d^{\rho}(x, D) \right\} \n\leq C_{f,\rho} \left\{ \left(K_{n,q}^{(\alpha)} \left((t-x)^2; x \right) \right)^{\frac{\rho}{2}} + 2d^{\rho}(x, D) \right\} .
$$

In [\[38\]](#page-273-0), Lipcshitz type maximal function of the order ρ defined as

$$
\phi_{\rho}(f;x) = \sup_{x,t \in [0,\infty), x \neq t} \frac{|f(t) - f(x)|}{|t - x|^{\rho}}
$$
(17)

where $x \in [0, \infty)$ and $\rho \in (0, 1]$. In the next theorem we obtain local direct estimate of the operators $K_{n,q}^{(\alpha)}$ by using (17) .

Theorem 3. Let $f \in C_B[0,\infty)$, $0 < q < 1$, $\alpha > 0$ and $\rho \in (0,1]$. Then, for all $x \in [0, \infty)$, we have

$$
\left| K_{n,q}^{(\alpha)}(f;x) - f(x) \right| \leq \phi_{\rho}(f;x) \left(K_{n,q}^{(\alpha)} \left((t-x)^2 ; x \right) \right)^{\frac{\rho}{2}}.
$$

Proof. From the equation (17) , we have

$$
\left| K_{n,q}^{(\alpha)}(f;x) - f(x) \right| \le \phi_{\rho}(f;x) K_{n,q}^{(\alpha)}(|t-x|^{\rho};x)
$$

Applying the Hölder inequality with $a_1 = \frac{2}{\rho}$ and $a_2 = \frac{2}{2-\rho}$, we get

$$
\left| K_{n,q}^{(\alpha)}(f;x) - f(x) \right| \leq \phi_{\rho}(f;x) \left(K_{n,q}^{(\alpha)} \left((t-x)^2 ; x \right) \right)^{\frac{\rho}{2}}.
$$

Theorem 4. For $0 < q < 1$, $\alpha > 0$, $f \in C_2[0,\infty)$, $w_{a+1}(f;\delta)$ is the modulus of continuity of f on the interval $[0, a + 1] \subset [0, \infty)$, $a > 0$. Then, we have

$$
\left\| K_{n,q}^{(\alpha)}(f;x) - f(x) \right\|_{C[0,a]} \le 4N_f \left(1 + a^2 \right) \delta_n(x) + 2w_{a+1} \left(f; \sqrt{\delta_n(x)} \right).
$$

where $\sqrt{K_{n,q}^{(\alpha)}((t-x)^2;x)}$ given by Lemma 4 and $||f||_{C[0,a]} = \sup_{x \in [0,a]}$ $|f(x)|$.

Proof. For $0 \le x \le a$ and $a + 1 < t$, since $1 < t - x$, we have

$$
|f(t) - f(x)| \le M_f (x^2 + t^2 + 2)
$$

\n
$$
\le M_f (2(t - x)^2 + 2 + 3x^2)
$$

\n
$$
\le M_f (t - x)^2 (4 + 3x^2)
$$

\n
$$
\le 4M_f (t - x)^2 (1 + a^2).
$$
 (18)

Also, for $0 \le x \le a$ and $a + 1 \ge t$, we have

$$
|f(t) - f(x)| \le w_{a+1} (f; |t - x|)
$$

$$
\leq \left(1 + \frac{|t - x|}{\delta}\right) w_{a+1}(f; \delta), \tag{19}
$$

with $\delta > 0$.

For $0 \le x \le a$ and $t \ge 0$, combining [\(18\)](#page-253-1) and [\(19\)](#page-254-0) gives

$$
|f(t) - f(x)|
$$
\n
$$
\leq 4M_f(t - x)^2 (1 + a^2) + \left(1 + \frac{|t - x|}{\delta}\right) w_{a+1}(f; \delta),
$$
\n(20)

Applying Cauchy-Schwarz's inequality to the above inequality (20) , we get

$$
\begin{aligned}\n\left| K_{n,q}^{(\alpha)}(f;x) - f(x) \right| \\
&\leq K_{n,q}^{(\alpha)}(f;x) \left(|f(t) - f(x)|;x \right) \\
&\leq 4M_f \left(1 + a^2 \right) K_{n,q}^{(\alpha)} \left((t - x)^2; x \right) + \left(1 + \frac{\sqrt{K_{n,q}^{(\alpha)} \left((t - x)^2; x \right)}}{\delta} \right) w_{a+1} \left(f; \delta \right) \\
&\leq 4M_f \left(1 + a^2 \right) K_{n,q}^{(\alpha)} \left((t - x)^2; x \right) + 2w_{a+1} \left(f; \delta_n(x) \right) \\
&\text{on choosing } \delta := \delta_n(x) = \sqrt{K_{n,q}^{(\alpha)} \left((t - x)^2; x \right)}.\n\end{aligned}
$$

5. Weighted Approximation

Theorem 5. Let $q = q_n \in (0,1]$ such that $q_n \to 1$ as $n \to \infty$ and $\alpha > 0$. Then for each $f \in C_2^*$ $[0, \infty)$, we have:

$$
\lim_{n \to \infty} \| K_{n, q_n}^{(\alpha)}(f; x) - f(x) \|_2 = 0.
$$

Proof. • Since the Korovkin type theorem on the weighted approximation($[12]$, we need to verify

$$
\lim_{n \to \infty} \left\| K_{n,q_n}^{(\alpha)}(t^m; x) - x^m \right\|_2 = 0, \quad m = 0, 1, 2. \tag{21}
$$

• For $m = 0$, obvious.

• For $m = 1$ and $m = 2$, using Lemma $\boxed{3}$, we can write:

$$
\lim_{n \to \infty} \left\| K_{n,q_n}^{(\alpha)}(t; x) - x \right\|_2 = \sup_{x \ge 0} \frac{\left| K_{n,q_n}^{(\alpha)}(t; x) - x \right|}{1 + x^2}
$$
\n
$$
= \sup_{x \ge 0} \frac{1}{1 + x^2} \left| \frac{1}{[n]_{q_n} [\alpha + 1]_{q_n}} \right|
$$
\n
$$
= \frac{1}{[n]_{q_n} [\alpha + 1]_{q_n}} \sup_{x \ge 0} \frac{1}{1 + x^2}
$$
\n
$$
\le \frac{1}{[n]_{q_n} [\alpha + 1]_{q_n}} \to 0, n \to \infty
$$

and
\n
$$
\lim_{n \to \infty} \left\| K_{n,q_n}^{(\alpha)}(t^2; x) - x^2 \right\|_2
$$
\n
$$
= \sup_{x \ge 0} \frac{\left| K_{n,q_n}^{(\alpha)}(t^2; x) - x^2 \right|}{1 + x^2}
$$
\n
$$
= \sup_{x \ge 0} \frac{1}{1 + x^2} \left| \frac{[2]_q}{[\alpha + 1]_{q_n} [\alpha + 2]_{q_n} [n]_{q_n}^2} + \frac{\left(2 + [\alpha + 1]_q\right)}{[\alpha + 1]_{q_n} [n]_{q_n}} x + \frac{x^2}{q_n} - x^2 \right|
$$
\n
$$
\le \left(\frac{1}{q_n} - 1\right) \sup_{x \ge 0} \frac{x^2}{1 + x^2} + \frac{\left(2 + [\alpha + 1]_{q_n}\right)}{[\alpha + 1]_{q_n} [n]_{q_n}} \sup_{x \ge 0} \frac{x}{1 + x^2}
$$
\n
$$
+ \frac{[2]_q}{[\alpha + 1]_{q_n} [\alpha + 2]_{q_n} [n]_{q_n}^2} \sup_{x \ge 0} \frac{1}{1 + x^2}
$$
\n
$$
\le \left(\frac{1}{q_n} - 1\right) + \frac{\left(2 + [\alpha + 1]_{q_n}\right)}{[\alpha + 1]_{q_n} [n]_{q_n}} + \frac{[2]_q}{[\alpha + 1]_{q_n} [\alpha + 2]_{q_n} [n]_{q_n}^2} \to 0, n \to \infty,
$$
\nwhich implies that

$$
\lim_{n \to \infty} \| K_{n,q}^{(\alpha)}(t^m; x) - x^m \|_2 = 0, \quad m = 0, 1, 2.
$$

In the next theorem, we present a weighted approximation theorem for $f \in$ $C_2^*[0, \infty)$, where Doğru studied for classical Szász operators in [\[13\]](#page-272-0).

Theorem 6. Let $q = q_n \in (0,1]$ such that $q_n \to 1$ as $n \to \infty$ and $\alpha > 0$. For each $f \in C_2^* [0, \infty)$ and $\beta > 0$, we have

$$
\lim_{n \to \infty} \sup_{x \ge 0} \frac{\left| K_{n,q_n}^{(\alpha)}(f;x) - f(x) \right|}{(1+x^2)^{1+\beta}} = 0.
$$

Proof. Let $x_0 \in [0, \infty)$ be arbitrary but fixed. Then

$$
\sup_{x \in [0,\infty)} \frac{\left| K_{n,q_n}^{(\alpha)}(f;x) - f(x) \right|}{(1+x^2)^{1+\beta}}
$$
\n
$$
= \sup_{x \in [0,x_0]} \frac{\left| K_{n,q_n}^{(\alpha)}(f;x) - f(x) \right|}{(1+x^2)^{1+\beta}} + \sup_{x \in (x_0,\infty)} \frac{\left| K_{n,q_n}^{(\alpha)}(f;x) - f(x) \right|}{(1+x^2)^{1+\beta}}
$$
\n
$$
\leq \left\| K_{n,q_n}^{(\alpha)}(f) - f \right\|_{C[0,x_0]} + \|f\|_2 \sup_{x \in (x_0,\infty)} \frac{\left| K_{n,q_n}^{(\alpha)}(1+t^2;x) \right|}{(1+x^2)^{1+\beta}}
$$

+
$$
\sup_{x \in (x_0, \infty)} \frac{|f(x)|}{(1+x^2)^{1+\beta}}
$$

= $H_1 + H_2 + H_3$.

Since $|f(x)| \leq N_f(1+x^2)$, we have

$$
H_3 = \sup_{x \in (x_0, \infty)} \frac{|f(x)|}{(1+x^2)^{1+\beta}} \le \sup_{x \in (x_0, \infty)} \frac{N_f}{(1+x^2)^{\beta}} \le \frac{N_f}{(1+x_0^2)^{\beta}}.
$$

Firstly, From Theorem (4) , we have

 H_1 goes to zero as $n \to \infty$.

Secondly, by Theorem 5 ,

$$
H_2 = ||f||_2 \lim_{n \to \infty} \sup_{x \in (x_0, \infty)} \frac{\left| K_{n,q}^{(\alpha)} (1 + t^2; x) \right|}{(1 + x^2)^{1 + \beta}}
$$

=
$$
\sup_{x \in (x_0, \infty)} \frac{(1 + x^2)}{(1 + x^2)^{1 + \beta}} ||f||_2
$$

=
$$
\sup_{x \in (x_0, \infty)} \frac{||f||_2}{(1 + x^2)^{\beta}} \le \frac{||f||_2}{(1 + x_0^2)^{\beta}}.
$$

Moreover, if we choose $x_0 > 0$ large enough, we can see that

$$
H_2 \to 0
$$
 and $H_3 \to 0$ as $n \to \infty$,

Combining, H_1, H_2 and H_3 , we get desired result. \Box

In the next theorem we obtain direct estimation in terms of weighted modulus of continuity. For every $f \in C_m^*[0, \infty)$ the weighted modulus of continuity defined as $|f(x + h) - f(x)|$

$$
\Omega_m(f,\delta) = \sup_{x \ge 0, \ 0 < h \le \delta} \frac{|f(x+h) - f(x)|}{1 + (x+h)^m},\tag{22}
$$

Lemma 9. [\[39\]](#page-273-1) If $f \in C_m^*$ [0, ∞), $m \in \mathbb{N}$, then

- (i) $\Omega_m(f,\delta)$ is a monotone increasing function of δ ,
- (*ii*) $\lim_{\delta \to 0^+} \Omega_m(f, \delta) = 0$,
- (iii) for any $\rho \in [0, \infty), \Omega_m(f, \rho \delta) \leq (1 + \rho) \Omega_m(f, \delta).$

In the next theorem, we express the approximation error of $K_{n,q}^{(\alpha)}$ by using Ω_m . **Theorem 7.** For $f \in C_m^*$ [0, ∞), we have

$$
\left\| K_{n,q}^{(\alpha)}(f) - f \right\|_{m+1} \leq N\Omega_m(f, (1/\sqrt{qn})),
$$

where N is a constant independent of f and n .

Proof. From (22) and Lemma $\overline{9}$, we can write

$$
|f(t) - f(x)| \le (1 + (x + |t - x|)^m) \left(\frac{|t - x|}{\delta} + 1\right) \Omega_m(f, \delta)
$$

$$
\le (1 + (2x + t)^m) \left(\frac{|t - x|}{\delta} + 1\right) \Omega_m(\varphi, \delta).
$$

Then, we have

$$
\begin{aligned}\n&\left|K_{n,q}^{(\alpha)}(f;x) - f(x)\right| \\
&\leq K_{n,q}^{(\alpha)}\left|\left(f(t) - f(x)\right|;x\right) \\
&\leq \Omega_m(f,\delta)\left(K_{n,q}^{(\alpha)}((1+(2x+t)^m);x) + K_{n,q}^{(\alpha)}\left((1+(2x+t)^m)\frac{|t-x|}{\delta};x\right)\right). \\
&= \Omega_m(f,\delta)\left(K_{n,q}^{(\alpha)}(1+(2x+t)^m;x) + I_1\right).\n\end{aligned}
$$

Applying Cauchy-Schwartz inequality to the $\mathcal{I}_1,$ we get

$$
I_1 \le (K_{n,q}^{(\alpha)}((1+(2x+t)^m)^2;x))^{1/2}\left(K_{n,q}^{(\alpha)}\left(\frac{|t-x|^2}{\delta^2};x\right)\right)^{1/2}.
$$

Therefore,

$$
\begin{split} & \left| K_{n,q}^{(\alpha)}(f;x) - f(x) \right| \\ &\leq \Omega_m(f,\delta) K_{n,q}^{(\alpha)}((1 + (2x + t)^m);x) \\ &+ \Omega_m(f,\delta) (K_{n,q}^{(\alpha)}\left((1 + (2x + t)^m)^2; x)\right)^{1/2} \left(K_{n,q}^{(\alpha)}\left(\frac{|t-\tau|^2}{\delta^2};x\right) \right)^{1/2} . \end{split} \tag{24}
$$

By Lemma $\boxed{7}$ and Lemma $\boxed{4}$

$$
K_{n,q}^{(\alpha)}(1 + (2x+t)^m; x) \le C_{m,j}(q,\alpha) (1+x^m),
$$

$$
(K_{n,q}^{(\alpha)}((1 + (2x+t)^m)^2; x))^{1/2} \le C_{m,j}^1(q,\alpha) (1+x^m).
$$
 (25)

and

$$
\left(K_{n,q}^{(\alpha)}\left(\frac{\left|t-x\right|^2}{\delta^2};x\right)\right)^{1/2} \leq \frac{1}{\delta}\sqrt{\frac{[2]_q}{[\alpha+1]_q\left[\alpha+2\right]_q\left[n\right]_q^2} + \frac{x}{[n]} + x^2\left(\frac{1}{q}-1\right)}
$$

$$
\leq \frac{(2+x)}{\delta\sqrt{qn}}.
$$
\n(26)

Combining $\boxed{23}$, $\boxed{25}$ and $\boxed{26}$, we have

$$
\begin{aligned} & \left| K_{n,q}^{(\alpha)}(f;x) - f(x) \right| \\ &\leq \Omega_m(f,\delta) \left(C_{m,j}(q,\alpha) \left(1 + x^m \right) + C_{m,j}^1(q,\alpha) \frac{\left(1 + x^m \right) (2+x)}{\delta \sqrt{qn}} \right) \end{aligned}
$$

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$$
= \Omega_m(f,\delta) \left(C_{m,j}(q,\alpha) \left(1+x^m\right) + C_{m,j}^1(q,\alpha) C_1 \frac{\left(1+x^{m+1}\right)}{\delta \sqrt{qn}} \right),
$$

where

$$
C_1 = \sup_{x \ge 0} \frac{\left(2 + 2x^m + x + 2x^{m+1}\right)}{1 + x^{m+1}}.
$$

if we take $\delta = \left(1/\sqrt{q\left[n\right]_q}\right)$ in the above inequality, we obtain the desired result. \Box

Next result is a Voronovskaja type formula for the operators $K_{n,q}^{(\alpha)}(f; x)$.

6. Voronovskaja Type

Theorem 8. Let $q = q_n \in (0,1]$ such that $q_n \to 1$, $q_n^n \to b$ as $n \to \infty$ and $\alpha > 0$. For any $f \in C_2^*$ $[0, \infty)$ such that f' , $f'' \in C_2^*$ $[0, \infty)$ the following equality holds

$$
\lim_{n \to \infty} [n]_{q_n} \left[K_{n,q_n}^{(\alpha)}(f;x) - f(x) \right]
$$

= $\frac{1}{(\alpha + 1)} f'(x) + \frac{1}{2} (x + x^2 (1 - b)) f''(x).$

Proof. By the Taylor's formula, we can write

$$
f(t) = f(x) + f'(x)(t - x) + \frac{1}{2}f''(x)(t - x)^2 + r(t, x)(t - x)^2
$$
 (27)

where $r(t, x)$ is Peano form of remainder, $r(., x) \in C_2^* [0, \infty)$ and $\lim_{t \to x} r(t, x) = 0$.

Applying $K_{n,q}^{(\alpha)}$ to the both sides of (27) , we get

$$
[n]_{q_n} \left[K_{n,q_n}^{(\alpha)}(f;x) - f(x) \right]
$$

= $f'(x) [n]_{q_n} K_{n,q_n}^{(\alpha)}((t-x);x) + \frac{1}{2} f''(x) [n]_{q_n} K_{n,q_n}^{(\alpha)}((t-x)^2;x)$
+ $[n]_{q_n} K_{n,q_n}^{(\alpha)} (r(t,x)(t-x)^2;x).$

By Cauchy-Schwarz inequality, we have

$$
K_{n,q_n}^{(\alpha)}\left(r(t,x)(t-x)^2;x\right) \le \sqrt{K_{n,q_n}^{(\alpha)}\left(r^2(t,x);x\right)}\sqrt{K_{n,q_n}^{(\alpha)}\left((t-x)^4;x\right)}.\tag{28}
$$

Observe that $r^2(t, x) = 0$ and $r^2(., x) \in C_2^*[0, \infty)$. Then, it follows from that Corollary (1) ,

$$
\lim_{n \to \infty} [n]_{q_n} K_{n,q_n}^{(\alpha)} (r^2(t,x);x) = r^2(x,x) = 0.
$$
 (29)

Moreover, from $[6]$, $[28]$ and $[29]$, we can obtain

$$
\lim_{n \to \infty} K_{n,q_n}^{(\alpha)} \left(r(t,x)(t-x)^2; x \right) = 0 \tag{30}
$$

Hence, combining $\langle 4 \rangle$, $\langle 5 \rangle$ and $\langle 30 \rangle$, we get

$$
\lim_{n \to \infty} [n]_{q_n} \left[K_{n,q_n}^{(\alpha)}(f;x) - f(x) \right]
$$
\n
$$
= \frac{1}{(\alpha+1)} f'(x) + \frac{1}{2} (x + x^2 (1-b)) f''(x).
$$

7. Bivariate Fractional q-Integral

In this section, we introduce the bivariate fractional q -integral of Riemann-Liouville integral type $K_{n,q}^{(\alpha)}(f;x)$ [\(2\)](#page-244-0) as follows:

$$
K_{n_1,n_2,q_1,q_2}^{(\alpha_1,\alpha_2)}(f;x,y)
$$

=
$$
\sum_{k_1=0}^{\infty} \sum_{k_2=0}^{\infty} [\alpha_1]_{q_1} [\alpha_2]_{q_2} s_{n_1,k_1}(q_1;x) s_{n_2,k_2}(q_2;y)
$$

$$
\int_{0}^{1} \int_{0}^{1} f\left(\frac{q_1^{1-k_1} [k_1]_{q_1} + t_1}{[n_1]_{q_1}}, \frac{q_2^{1-k_2} [k_2]_{q_2} + t_2}{[n_2]_{q_2}}\right) (1-t_1)^{\alpha_1-1} (1-t_2)^{\alpha_2-1} d_{q_1} t_1 d_{q_2} t_2
$$

where $(x, y) \in I^2 = [0, \infty) \times [0, \infty)$ and $\alpha_1, \alpha_2 > 0$.

Fractional q-integral of Riemann-Liouville integral type $K_{n_1,n_2,q_1,q_2}^{(\alpha_1,\alpha_2)}(\cdot; x, y)$ can be rewritten as

$$
K_{n_1,n_2,q_1,q_2}^{(\alpha_1,\alpha_2)}(\cdot;x,y)=K_{n_1,q_1}^{(\alpha_1)}(\cdot;x)\times K_{n_2,q_2}^{(\alpha_2)}(\cdot;y).
$$

Lemma 10. Let $e_{ij}(x, y) = x^i y^j$, $0 < q_1, q_2 < 1$, $0 \le i + j \le 2$ and $\alpha_1, \alpha_2 > 0$. For $(x, y) \in I^2 = [0, \infty) \times [0, \infty)$, we have

$$
K_{n_1,n_2,q_1,q_2}^{(\alpha_1,\alpha_2)}(e_{00};x,y) = 1,
$$

\n
$$
K_{n_1,n_2,q_1,q_2}^{(\alpha_1,\alpha_2)}(e_{10};x,y) = x + \frac{1}{[n_1]_{q_1} [\alpha_1 + 1]_{q_1}},
$$

\n
$$
K_{n_1,n_2,q_1,q_2}^{(\alpha_1,\alpha_2)}(e_{01};x,y) = y + \frac{1}{[n_2]_{q_2} [\alpha_2 + 1]_{q_2}},
$$

\n
$$
K_{n_1,n_2,q_1,q_2}^{(\alpha_1,\alpha_2)}(e_{20};x,y) = \frac{[2]_{q_1}}{[\alpha_1 + 1]_{q_1} [\alpha_1 + 2]_{q_1} [n_1]_{q_1}^2} + \frac{2 + [\alpha_1 + 1]_{q_1}}{[\alpha_1 + 1]_{q_1} [n_1]_{q_1}}x + \frac{x^2}{q_1},
$$

\n
$$
K_{n_1,n_2,q_1,q_2}^{(\alpha_1,\alpha_2)}(e_{02};x,y) = \frac{[2]_{q_2}}{[\alpha_2 + 1]_{q_1} [\alpha_2 + 2]_{q_2} [n_2]_{q_2}^2} + \frac{2 + [\alpha_2 + 1]_{q_2}}{[\alpha_2 + 1]_{q_2} [n_2]_{q_2}}y + \frac{y^2}{q_2}.
$$

Remark 1. According to above Lemma $\overline{10}$, we get

$$
K_{n_1,n_2,q_1,q_2}^{(\alpha_1,\alpha_2)}(e_{10}-x;x,y)=\frac{1}{[n_1]_{q_1}[\alpha_1+1]_{q_1}},
$$

$$
K_{n_1, n_2, q_1, q_2}^{(\alpha_1, \alpha_2)}(e_{01} - y; x, y) = \frac{1}{[n_2]_{q_2} [\alpha_2 + 1]_{q_2}},
$$

\n
$$
K_{n_1, n_2, q_1, q_2}^{(\alpha_1, \alpha_2)}((e_{10} - x)^2; x, y) = \frac{[2]_{q_1}}{[\alpha_1 + 1]_{q_1} [\alpha_1 + 2]_{q_1} [n_1]_{q_1}^2} + \frac{x}{[n_1]_{q_1}} + x^2 \left(\frac{1}{q_1} - 1\right)
$$

\n
$$
= \delta_{n_1}^{(\alpha_1)}(q_1; x),
$$

\n
$$
K_{n_1, n_2, q_1, q_2}^{(\alpha_1, \alpha_2)}((e_{01} - y)^2; x, y) = \frac{[2]_{q_2}}{[\alpha_2 + 1]_{q_2} [\alpha_2 + 2]_{q_2} [n_2]_{q_2}^2} + \frac{y}{[n_2]_{q_2}} + y^2 \left(\frac{1}{q_2} - 1\right)
$$

\n
$$
= \delta_{n_2}^{(\alpha_2)}(q_2; y).
$$

In the next theorem, we obtain the uniform convergence of the bivariate q -Riemann-Liouville fractional integral type of q-Szász-Mirakyan-Kantorovich operators to the bivariate functions defined on $I^2 = [0, \infty) \times [0, \infty)$.

Theorem 9. Let $C(I^2)$ be the space of continuous bivariate function on $I^2 =$ $[0, \infty) \times [0, \infty)$ and $\alpha_1, \alpha_2 > 0$. Then for any $f \in C(I^2)$, we have

$$
\lim_{n_1, n_2 \to \infty} \left\| K_{n_1, n_2, q_1, q_2}^{(\alpha_1, \alpha_2)} f - f \right\| = 0.
$$

Proof. Using lemma $\boxed{1}$, we get

$$
\left\| K_{n_1, n_2, q_1, q_2}^{(\alpha_1, \alpha_2)} e_{00} - e_{00} \right\| = 0, \left\| K_{n_1, n_2, q_1, q_2}^{(\alpha_1, \alpha_2)} e_{10} - e_{10} \right\| \to 0
$$

$$
\left\| K_{n_1, n_2, q_1, q_2}^{(\alpha_1, \alpha_2)} e_{01} - e_{01} \right\| \to 0, \left\| K_{n_1, n_2, q_1, q_2}^{(\alpha_1, \alpha_2)} (e_{20} + e_{02}) - (e_{20} + e_{02}) \right\| \to 0
$$

as $n_1, n_2 \to \infty$

As a result, by Volkov's theorem $[40]$, we get

$$
\lim_{n_1, n_2 \to \infty} \left\| K_{n_1, n_2, q_1, q_2}^{(\alpha_1, \alpha_2)} f - f \right\| = 0.
$$

For bivariate real functions, modulus of continuity defined as

 $w(f; \delta_n, \delta_m) = \sup \{|f(t, s) - f(x, y)| : (t, s), x, y \in I^2, |t - x| \leq \delta_n, |s - y| \leq \delta_m\}.$ **Theorem 10.** Let $f \in C(I^2)$, $0 < q_1, q_2 < 1$ and $\alpha_1, \alpha_2 > 0$. Then for all $(x, y) \in$ I^2 , the inequality

$$
\left| K_{n_1,n_2,q_1,q_2}^{(\alpha_1,\alpha_2)}(f;x,y) - f(x,y) \right| \le 4w \left(f; \delta_{n_1}^{(\alpha_1)}(q_1;x), \delta_{n_2}^{(\alpha_2)}(q_2;y) \right)
$$

holds, where $\delta_{n_1}^{(\alpha_1)}(q_1; x), \delta_{n_2}^{(\alpha_2)}(q_2; y)$ are as in Remark [1.](#page-259-1)

Proof. By the positivity and linearity properties of the $K_{n_1,n_2,q_1,q_2}^{(\alpha_1,\alpha_2)}$, we can write $\left| K_{n_1,n_2,q_1,q_2}^{(\alpha_1,\alpha_2)}(f;x,y)-f(x,y)\right| \leq K_{n_1,n_2,q_1,q_2}^{(\alpha_1,\alpha_2)}\left(|f(t,s)-f(x,y)|\,;x,y\right)$

$$
\leq w(f; \delta_1, \delta_2) \left(K_{n_1, q_1}^{(\alpha_1)}(1; x) + \frac{1}{\delta_1} K_{n_1, q_1}^{(\alpha_2)}(|t - x|; x) \right) \times \left(K_{n_2, q_2}^{(\alpha_2)}(1; y) + \frac{1}{\delta_2} K_{n_2, q_2}^{(\alpha_2)}(|s - y|; y) \right)
$$

Applying Cauchy-Schwarz inequality, we obtain

$$
K_{n_1,q_1}^{(\alpha_1)} (|t-x|;x) \le K_{n_1,q_1}^{(\alpha_1)} (|t-x|^2; x)^{\frac{1}{2}}
$$

$$
K_{n_2,q_2}^{(\alpha_2)} (|s-y|;y) \le K_{n_2,q_2}^{(\alpha_2)} (s-y)^2; y^{\frac{1}{2}}
$$

Choosing $\delta_1 = \delta_{n_1}^{(\alpha_1)}(q_1; x)$ and $\delta_2 = \delta_{n_2}^{(\alpha_2)}(q_2; y)$, we have desired result. \Box

Now, we are present some graphs and numerical results for $K_{n,q}^{(\alpha)}$ and $K_{n_1,n_2,q_1,q_2}^{(\alpha_1,\alpha_2)}$ obtained by using Matlab.

8. Graphical Simulations

Example 1. Consider $f(x) = x^3 - 9x^2 - 15x + 9$ with $x \in [0, 6]$. Here we take the value of $q \in \{0.75, 0.85, 0.95\}$, for $K_{100,q}^{(2)}$. The Figure 1 demonstrate the convergence of operators $K_{100,q}^{(2)}$ to $f(x)$ for increasing values of q and fixed α, n . Moreover, absolute error function $E_{100,q}^{(2)}(f;x) = \left| K_{100,q}^{(2)}(f;x) - f(x) \right|$ is illustrated in Figure $\overline{}$ 2. Then, numerical values of $E^{(2)}_{100,q}(f;x)$ at some points on the interval [0,6] for ${q \in 0.75, 0.85, 0.95}$ are given in Table 1.

TABLE 1. Estimation of the absolute error function $E_{100,q}^{(2)}$ with $f(x) = x^3 - 9x^2 - 15x + 9$ for some values of x in [0,6] and $q \in$ $\{0.75, 0.85, 0.95\}.$

\boldsymbol{x}	.00, 0.75	100,0.85	100,0.95
θ	0.479	0.170	0.056
1	1.830	0.761	0.178
$\overline{2}$	1.013	0.283	0.692
3	16.273	6.732	3.551
$\overline{4}$	52.172	22.357	9.397
5	116.932	50.926	19.228
6	218.776	96.211	34.043

As we increase the value of q and fixed α and n, the approximation is good, i.e for the largest value of q, the error is minumum.

FIGURE 1. Approximation to $f(x) = x^3 - 9x^2 - 15x + 9$ by $K_{100,q}^{(2)}(f;x)$ for $q \in \{0.75, 0.85, 0.95\}.$

FIGURE 2. $E_{100,q}^{(2)}(f;x)$ for $f(x) = x^3 - 9x^2 - 15x + 9$ and $q = \{0.75, 0.85, 0.95\}.$

Example 2. Let $f(x) = x^6$ with $x \in [0, 6]$. Here we take the value of $n \in \{10, 100\}$, $\alpha = 5$ and $q = 0.95$. The Figure 3 demonstrate the convergence of operators

 $K_{n,0.95}^{(5)}$ to $f(x)$ for increasing values of n. Secondly, The absolute error function $E_{n,0.95}^{(5)}(f;x) = \left| K_{n,0.95}^{(5)}(f;x) - f(x) \right|$ is illustrated in Figure 4. Finally, numerical values of $E_{n,0.95}^{(5)}(f;x)$ at some points on the interval [0,6] for $n \in \{10,100\}$ are given in Table 2.

FIGURE 3. Approximation to $f(x) = x^6$ by $K_{n,0.95}^{(5)}(f; x)$ for $n \in \{10, 100\}.$

TABLE 2. Estimation of the absolute error function $E_{n,0.95}^{(5)}$ with $f(x) = x^6$ for some values of x in [0, 6] and $n \in \{10, 100\}$

\boldsymbol{x}	$E_{10,0.95}^{\circ\circ}$	$E_{100,0.95}^{(9)}$
1	8.57	3.06
\mathfrak{D}	401.68	164.39
3	4067.76	1758.63
4	21481.26	9566.43
5	78862.08	35778.98
6	229422.04	105421.79

As we increase the value of n and fixed α and q, the approximation is good, i.e for the largest value of n, the error is minumum.

FIGURE 4. $E_{n,0.95}^{(5)}(f;x)$ for $n = \{10, 100\}, f(x) = x^6$.

Example 3. Let $f(x) = x^3 - 4x^2 + 2$ with $x \in [0, 5]$. Here we take the value of $\alpha \in \{0.1, 10\}$, $n = 150$ and $q = 0.95$. The Figure 5 demonstrate the convergence of operators $K_{150,0.95}^{(\alpha)}$ to $f(x)$ for increasing values of α . Secondly, The absolute error function $E_{150,0.95}^{(\alpha)}(f;x) = |K_{n,q}^{(\alpha)}(f;x) - f(x)|$ is illustrated in Figure 6. Finally, \vert numerical values of $E_{150,0.95}^{(\alpha)}$ at some points on the interval [3, 5] for $\alpha \in \{0.1, 10\}$ are given in Table 3.

TABLE 3. Estimation of the absolute error function $E_{150,0.95}^{(\alpha)}$ with $f(x) = x^3 - 4x^2 + 2$ for some values of x in [3, 5] and $\alpha \in \{0.1, 10\}$

\boldsymbol{x}	$E^{(0.1)}_{150,0.95}$	$E^{(10)}_{150,0.95}$
3	6.682	6.472
3.5	9.853	9.388
4	13.944	13.161
4.5	19.081	17.917
5	25.388	23.780

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FIGURE 5. Approximation to $f(x) = x$ $3 - 4x^2 + 2$ by $K_{150,0.95}^{(\alpha)}(f;x)$ for $\alpha \in \{0.1, 10\}.$

FIGURE 6. $E_{150,0.95}^{(\alpha)}(f;x)$ for $f(x) = x^3 - 4x^2 + 2$ and $\alpha \in \{0.1, 10\}$.

Now, we are present some graphs and numerical results for the convergence of bivariate fractional q -integral Riemann-Liouville integral type Szász-Mirakyan-Kantorovich operators $K_{n_1,n_2,q_1,q_2}^{(\alpha_1,\alpha_2)}$ by considering the function $f(x,y) = x + y$.

Example 4. Consider, $f(x, y) = x + y$ with $(x, y) \in [0, 4] \times [0, 4]$. Here we take the value of $n_1, n_2 \in \{5, 150\}$, $q_1 = q_2 = 0.75$ and $\alpha_1 = \alpha_2 = 0.1$. The Figure 7 explains the convergence of the operators $K_{n_1,n_2,0.75,0.75}^{(0.1,0.1)}$ towards the function $f(x, y)$ for increasing values of n_1, n_2 . Secondly, The absolute error function $E_{n_1,n_2,0.75,0.75}^{(0.1,0.1)}(f;x,y) = \left| K_{n_1,n_2,0.75,0.75}^{(0.1,0.1)}(f;x,y) - f(x,y) \right|$ is illustrated Figure 8. Finally numerical values of $E^{(0.1,0.1)}_{n_1,n_2,0.75,0.75}$ at some points on the interval $[0,4] \times$ [0, 4] for $n_1, n_2 \in \{5, 150\}$ are given in Table 4.

FIGURE 7. Convergence of the operators $K_{n_1,n_2,0.75,0.75}^{(0.1,0.1)}$ to the function $f(x, y) = x + y$.

As we increase the value of n_1 and n_2 and fixed α_1 , α_2 , q_1 and q_2 , the approximation is good, i.e for the largest value of n_1 and n_2 and fixed α_1 , α_2 , q_1 and q_2 , the error is minumum.

Example 5. Consider $f(x, y) = x + y$ with $(x, y) \in [0, 4] \times [0, 4]$. Here we take the value of $\alpha_1, \alpha_1 \in \{0.1, 10\}$, $q_1 = q_2 = 0.75$ and $n_1 = n_2 = 5$. The Figure 9 explains the convergence of the operators $K_{5,5,0.75,0.75}^{(\alpha_1,\alpha_2)}$ towards the function $f(x, y)$ for increasing values of $\alpha_1, \alpha_2 \in \{0.1, 10\}$. Secondly, absolute error function

FIGURE 8. $E_{n_1,n_2,0.75,0.75}^{(0.1,0.1)}$ with $f(x,y) = x + xy + 12y^2$ for $n_1, n_2 \in$ $\{5, 150\}$ on the interval $[0, 4] \times [0, 4]$.

TABLE 4. Estimation of the absolute error function $E_{n_1,n_2,0.75,0.75}^{(0.1,0.1)}$ with $f(x,y) = x + y$ for some values of (x, y) in $[0,4] \times [0,4]$ and $n_1, n_2 \in \{5, 150\}$.

\boldsymbol{x}	\boldsymbol{y}	$E_{5,5,0.75,0.75}^{(0.1,0.1)}$	$E^{(0.1,0.1)}_{150,150,0.75,0.75}$
$\overline{0}$	θ	0.604	0.461
$\overline{0}$	$0.5\,$	0.604	0.461
$\overline{0}$	$\mathbf{1}$	0.604	0.461
$\overline{0}$	1.5	0.604	0.461
$\overline{0}$	$\overline{2}$	0.604	0.461
$\overline{0}$	2.5	0.604	0.461
$\overline{0}$	3	0.604	0.461
$\overline{0}$	3.5	0.604	0.461
0	4	0.604	0.461

 $E_{5,5,0.75,0.75}^{(\alpha_1,\alpha_2)}(f;x,y) = \left| K_{n_1,n_2,1,q_1,q_2}^{(\alpha_1,\alpha_2)}(f;x,y) - f(x,y) \right|$ is illustrated Figure 10. Finally, numerical values of $E_{5,5,0.75,0.75}^{(\alpha_1,\alpha_2)}$ at some points on the interval $[0,4] \times [0,4]$ for $\alpha_1, \alpha_2 \in \{0.1, 10\}$ are given in Table 5.

FIGURE 9. Convergence of the operators $K_{5,5,0.75,0.75}^{(\alpha_1,\alpha_2)}(f;x,y)$ to the function $f(x, y) = x + y$.

TABLE 5. Estimation of the absolute error function $E_{50,50,0.75,0.75}^{(\alpha_1,\alpha_2)}$ with $f(x,y) = x + y$ for some values of (x, y) in $[0,4] \times [0,4]$ and $\alpha_1, \alpha_2 \in \{0.1, 10\}$.

\boldsymbol{x}	Y	$E_{50,50,0.75,0.75}^{(0.1,0.1)}$	$E_{50,50,0.75,0.75}^{(10,10)}$
0.1	0.1	0.461	0.131
0.1	0.5	0.461	0.131
0.1	1	0.461	0.131
0.1	1.5	0.461	0.131
0.1	$\overline{2}$	0.461	0.131
0.1	2.5	0.461	0.131
0.1	3	0.461	0.131
0.1	3.5	0.461	0.131
0.1	4	0.461	0.131

As we increase the value of α_1 and α_2 and fixed q_1, q_2, n_1 and n_2 , the approximation is good, i.e for the largest value of α_1 and α_2 and fixed q_1, q_2, n_1 and n_2 , the error is minumum.

FIGURE 10. For some (x, y) points, error function $E_{50,50,0.75,0.75}^{(\alpha_1,\alpha_2)}$ with $f(x, y) = x + y$.

Example 6. Consider $f(x) = x + y$ with $(x, y) \in [0, 4] \times [0, 4]$. Here we take the value of $q \in \{0.35, 0.75\}$, $n_1 = n_2 = 10$ and $\alpha_1 = \alpha_1 = 5$ for $K_{n_1,n_2,q_1,q_2}^{(\alpha_1,\alpha_2)}$. The Figure 11 demonstrate the convergence of operators $K_{10,10,q_1,q_2}^{(5,5)}$ to $f(x,y)$ for increasing values of q_1 and q_2 . Moreover, function of absolute error $E_{10,10,q_1,q_2}^{(5,5)}(f;x,y)$ = $\left| K_{n_1,n_2,q_1,q_2}^{(\alpha_1,\alpha_2)}(f;x,y) - f(x,y) \right|$ in is illustrated Figure 12. Then, numerical values $\overline{}$ of $E_{10,10,q_1,q_2}^{(5,5)}$ at some points on the interval $[0,4] \times [0,4]$ for $q_1, q_2 \in \{0.35, 0.75\}$ are given in Table 6.

As we increase the value of q_1 and q_2 and fixed α_1, α_2, n_1 and n_2 , the approximation is good, i.e for the largest value of q_1 and q_2 and fixed α_1, α_2, n_1 and n_2 , the error is minumum.

FIGURE 11. Approximation to $f(x,y) = x + y$ by $K_{10,10,q_1,q_2}^{(5,5)}$
 $q_1, q_2 \in \{0.35, 0.75\}$.

FIGURE 12. $E_{10,10,q_1,q_2}^{(2,2)}(f;x)$ for $f(x,y)=x+y$ and $q_1,q_2=\{0.35,0.75\}.$

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A STUDY ON MODELING OF RAT TUMORS WITH THE DISCRETE-TIME GOMPERTZ MODEL

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Abstract. Cancer formation is one of the pathologies whose frequency has increased in the recent years. In the literature, the compartment models, which are non-linear, are used for such problems. In nonlinear compartment models, nonlinear state space models and the extended Kalman filter (EKF) are used to estimate the parameter and the state vector. This paper presents a discrete-time Gompertz model (DTGM) for the transfer of optical contrast agent, namely indocyanine green (ICG), in the presence of tumors between the plasma and extracellular extravascular space (EES) compartments. The DTGM, which is proposed for ICG and the estimation of ICG densities used in the vascular invasion of tumor cells of the compartments and in the measurement of migration from the intravascular area to the tissues, is obtained from the experimental data of the study. The ICG values are estimated online (recursive) using the DTGM and the adaptive Kalman filter (AKF) based on the experimental data. By employing the data, the results show that the DTGM in conjunction with the AKF provides a good analysis tool for modeling the ICG in terms of mean square error (MSE), mean absolute percentage error (MAPE) and R^2 . When the results obtained from the compartment model used in the reference $\boxed{9}$ are compared with the results obtained with the DTGM, the DTGM gives better results in terms of MSE, MAPE and R^2 criteria. The DTGM and the AKF compartment model require less numerical processing when compared to the EKF, which indicates that DTGM is a less complicated model. In the literature, EKF is used for such problems.

1. Introduction

In recent years the use of optical contrast agents and advanced medical imaging techniques to analyze and diagnose tissue abnormalities has become almost a standard procedure $\left[\mathbf{1}\right]$. The existence of tumors is one of the main causes of tissue

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abnormalities, and in $\boxed{2}$ it is shown that tumor vessel permeability to macromolecular blood solutes correlates with tumor growth as well as vascular growth. ICG is a blood pool agent that binds to globulin proteins (predominantly albumin) in blood $\overline{3}$, and because of its ability to bind to plasma proteins, it behaves as a macromolecular contrast agent with a low or no vascular permeability. Once injected, ICG rapidly and completely binds to albümin. Its macromolecular behavior results in a slow leakage which permits application of a pharmacokinetic model that in return allows for the determination of individual vascular parameters, such as capillary permeability. Compartmental analysis is a method of bio-mathematical modeling which assumes that a biological system can be divided into a series of homogeneous compartments which interact by exchanging material. For compartmental models used in pharmacokinetics, the material concentration varies with time depending on individual pharmacokinetics parameters [\[4\]](#page-283-4). If the appropriate parameters are known, the concentration level in a particular compartment can be predicted by applying suitable pharmacokinetic equations. Thus, a robust method of identifying and estimating individual parameters is required. The parameter identification is a common nonlinear estimation problem. In essence, it is the problem of estimating a model parameter that occurs as a coefficient of a dynamic system state variable - either as a dynamic coefficient or as measurement sensitivity. When this estimation problem is solved simultaneously with the state estimation problem (via state vector augmentation), the linear model becomes nonlinear. The extended Kalman filter (EKF) is one of the most popular and intensively investigated estimation technique for the nonlinear state estimation. It consists of applying the standard Kalman filter equations to the first-order approximation of the nonlinear model of the last estimate $\overline{5}$. This study addresses the most commonly used growth models, the DTGM to estimate the ICG level without resorting to nonlinear models. The growth curves are used for modelling the increase in the number of plants, bacteria or viruses in an environment. The rest of this article is organized as follows. In Section 2, information about The ICG Compartment Model is presented. In Section 3, the mathematical and computational methodologies of DTGM are specified and the mathematical equations, that are aimed to be used further in the study are given, and the modeling analysis and estimation results are also presented. Finally, Section 4 concludes the study.

2. The Icg Compartment Model

If there is a tumor in any tissue, the given ICG passes through the vessel into the tumor tissue area. There is also a return to the vein from the tumor tissue. In accordance with this physiological structure, a two-compartment model can be considered. In this compartment model, C_p indicates ICG concentration in the vessel, C_e indicates ICG concentration in tumor tissue. k_1 ratio is the ratio of ICG passing from the vessel to the tumor tissue, k_2 is the ratio of ICG passing from the tumor area into the vessel, and $k₃$ is the ratio of ICG passing from the plasma to the liver and kidney. Since the mentioned ratio is quite small, this ratio is ignored while creating the mathematical model. The ICG density in the tumor tissue tend to increase as the k_1 ratio increase of the ICG density in the vessel (since there are transitions from here) and tend to decrease as the k_2 ratio of its own density. Accordingly, the change in ICG density in the tumor tissue per unit time is expressed as in Equation $\overline{1}$,

$$
\frac{dC_e(t)}{dt} = k_1 C_p(t) - k_2 C_\theta(t)
$$
\n(1)

As mentioned above, its rate can be ignored, and the change in ICG density in the tumor tissue per unit time is defined as in Equation 2 ,

$$
\frac{dC_p(t)}{dt} = -k_1 C_p(t) + k_2 C_e(t)
$$
\n(2)

Because the ratio of ICG concentration, which is only in the vessel, is expected to be transferred to the tumor tissue per unit time. k_1 and k_2 show the permeability parameters mentioned before. According to the model, there is no information about the permeability parameters and there is no need for their estimates. When the differential equation system given by equations π discrete time-state space model is obtained. In this model, both the parameter and the state vector are required to be estimated simultaneously. In the literature, the EKF is used for such problems $[6]$ - $[11]$.

3. Discrete-Time Gompertz Model

In this study, DTGM, one of the growth models, is used to estimate the ICG level without considering the nonlinear models.

The growth curves are used for modelling the increase in the number of plants, bacteria or viruses in an environment. Expressing the growth of an organism or an increase in the number of viruses temporally is called "growth". The identification of the complex growth process is aimed at using the growth curves $\boxed{12}$ - $\boxed{14}$. DTGM is well known and widely used model in many sub-fields of biology [\[15\]](#page-284-3)- [\[18\]](#page-284-4). Numerous parametrizations and re-parametrizations of the DTGM can be found in the literature [\[17\]](#page-284-5). DTGM was originally recommended to explain human mortality curves Gompertz $\boxed{12}$, and it has been further used in the description of growth processes, for example, growing of bacterial colonies [\[15\]](#page-284-3) and tumors [\[16\]](#page-284-6). The model, a stochastic version of the DTGM, can be transformed into a linear Gaussian state-space model for the convenient fitting to time-series data. In this study, ICG values are estimated online using the DTGM and the AKF based on the experimental data. By employing the data, the results show that the DTGM in conjunction with AKF provides a good analysis tool for modeling the ICG in terms of mean square error (MSE), mean absolute percentage error (MAPE), and $R²$. When the results obtained from the compartment model used in the reference [\[9\]](#page-283-0) are compared with the results obtained with the DTGM, the DTGM gives 1172 L. ÖZBEK

better results in terms of MSE, MAPE and R^2 criteria. The DTGM and the AKF compartment model require less numerical processing when compared to the EKF, which indicates that DTGM is a less complicated model.

Let n_t denote ICG level at time t. The process model is as:

$$
n_t = n_{t-1} \exp(a + b l n_{t-1} + e_t) \tag{3}
$$

where a and b are constants, and e_t is a random variable distributed as $e_t N(0, \sigma_1^2)$. The random variables e_1, \ldots, e_n are assumed to be uncorrelated. On the logarithmic scale, the DTGM is a linear autoregressive time-series model of order 1 $[AR(1)$ process] defined as equation [4.](#page-277-0)

$$
y_t = y_{t-1} + a + by_{t-1} + e_t = a + cy_{t-1} + e_t \tag{4}
$$

where, $y_t = \ln n_t$ and $c = b + 1$. For statistical properties of DTGM, see [\[18\]](#page-284-4).

The model has a long history in density-dependence modeling see $[19]$ - $[21]$. A freguently seen alternative is a stochastic version of the Moran-Ricker model $[21]$, which uses n_{t-1} instead of ln n_{t-1} in the exponential function; in comparative data analysis studies, the Gompertz model has performed as well as the Moran-Ricker $\boxed{22}$. The probability distribution of n_{t-1} is a normal distribution with mean and variance that change as functions of time. If $-1 < c < 1$, the probability distribution of n_t eventually approaches a time-independent stationary distribution that is a normal distribution with a mean of $a/(1-c)$ and a variance of $\sigma_1^2/(1-c^2)$. The stationary distribution is the stochastic version of an equilibrium in the deterministic model, and is an important statistical manifestation of density dependence in the population growth model Dennis $[18]$. In equation 4, a is the intrinsic growth rate, b is the density-dependent influence $[18]$.

3.1. Mathematical and Computational Methodologies. The optimum linear filtering and estimation methods introduced by Kalman [\[31\]](#page-284-10) have been considered as one of the greatest achievements in estimation theory. Discrete-time linear statespace models and Kalman filtering (KF) have been employed since the 1960s, mostly in the control and signal processing areas. The KF has been extensively employed in many areas of estimation. The extensions and applications of discrete-time linear state-space models can be found in almost all disciplines $\boxed{20}$ - $\boxed{28}$. In this study, KF has been used to estimate the time-varying parameter of the DTGM. KF is a recursive estimator to estimate the time-varying parameters. If $a = 0$ in Eq. [\(4\)](#page-277-0), n_t takes the case counts observed until t and $y_t = lnn_t$. Then the equation

$$
y_t = cy_{t-1} + e_t \tag{5}
$$

is acquired. In the case that the parameter c in Eq. $[5]$ is time-varying and presumed as random walk process, that is . Then state-space model,

$$
y_t = c_t y_{t-1} + e_t \tag{6}
$$

$$
c_t = c_{t-1} + w_t \tag{7}
$$

is obtained and w_t is distributed as $w_t N(0, \sigma_2^2)$. The random variables w_1, w_2, \ldots, w_n are assumed to be uncorrelated. Here, the state variable is unobservable, timevarying, and can be estimated through AKF (explanation regarding AKF is given in the Appendix section). If this time-varying parameter is estimated using on-line AKF, the ICG level in times $t+1,t+2, \ldots$ can be estimated via this online-estimated parameter. When the models given in equations (6) and (7) are compared with the state space model given in the Appendix, the following equations are obtained.

$$
x_t = c_t, F_t = 1, G_t = 1, H_t = y_{t-1}, R_t = \sigma_1, Q_t = \sigma_2 \tag{8}
$$

3.2. Application of DTGM. Details of the experimental setup, and how the data were collected can be found in $\boxed{23}$. Data is given in Table $\boxed{1}$. Since this study deals with the collected data, here only a very brief discussion regarding the experiments is given in order to put more emphasis on the mathematical representation, along with parameter estimation. In the experiments, bolus injections of the optical contrast agent ICG were administered to the rat through the tail vein. The measurements were collected by placing the probe normal to the tumor surface and probing the whole tissue including plasma. After injection, ICG rapidly and completely binds to albumin, after which the kinetics of ICG are governed by the temporal dynamics of albumin in and between the vascular compartment and the EES.

3.3. Estimation (AKF) Algorithm. The steps of the AKF algorithm using to estimate the parameter in DTGM are as follows. The code is written in Matlab program for the estimation algorithm.

Step 1. Initial values $\hat{c}_0 = 0.9, P_0 = 1, R_t = \sigma_1 = \text{std}(y_t), t = 1, 2, \ldots, n, Q_t =$ $\sigma_2 = 0.01, t = 1, 2, \ldots, n, \alpha = 1.0001$

- Step 2. $\hat{c}_{t|t-1} = \hat{c}_{t-1}$ Predicted (a priori) state estimate
- Step 3. $P_{t|-1} = \alpha (P_{t-1}t-1} + \sigma_2)$ Predicted (a priori) estimate covariance
- Step 4. $K_t = P_{t|t-1} y_{t-1} (y_{t-1} P_{t|t-1} y_{t-1} + \sigma_1)^{-1}$ Optimal Kalman gain
- Step 5. $P_{t=} = [I K_t y_{t-1}] P_{tt-1}$ Updated (a posteriori) estimate covariance
- Step 6. $\hat{c}_t = \hat{c}_{t|t-1} + K_t (y_t y_{t-1}\hat{c}_{t|t-1})$ Updated (a posteriori) state estimate

In the experiment, the ICG concentration in the lump space, i.e. EES and plasma, was monitored for 500 seconds. According to the estimation results obtained by using the ICG level in DTGM, the MSE, MAPE, R^2 and values are calculated (see Table $\overline{2}$). These calculated values indicate that the compatibility of the model with real data is quite high. This tells us estimating the ICG level via $DTGM$ is a reliable method. Since estimation using the $AR(1)$ stochastic process does not require any other model assumption. As for AKF, utilizing only the observation in time and the preceding estimation is the most advantageous aspect of this method.

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Model	MSE	$\mid R^2$	MAPE
DTGM		0.0001 0.9973 5.7385	
Compartment Models 0.0004 0.9826 19.2059			

TABLE 2. Calculated R^2 , MSE, MAPE.

Figure $\boxed{1}$ depicts the observed ICG concentration and the model fit obtained through the use of DTGM. Figure 2 depicts the observed ICG concentration and the model fit obtained through the use of compartment models $[9]$. Figure [3](#page-281-0) depicts the observed ICG concentration and the model fit obtained through the use of compartment models and DTGM. It is clearly seen that the DTGM mathematical model provides a rather good fit to the observations, which indicates the correctness of the model.

Figure 1. DTGM: Observed ICG concentration and the model fit.

4. CONCLUSION

In this study, we introduced a DTGM representing the metabolic elimination and transfer of ICG between compartments in rat tumors, and presented a method for the quantitative analysis of experimentally obtained ICG concentration data. This will be useful in the analysis of tumor cell behavior patterns in cancerous tissues. In this study, ICG concentration data have been estimated online using DTGM and AKF. The ICG concentration data is modeled with DTGM, and the time-varying parameters of the obtained $AR(1)$ stochastic time series are estimated 1176 $\,$ L. ÖZBEK $\,$

Figure 2. Compartment Models: Observed ICG concentration and the model fit.

Figure 3. Compartment Models and DTGM: Observed ICG concentration and the model fit.

by the on-line AKF. The estimation by the acquired data shows that employing the DTGM model and the AKF in terms of MSE, MAPE, and R^2 provide efficient analysis for modeling the ICG concentration data. It is proposed that using the DTGM and the AKF will be appropriate. It is quite a simple method to model the ICG concentration time series data with the time-varying parameter $AR(1)$ stochastic process and to estimate the time-varying parameter with the online AKF. When the results obtained from the compartment model used in the reference $[9]$ are compared with the results obtained with the DTGM, the DTGM offers better results according to MSE, MAPE and R^2 criteria. The DTGM and the AKF compartment model require less numerical processing compared to the EKF, and DTGM is a simpler model. In the literature, the EKF is used for such problems. As far as we know no other method has been used before.

Appendix

State-Space Model and Adaptive Kalman Filter (AKF)

Let us consider a general discrete-time stochastic system represented by the state and measurement models given as:

$$
x_{t+1} = F_t x_t + G_t w_t
$$

$$
y_t = H_t x_t + v_t
$$

where x_t is an $n \times 1$ system vector, y_t is an $m \times 1$ observation vector, F_t is an $n \times n$ system matrix, H_t is an $m \times n$ matrix, w_t an $n \times 1$ vector of zero mean white noise sequence and v_t is an $m \times 1$ measurement error vector assumed to be a zero mean white sequence uncorrelated with the w_t sequence. The covariance matrices w_t and w_t are defined by $w_t \sim N(0, Q_t)$, $v_t \sim N(0, R_t)$. The filtering problem is the problem of determining the best estimate of its x_t condition, given its observations $Y_t = (y_0, y_1, \ldots, y_t)$ [14−20]. When $Y_t = (y_0, y_1, \ldots, y_t)$ observations are given, the estimation of state x_t with

$$
\hat{x}_t = E(x_t \mid y_0, y_1, \dots, y_t) = E(x_t \mid Y_t)
$$

and the covariance matrix of the error with

$$
P_{t|t} = E [(x_t - \hat{x}_{t|}) (x_t - \hat{x}_{tt})' | Y_t]
$$

when $Y_{t-1} = (y_0, y_1, \ldots, y_{t-1})$ observations are given, the estimation of state x_t with $\hat{x}_{t|t-1} = E(x_t | y_0, y_1, \dots, y_{t-1}) = E(x_t | Y_{t-1})$

and the covariance matrix of the error are shown with

$$
P_{t|t-1} = E\left[(x_t - \hat{x}_{t|-1}) (x_t - \hat{x}_{t|t-1})' | Y_{t-1} \right].
$$

Let the initial state be assumed to have a normal distribution in the form of $x_0 \sim N(\bar{x}_0, P_0).$

The optimum update equations for KF are,

$$
\hat{x}_{t|-1} = F_{t-1}\hat{x}_{t-1}
$$

$$
P_{t|t-1} = F_{t-1}P_{t-4t-1}F'_{t-1} + G_{t-1}Q_{t-1}G'_{t-1}
$$

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$$
K_t = P_{t|t-1} H'_t (H_t P_{t|t-1} H'_t + R_t)^{-1}
$$

$$
P_{t|t} = [I - K_t H_t] P_{t|t-1}
$$

$$
\hat{x}_t = \hat{x}_{t|t-1} + K_t (y_t - H_t \hat{x}_{t|t-1})
$$

In the above equations, $\hat{X}_{t|t-1}$ is the a priori estimation and \hat{X}_t is the a posteriori estimation of x_t . Also, $P_{t|t-1}$ and $P_{t|t}$ are the covariance of a priori and a posteriori estimations respectively $\boxed{24}$ - $\boxed{33}$. In some cases, divergence problems may ocur in the KF due to the incorrect installation of the model. In order to eliminate divergence in the KF, adaptive methods are used $[5]$, $[32]$, $[33]$. One of these is the use of the forgetting factor. A forgetting factor is proposed by $[32]$.

$$
P_{t|t-1} = \alpha \left(F_{t-1} P_{t-1|-1} F'_{t-1} + G_{t-1} Q_{t-1} G'_{t-1} \right)
$$

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STUDY AND SUPPRESSION OF SINGULARITIES IN WAVE-TYPE EVOLUTION EQUATIONS ON NON-CONVEX DOMAINS WITH CRACKS

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ABSTRACT. One of the objectives of this paper is to establish the exact controllability for wave-type evolution equations on non-convex and/or cracked domains with non-concurrent support crack lines. Admittedly, we know that according to the work of Grisvard P., in domains with corners or cracks, the formulas of integrations by parts are subject to geometric conditions: the lines of cracks or their supports must be concurrent. In this paper, we have established the exact controllability for the wave equation in a domain with cracks without these additional geometric conditions.

1. INTRODUCTION

The presence of a crack in equipment (especially under pressure) requires, for obvious safety reasons, to know precisely its degree of harmfulness. When this crack propagates, under cyclic loading, it is important to evaluate and to quickly control the evolution of this degree of harmfulness and more concretely the residual life of the cracked structure.

In the works of the pioneers and precursors, not least Kondratiev [1], Grisvard [2], Moussaoui [3] and Niane [4], the control and removal of singularities were established in domains with corners or cracks.

Indeed, when these cracks propagate, under cyclic loading, it is important to evaluate and to quickly control the evolution of this degree of harmfulness and more concretely the residual life of the cracked structure. Thin plates and shells are widely used in aeronautics. Due to the significant stresses to which the structure

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of an aircraft is subjected in flight, for example, the appearance of small cracks is inevitable. Depending on the situation, these cracks are more or less dangerous; thus, certain cracks do not propagate, on the other hand, others present a certain risk.The risks alluded to earlier must, consequently, be curbed. So, once a crack has been detected, it is important to know if it can be dangerous or not? The safety of persons and that of the goods involved means repairing the work first and foremost. Notwithstanding, repairing all the cracks won't be necessary as, if the crack is not dangerous, it is no good repairing as it will be costly.

Accordingly, it is important to figure out whether or not the crack is dangerous, and whether it can be spread. Apart from extreme cases (very small or very large cracks), this diagnosis is not easy to make because even a small crack can spread brutally. It is very clear that the accuracy of this diagnosis is very important.

More recently, Seck [5], Bayili [6], taking inspiration from the exact controllability in Lipschitzian domains by Costabel [7], Niane [8] and Lions [9, 10], established results of exact controllability of the wave equation in non-regular Sobolev spaces. But, in all these works, the domains admit a crack or a corner or even cracks with condition of control: the lines of cracks are concurrent (or the supports of the lines of cracks are concurrent).

In this paper, without making additional assumptions and conditions on the crack lines and their supports, an exact controllability result was established for wave equation.

2. Reminders of Fundamental Results

2.1. Problem position. We denote by Ω an open polygonal uncracked, non convex and bounded of \mathbb{R}^2 and for $T > 0$, we denote by $Q_T = \Omega \times]0, T[$.

Let Γ the boundary of Ω , $\nu(x)$ the external unit normal at all points x (apart from the vertices) of Γ and Σ_T the lateral border of the cylinder Q_T .

Γ is the union of a finite number of closed line segments; the corresponding open segments are denoted Γ_j , $0 \leq j \leq N$ and S_{ij} the end common to Γ_j and Γ_i if it exists. We denote by ω_{ij} the measure of the angle made by Γ_j and Γ_i in S_{ij} towards the interior of $Ω$.

We denote by ν_i the unit normal vector outside Γ_i and τ_j the unit vector tangent to Γ_j and directed towards the vertex S_i . For x_0 any point of \mathbb{R}^2 , we consider the function $m(x) = x - x_0$ and a partition of the border as follows:

$$
\Gamma_0 = \{ x \in \Gamma; m(x) \cdot \nu \ge 0 \}, \quad \Gamma_0^* = \{ x \in \Gamma; m(x) \cdot \nu < 0 \},
$$

and

$$
\Sigma_0^* = \Gamma_0^* \times]0, T[.
$$

Let $\Vert . \Vert$ be the Euclidean norm in \mathbb{R}^2 and introduce the following constants

$$
R_0 = R(x_0) = \max_{x \in \overline{\Omega}} ||x - x_0||, \text{ and } T_0 = 2R(x_0).
$$

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Let also $f \in L^2(\Omega)$ and $y \in H_0^1(\Omega)$ be the unique solution of the homogeneous Dirichlet problem

$$
(P1)\begin{cases}\n-\triangle y = f, \\
y_{|\Gamma} = 0.\n\end{cases} (1)
$$

In the space $H = L^2(\Omega)$, we consider A the operator defined by:

$$
D(A) = \left\{ y \in H_0^1(\Omega); -\triangle y \in L^2(\Omega) \right\},\
$$

$$
\forall A \in D(A), Ay = -\triangle y.
$$

A: is a compact positive inverse self-adjoint operator see Brezis [11] and Hormander [12].

y is solution of $(\mathbf{1})(P1) \Longrightarrow y \in D(A)$.

Let $m + 1$ be the number of non-convex angles of the ∂Ω boundary of the domain Ω having $m + 1$ vertices $(S_i)_{0 \leq i \leq m}$.

It has been proved in Niane [4] that if $\bar{\omega}$ is an arbitrarily small part of Ω not meeting any vertex of cracks, there exist regular functions $(g_i)_{1 \leq i \leq m}$ with compact support in $\bar{\omega}$ such that for all $f \in L^2(\Omega)$, if $(\lambda_i)_{1 \leq i \leq m}$ are the coefficients of singularities of the problem $(P1)$ then the problem

$$
(P2)\begin{cases}\n-\Delta \tilde{y} = f + u, \\
\tilde{y}_{|\Gamma} = 0.\n\end{cases}
$$
\n(2)

admits a solution $\tilde{y} \in H^2(\Omega)$, with $u = -\sum_{i=0}^m \lambda_i h_i$, $\lambda_i = \int_{\Omega} f w_i dx$ where w_i the singular functions Cf. Grisvard [2] and $\langle g_i, w_j \rangle = \delta_{ij}$ Moussaoui [3] and Niane [4].

Let for $i \in \{0, ..., m\}$, (r_i, θ_i) represent the polar coordinates of a point M of Ω relatively to the vertex S_i with $r_i = || \overrightarrow{S_i M} ||$ Gilbert [6].

Remark 1. The singular functions w_i are harmonic

$$
\begin{cases}\n-\triangle w_i = 0 \, \text{sur } \Omega, \\
\omega_i = 0 \, \partial\Omega \setminus \{x_i\}.\n\end{cases}
$$

2.2. Internal control of the homogeneous waves equation on a non-convex domain. Let y: solution of the following homogeneous wave equation

$$
(EOH): \begin{cases} y'' - \Delta y = 0 & \text{in } Q_T, \\ y = 0, & \text{in } \Sigma_T, \\ y(0) = y_0 \ y'(0) = y_1 & \text{in } \Omega. \end{cases}
$$
 (3)

Figure 1. Non-convex cracked domain

$$
(EOH) \Longleftrightarrow (EOH)' \begin{cases} -\Delta y = -y'' & \text{in } Q_T, \\ y = 0, & \text{in } \Sigma_T, \\ y(0) = y_0 \ y'(0) = y_1 & \text{in } \Omega. \end{cases}
$$
 (4)

Let $(y_o, y_1) \in D(A) \times H_0^1(\Omega) \Longrightarrow$ the solution y of the equation (EOH) [\(3\)](#page-287-0) verified $y \in C(O, T; D(A)) \cap C^1(0, T; H_0^1(\Omega)) \cap C^2(0, T; L^2(\Omega)).$

In addition, in Grisvard [2], the solution can be decomposed as folloow: $y = y_R + \sum_{i=1}^m \lambda_i(t) S_i(t)$ with: $\lambda_i(t) = \int_{\Omega} (-y'')w_i(t)dt$ and $S_i(t) = r^{\alpha_i} sin(\alpha_i \theta_i)$ with α_i : the singularity exponent defined by $\alpha_i = \frac{\pi}{w_i}$, w_i : the aperture angle at the vertex S_i . As in the first part, we can, for any $t > 0$, add an internal check $u(t) = -\sum_{i=1}^{m} \lambda_i(t)g_i(t)$

of such that if \hat{y} is the regularized solution of the equation

$$
\begin{cases}\n-\Delta \tilde{y} = -\tilde{y}'' + u(t) & \text{in } Q_T, \\
\tilde{y} = 0 & \text{in } \Sigma_T, \\
\tilde{y}(0) = \tilde{y}_0, \quad \tilde{y}'(0) = \tilde{y}_1 & \text{in } \Omega.\n\end{cases}
$$
\n(5)

then $\hat{y} \in H^2(\Omega)$.

In fact, $\hat{y} = 0$ on the edge Σ_T , the solution $\hat{y} \in C(0,T; H^2(\Omega) \cap H_0^1(\Omega)).$ Let \tilde{V} be a subspace of $H^1(\Omega)$ of admissible solutions for the problem $(EOH)'$ defined by

$$
\tilde{V} = \{ \hat{y} \in H^1(\Omega) / \hat{y}_{|\Sigma_T} = 0 \}.
$$
\n(6)

For continuity, let us state the following proposition:

Proposition 1. The problem $(EOH)'$ (A) admits an unique solution \hat{y} in the space \tilde{V} and there exist a constant $C_T > 0$ such that

$$
||\hat{y}||_{C(0,T;H_0^1(\Omega))} \le C_T \left[||\tilde{y_0}||_{H_0^1(\Omega)} + ||\tilde{y_1}||_{L^2(\Omega)}\right]^{\frac{1}{2}}.
$$
\n(7)

Proof. Let A be the unbounded operator of $L^2(\Omega)$ previously defined. According to Spectral Theory and by Fourier transform, A is diagonalizable and there exists a countable Hilbertian basis of $L^2(\Omega)$ made up of eigenvectors $(z_k)_{k \in \mathbb{N}^*} \subset D(A)$ such that the sequence of eigenvalues $(\lambda_k)_{k\geq 1}$ of associated eigenvalues verify: $(\lambda_k) \nearrow +\infty$ and $\lambda_1 > 0$.

$$
z_k \in H_0^1(\Omega), -\Delta z_k = \lambda_k z_k \tag{8}
$$

The family $Z = (z_k)_{k \geq 1}$ Hilbert base of $L^2(\Omega)$ ie $\hat{y} \in L^2(\Omega) \implies \hat{y} = \sum_{k \geq 1} \hat{y}_k z_k$ with $\hat{y}_k = \langle \hat{y}, z_k \rangle_{L^2(\Omega)} \text{ and } \sum_{k \ge 1} z_k^2 \langle \hat{z} | \infty. \text{ What's more } ||\hat{y}||_{L^2(\Omega)} = \left(\sum_{k=1}^{+\infty} \hat{y}_k^2 \right)^{\frac{1}{2}}.$ $\sqrt{2}$ $\sqrt{\frac{1}{2}}$

$$
\hat{y} \in H_0^1(\Omega) \Longleftrightarrow \hat{y} = \sum_{k \ge 1} \hat{y}_k z_k, \sum_{k \ge 1} \lambda_k \hat{y}_k^2 < +\infty \text{ and } ||\hat{y}||_{H_0^1(\Omega)} = \left(\sum_{k \ge 1} \lambda_k \hat{y}_k^2\right) . \tag{9}
$$

So, if \hat{y} is solution of $(EOH)'$ [\(4\)](#page-288-0) then

$$
\begin{cases}\n\hat{y}(t,x) = \sum_{k\geq 1} \hat{y}_k(t) z_k(x), \\
\hat{y}_{0k}(x) = \sum_{k\geq 1} \hat{y}_{0k} z_k(x), \\
\hat{y}_{1k}(x) = \sum_{k\geq 1} \hat{y}_{1k} z_k(x), \\
\sum_{k\geq 1} (\hat{y}_k''(t) - \lambda_k \hat{y}_k(t)) z_k(x) = 0.\n\end{cases}
$$
\n(10)

We multiply the relation (10) by the eigenfunctions z_k and integrate on the cylinder Q_T

$$
\begin{cases}\n\hat{y}_k''(t) - \lambda_k \hat{y}_k(t) = 0, \\
\hat{y}_k(0) = \hat{y}_{0k}, \\
\hat{y}_k(1) = \hat{y}_{1k}.\n\end{cases}
$$
\n(11)

And, for all $k \geq 1$, the solution of $\overline{11}$ (see Lions [9, 10]) is under the form

$$
\hat{y}_k(t) = \hat{y}_{0k} \cos(\sqrt{\lambda_k}t) + \hat{y}_{1k} \frac{\sin(\sqrt{\lambda_k}t)}{\sqrt{\lambda_k}},
$$
\n(12)

So

$$
\hat{y}_k(t,x) = \sum_{k \ge 1} \left(\hat{y}_{0k} \cos(\sqrt{\lambda_k}t) + \hat{y}_{1k} \frac{\sin(\sqrt{\lambda_k}t)}{\sqrt{\lambda_k}} \right) z_k(x). \tag{13}
$$

Assume

$$
\begin{array}{rcl} ||\hat{y}||_{C(0,T;H_0^1(\Omega))}^2 & = & sup_{t \in [0,T]} ||\hat{y}(t,.)||_{H_0^1(\Omega)}^2 \\ & = & sup_{t \in [0,T]} \sum_{k \ge 1} |\lambda_k| |\hat{y}_k(t)|^2 \end{array}
$$

=⇒

$$
||\hat{y}||_{C(0,T;H_0^1(\Omega))}^2 \le \sum_{k\ge 1} |\lambda_k| \sup_{t\in[0,T]} |\hat{y}_k(t)|^2
$$
\n(14)

Based on the relationship [\(13\)](#page-290-1)

$$
\begin{array}{rcl} ||\hat{y}||_{C(0,T;H_0^1(\Omega))}^2 & \leq & 2. \sum_{k>1} \lambda_k \left[\hat{y}_{0k}^2 + \frac{\hat{y}_{1k}^2}{\lambda_k} \right] \\ & \leq & 2. \sum_{k>1} \lambda_k \left[\hat{y}_{0k}^2 + \hat{y}_{1k}^2 \right] \end{array}
$$

let's remember that

$$
\hat{y}_0 \in H_0^1(\Omega) \Longleftrightarrow \begin{cases} \hat{y}_0(x) = \sum_{k>1} \hat{y}_{0k} z_k(x), \\ \sum_{k>1} \lambda_k \tilde{y}_{0k}^2 < +\infty \text{ and} \\ ||\hat{y}_0||_{H_0^1(\Omega)}^2 = \sum_{k>1} \lambda_k \hat{y}_{0k}^2. \end{cases}
$$
\n(15)

and

$$
\hat{y}_1 \in L^2(\Omega) \Longleftrightarrow \begin{cases} \hat{y}_1(x) = \sum_{k>1} \hat{y}_{1k} z_k(x), \\ \sum_{k>1} \lambda_k \hat{y}_{1k}^2 < +\infty \text{ and} \\ ||\hat{y}_1||_{H_0^1(\Omega)}^2 = \sum_{k>1} \lambda_k \hat{y}_{1k}^2. \end{cases}
$$
(16)

Therefore, we get that $\hat{y} \in C(0, T; H_0^1(\Omega))$ [\(1\)](#page-289-1) with

$$
||\hat{y}||_{C(0,T;H_0^1(\Omega))} \leq C_T \left(||\hat{y}_0||_{H_0^1(\Omega)} + ||\hat{y}_1||_{L^2(\Omega)} \right)
$$
\n
$$
\Box
$$

2.3. Application to the removal of singularities. Let \tilde{y} regularized solution of the equation

$$
(EOS): \begin{cases} \tilde{y}'' - \Delta \tilde{y} + \sum_{i=1}^{m} g_i \int_{\Omega} (\tilde{y}'') w_i dx = 0 & \text{in } Q_T, \\ \tilde{y} = 0, & \text{in } \Sigma_T, \\ \tilde{y}(0) = \tilde{y}_0, \ \tilde{y}'(0) = \tilde{y}_1 & \text{in } \Omega. \end{cases}
$$
(18)

It will then be a matter of showing that the solution \tilde{y} of the equation (EOH) [\(3\)](#page-287-0) is in $C(0, T; H^2(\Omega) \cap H_0^1(\Omega))$?

In general, it was proved in Grisvard [2] that the following wave equation

$$
(EOS)_2: \begin{cases} \varphi'' - \Delta \varphi = f \in L^1(0, T; H_0^1(\Omega)), \\ \varphi = 0 & \text{in } \Sigma_T, \\ \varphi(0) = \varphi_0, \ \varphi'(0) = \varphi_1 & \text{in } \Omega, \\ (\varphi_0, \varphi_1) \in D(A) \times D(A^{\frac{1}{2}}). \end{cases}
$$
(19)

admit a solution $\varphi \in C(0,T;D(A)) \cap C^1(0,T;H^1(\Omega)) \cap C(0,T;L^2(\Omega))$ and that this solution verifies the inequality:

$$
||\varphi||_{C(0,T;D(A))} \le K\left(||\varphi_0||_{D(A)} + ||\varphi_1||_{D(A^{\frac{1}{2}})} + ||f||_{H_0^1(\Omega)}\right),\tag{20}
$$

called continuous dependence of the solution compared to the initial conditions and to the second member.

Let us apply this Grisvard result to the equation (EOS) (18) ; For this consider for

 $\zeta \in C^2(0,T;L^2(\Omega)), \tilde{y} = y(\zeta)$ is solution of the equation

$$
(EOS)_3: \begin{cases} \tilde{y}''(\zeta) - \Delta \tilde{y}(\zeta) = -\sum_{i=1}^m g_i \int_{\Omega} \zeta'' w_i dx & \text{in } Q_T, \\ \tilde{y}(\zeta) = 0 & \text{in } \Sigma_T, \end{cases}
$$

$$
\tilde{y}(\zeta)(0) = \tilde{y}(\zeta_0), \ \tilde{y}(\zeta)'(0) = \tilde{y}(\zeta_1) \text{ in } \Omega.
$$

 $(EOS)_3$ and the inequality [\(20\)](#page-291-1) implies a priori that $y(\zeta) \in C(0,T;D(A))$ and that

$$
||y(\zeta)||_{C(0,T;D(A))} \le K_1 \left(||\sum_{i=1}^m g_i \int_{\Omega} \zeta''||_{L^1(\Omega)} \right).
$$
 (21)

Consider the application $\Lambda : \zeta \mapsto y(\zeta)$; Let us show that Λ is contracting ?

Let $\zeta_1 \longmapsto y(\zeta_1), \ \zeta_2 \longmapsto y(\zeta_2)$ and $\zeta = \zeta_1 - \zeta_2 \longmapsto y(\zeta)$. Applying it to the equation $(EOS)_3$ we get

$$
(EOS)_4: \begin{cases} \tilde{y}''(\zeta) - \Delta \tilde{y}(\zeta) = -\sum_{i=1}^m g_i \left(\int_{\Omega} \zeta'' w_i dx \right) & \text{in } Q_T, \\ \tilde{y}(\zeta) = 0 & \text{in } \Sigma_T. \end{cases}
$$

More $y(\zeta_1)(0) = y(\zeta_2)(0) = 0$ $(y(\zeta_1)$ and $y(\zeta_2)$ have the same initial conditions as y_0 and y_1).

From inequality (21) we deduce

$$
||y(\zeta)||_{C(0,T;D(A))} \leq K_1 \left(||\sum_{i=1}^m g_i \int_{\Omega} \zeta'' w_i||_{L^1(\Omega)} dx\right), \tag{22}
$$

$$
\leq K_2 \left(\sum_{i=1}^m ||g_i||.||w_i||.|| \int_{\Omega} \zeta'' dx|| \right), \tag{23}
$$

$$
\leq K_3\left(\sum_{i=1}^m||g_i||.||w_i||||\zeta'||_{L^1(\Omega)}\cdot mes(\Omega)\right),\tag{24}
$$

$$
\leq K_4 \left(\sum_{i=1}^m ||g_i||_{H_0^1(\Omega)} ||w_i||_{L^1(\Omega)} ||\zeta||_{L^1(\Omega)} \right), \qquad (25)
$$

$$
\leq K ||\zeta|| \qquad (26)
$$

$$
\leq K_5 \|\zeta\|_{L^1(\Omega)}.\tag{26}
$$

With the constant $K_5 = \sum_{i=1}^m ||g_i||_{H_0^1(\Omega)} ||w_i||_{L^1(\Omega)}$. Let us show that $0 < K_5 < 1$ ie Λ is contracting?

We know that the dual singular functions are such that:

 $w_i = r^{-\alpha_i} sin(\alpha_i \theta_i) \eta_i + \zeta_i$ with $\alpha_i = \frac{\pi}{\omega_i}$ and $\omega_i > \pi$, η_i atruncation function in the neighborhood of the vertices of x_i and $\zeta_i \in H_0^1(\Omega)$ for all $i \in \{0, ..., m\}$.

The application Λ is Lipschitzian, let us show that it is contracting ie $0 < K_5 < 1$?

$$
||w_i|| = ||r^{-\alpha_i} \sin(\alpha_i \theta_i)\eta_i + \zeta_i||,
$$
\n(27)

$$
\leq \frac{1}{r^{\alpha_i}}||\sin(\alpha_i \theta_i)\eta_i|| + ||\zeta_i||,\tag{28}
$$

$$
\leq \frac{1}{r^{\alpha_0}} + ||\zeta||. \tag{29}
$$

where $\alpha_0 = \min_{i \in \{1, ..., m\}} \alpha_i$ thus $\frac{1}{r^{\alpha_i}} < \frac{1}{r^{\alpha_0}}$.

The functions $(g_i)_{1\leq i\leq m}$ are compact support on $\bar{\omega}$ which is compact, so there is $g_0 = max_{1 \leq i \leq m} g_i$ on $\bar{\omega}$ such that $||g_i|| \leq ||g_0||$ for all i. Therefore

$$
\sum_{i=1}^{m} ||g_i|| ||w_i|| \le m^2 ||g_0|| \frac{1}{r^{\alpha_0}} + C_1 \text{ with } C_1 > 1 \text{ a constant.}
$$

As a result,

$$
0 < K_5 < m^2 ||g_0|| \frac{1}{r^{\alpha_0}} + C_1.
$$

A sufficient condition for Λ to be contracting is that

$$
m^2||g_0||\frac{1}{r^{\alpha_0}} + C_1 < 1 \Longleftrightarrow r \ge e^{\frac{1}{\alpha_0}\log\left(\frac{m^2||g_0||}{1-C_1}\right)}.\tag{30}
$$

Remember that

$$
r = ||\vec{S_i M}|| = ||x - x_i||
$$

ie $M \neq S_i$, $\forall i \in \{1, ..., m\}$ on $\bar{\omega}$.

Hence if M is far from the top of the crack ie $r \gg 1$ the application Λ is contracting. Thereby,

$$
||y(\zeta)||_{C(0,T;D(A))} \le K_5 ||\zeta||_{C^2(0,T;L^2(\Omega))}
$$
\n(31)

Therefore, if $\boxed{30}$ holds then the application Λ is contracting and according to the Fixed Point Theorem $y(\zeta) = y(\zeta_1) - y(\zeta_2) = 0$ and y being continuous so ζ is unique.

Hence the equation

$$
(EOS)_3: \begin{cases} \tilde{y}''(\zeta) - \Delta \tilde{y}(\zeta) = -\sum_{i=1}^m g_i \int_{\Omega} (\tilde{y}''(\zeta)) w_i dx & \text{in } Q_T, \\ \tilde{y}(\zeta) = 0 & \text{in } \Sigma_T, \\ \tilde{y}(\zeta)(0) = \tilde{y}(\zeta)_0, \ \tilde{y}(\zeta)'(0) = \tilde{y}(\zeta)_1 & \text{in } \Omega. \end{cases}
$$
(32)

admits a unique solution $\tilde{y} \in C(0,T; H^2(\Omega) \cap H_0^1(\Omega))$.

Proposition 2. The solution \tilde{y} is the regularized solution, therefore the singularity coefficient $\tilde{\lambda}$ associated with it is null.

Proof. Let $\tilde{\lambda}$ the singularity coefficient associated with \tilde{y} . By definition,

$$
\tilde{\lambda} = \int_{\Omega} u(t)w_i dx, \tag{33}
$$

$$
= \int_{\Omega} \left(-\sum_{i=1}^{m} g_i \int_{\Omega} \tilde{y}^{"} w_i dx \right) . w_i dx, \tag{34}
$$

$$
= - \int_{\Omega} \int_{\Omega} \sum_{i=1}^{m} g_i \tilde{y}^{\prime \prime} w_i . w_i dx dx, \qquad (35)
$$

$$
= -\int_{\Omega} \int_{\Omega} \sum_{i=1}^{m} \langle g_i, w_i \rangle \Delta \tilde{y} w_i dx dx.
$$
 (36)

so $\langle g_i, w_i \rangle = 1 \Longrightarrow$

$$
\tilde{\lambda} = -\int_{\Omega} \int_{\Omega} \sum_{i=1}^{m} \Delta \tilde{y} w_i dx dx, \qquad (37)
$$

$$
= - \int_{\Omega} \int_{\Omega} \sum_{i=1}^{m} \tilde{y} \Delta w_i dx dx \qquad (38)
$$

because $\tilde{y}_{\vert \Sigma_T} = 0$.

We also know that the dual singular functions are harmonic ie $\Delta w_i = 0$ hence $\tilde{\lambda} = 0$

Remark 2. The corrective term or internal control $u(t)$ depends on \tilde{y}'' , therefore \tilde{y} .

3. Use in the Implementation of the Hum Method

3.1. Preliminaries. Let y solution of wave equation

$$
(EOH): \begin{cases} \n y'' - \Delta y = 0 & \text{in } Q_T, \\ \n y = 0 & \text{in } \Sigma_T, \\ \ny(0) = y_0, \ y'(0) = y_1 & \text{in } \Omega. \n\end{cases}
$$

For initial data y_0 and y_1 belonging respectively to $H_0^1(\Omega)$ and $L^2(\Omega)$. Let also be the energy of (EOH) defined by

$$
E_0 = \frac{1}{2} (||y_0||_{H_0^1(\Omega)} + ||y_1||_{H_0^1(\Omega)})
$$
\n(39)

We know that in a polygonal domain with corner, $(x - x_0) \cdot \nu \frac{\partial \varphi}{\partial \nu}$ is not always a square integrable on the edge near of corner. Grisvard [2] got around this difficulty by imposing drastic geometric conditions. And, in Seck [5] this result has been generalized with less constraints in non-regular Sobolev spaces. Also Niane [4] have shown, without geometric conditions, the exact controllability of the wave equation by combining a boundary control and an internal control on a small part whose support is in the vicinity of a vertex crack.

3.2. Implementation of the HUM method. Let us return to the equation of the following waves

$$
(EOS)_5: \begin{cases} \tilde{\varphi}'' - \Delta \tilde{\varphi} = u(\tilde{\varphi}) & \text{in } Q_T, \\ \tilde{\varphi} = 0 & \text{in } \Sigma_T, \\ \tilde{\varphi}(0) = \tilde{\varphi}_0, \ \tilde{\varphi}'(0) = \tilde{\varphi}_1 & \text{in } \Omega. \end{cases}
$$
(40)

From the above, with $u(\tilde{\varphi}) = \sum_{i=1}^{m} g_i \left(\int_{\Omega} \varphi'' \underline{w}_i dx \right)$, the solution $\tilde{\varphi} \in H^2(\Omega)$. Indeed, we multiply the equation $(EOS)_5$ (40) by $m\nabla\tilde{y}$ and integrate by parts:

$$
\int_{Q_T} (\tilde{\varphi}'' - \Delta \tilde{\varphi}) m \nabla \tilde{y} dx dt = \int_{Q_T} m \nabla \tilde{y} u(\tilde{\varphi}) dx dt,
$$
\n(41)

$$
= -m\nabla \tilde{y} \sum_{i=1}^{m} g_i \left(\int_{Q_T} \varphi'' w_i dx dt \right). \tag{42}
$$

Assume

$$
I = \int_{Q_T} (\tilde{\varphi}'' - \Delta \tilde{\varphi}) m \nabla \tilde{y} dx dt
$$
\n(43)

$$
= \underbrace{\int_{Q_T} \tilde{\varphi}'' m \nabla \tilde{y} dx dt}_{I_1} - \underbrace{\int_{Q_T} \Delta \tilde{\varphi} m \nabla \tilde{y} dx dt}_{I_2} \tag{44}
$$

3.3. Some integrations by parts.

3.3.1. First Term I_1 .

$$
I_1 = \int_{Q_T} \tilde{\varphi}'' m \nabla \tilde{y} dx dt,
$$

\n
$$
= \int_0^T \int_{\Omega} \tilde{\varphi}'' m(x) \nabla \tilde{y} dx dt,
$$

\n
$$
= \int_{\Omega} \tilde{\varphi}' m(x) \nabla \tilde{y} dx \vert_0^T - \int_0^T \int_{\Omega} \frac{\partial \tilde{\varphi}}{\partial t} m_k \frac{\partial^2 \tilde{y}}{\partial t \partial x_k} dt dx,
$$

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$$
= \int_{\Omega} \tilde{\varphi}' m(x) \nabla \tilde{y} dx|_{0}^{T} - \int_{0}^{T} \left(\int_{\Omega} \left(\frac{\partial \tilde{\varphi}}{\partial t} m_{k} \frac{\partial}{\partial t} (\frac{\partial \tilde{y}}{\partial x_{k}}) \right) dx \right) dt,
$$

\n
$$
= \int_{\Omega} \tilde{\varphi}' m(x) \frac{\partial \tilde{y}}{\partial x_{k}} dx|_{0}^{T} - \int_{0}^{T} \int_{\Omega} \left(\frac{\partial \tilde{\varphi}}{\partial t} m_{k} \frac{\partial^{2} \tilde{y}}{\partial t \partial x_{k}} \right) dx dt,
$$

\n
$$
= \int_{\Omega} \frac{\partial \tilde{\varphi}}{\partial t} m_{k} \frac{\partial \tilde{y}}{\partial x_{k}} dx|_{0}^{T} - \int_{0}^{T} \int_{\Omega} m_{k} \frac{\partial^{2} \tilde{\varphi}}{\partial t \partial x_{k}} \frac{\partial \tilde{y}}{\partial t} dx dt.
$$

Noting that: $N = 2$, $divm = \sum_{k=1}^{2} \frac{\partial m_k}{\partial x_k} = 2$ and applying Green again we have:

$$
I_1 = \int_{\Omega} \frac{\partial \tilde{\varphi}}{\partial t} m_k \frac{\partial \tilde{y}}{\partial x_k} dx \vert_0^T - \int_0^T \left[- \int_{\Omega} \frac{\partial m_k}{\partial x_k} \frac{\partial \tilde{\varphi}}{\partial t} \frac{\partial \tilde{y}}{\partial t} dx + \underbrace{\int_{\partial \Omega} m_k \tilde{\varphi}_k \frac{\partial \tilde{y}}{\partial t} d\sigma}_{=0} \right] dt,
$$

\n
$$
= \int_{\Omega} \frac{\partial \tilde{\varphi}}{\partial t} m_k \frac{\partial \tilde{y}}{\partial x_k} dx \vert_0^T + \int_{Q_T} \operatorname{div} m \frac{\partial \tilde{\varphi}}{\partial t} \frac{\partial \tilde{y}}{\partial t} dx dt.
$$

3.3.2. Second Term $\mathcal{I}_2.$

$$
I_2 = \int_{Q_T} \Delta \tilde{\varphi} m \nabla \tilde{y} dx dt,
$$

\n
$$
= \int_0^T \int_{\Omega} \Delta \tilde{\varphi} m \nabla \tilde{y} dx dt,
$$

\n
$$
= \int_0^T \left[\int_{\Omega} \nabla \tilde{\varphi} \cdot \nabla (m_k \frac{\partial \tilde{\varphi}}{\partial x_k}) dx - \int_{\partial \Omega} \frac{\partial \tilde{\varphi}}{\partial n} m_k \frac{\partial \tilde{y}}{\partial x_k} d\sigma \right] dt,
$$

\n
$$
= \int_0^T \left[\int_{\Omega} \frac{\partial \tilde{\varphi}}{\partial x_i} (m_k \frac{\partial \tilde{y}}{\partial x_k}) dx - \int_{\partial \Omega} \frac{\partial \tilde{\varphi}}{\partial n} m_k \frac{\partial \tilde{y}}{\partial x_k} d\sigma \right] dt,
$$

\n
$$
= \int_0^T \left[\int_{\Omega} \frac{\partial m_k}{\partial x_i} \cdot \frac{\partial \tilde{\varphi}}{\partial x_i} \frac{\tilde{y}}{\partial x_k} dx + \underbrace{\int_{\Omega} \frac{\partial \tilde{\varphi}}{\partial x_i} m_k \frac{\partial^2 \tilde{y}}{\partial x_i \partial x_k} dx}_{J} - \int_{\partial \Omega} \frac{\partial \tilde{\varphi}}{\partial n} m_k \frac{\partial \tilde{y}}{\partial x_k} d\sigma \right] dt.
$$

Let's study the integral $J\!$:

$$
J = \int_{\Omega} \frac{\partial \tilde{\varphi}}{\partial x_i} m_k \frac{\partial^2 \tilde{y}}{\partial x_i \partial x_k} dx,
$$

\n
$$
= \int_{\Omega} m_k \frac{\partial \tilde{\varphi}}{\partial x_i} \frac{\partial^2 \tilde{y}}{\partial x_i \partial x_k} dx,
$$

\n
$$
= \int_{\Omega} \frac{\partial}{\partial x_k} (m_k \frac{\tilde{\varphi}}{\partial x_i}) \frac{\partial \tilde{y}}{\partial x_i} dx - \int_{\partial \Omega} m_k n_k \frac{\partial \tilde{\varphi}}{\partial x_i} \frac{\partial \tilde{y}}{\partial x_i} d\sigma.
$$

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By grouping together we get:

$$
I_2 = \int_0^T \left[\int_{\Omega} \frac{\partial m_k}{\partial x_i} \cdot \frac{\partial \tilde{\varphi}}{\partial x_i} \frac{\tilde{y}}{\partial x_k} dx + \int_{\Omega} \frac{\partial}{\partial x_k} (m_k \frac{\tilde{\varphi}}{\partial x_i}) \frac{\partial \tilde{y}}{\partial x_i} dx \right. \\ - \left. \int_{\partial \Omega} m_k n_k \frac{\partial \tilde{\varphi}}{\partial x_i} \frac{\partial \tilde{y}}{\partial x_i} d\sigma - \int_{\partial \Omega} \frac{\partial \tilde{\varphi}}{\partial n} m_k \frac{\partial \tilde{y}}{\partial x_k} d\sigma \right] dt, \\ = \int_{Q_T} \frac{\partial m_k}{\partial x_i} \cdot \frac{\partial \tilde{\varphi}}{\partial x_i} \frac{\tilde{y}}{\partial x_k} dx dt + \int_{Q_T} \frac{\partial}{\partial x_k} (m_k \frac{\tilde{\varphi}}{\partial x_i}) \frac{\partial \tilde{y}}{\partial x_i} dx dt \right. \\ - \left. \int_{\Sigma_T} m.n \frac{\partial \tilde{\varphi}}{\partial x_i} \frac{\partial \tilde{y}}{\partial x_i} d\sigma dt - \int_{\Sigma_T} \frac{\partial \tilde{\varphi}}{\partial n} m_k \frac{\partial \tilde{y}}{\partial x_k} d\sigma dt \right]
$$

Back to $I = I_1 + I_2$ [\(43\)](#page-295-1) and (45):

$$
I = \int_{\Omega} \frac{\partial \tilde{\varphi}}{\partial t} m_k \frac{\partial \tilde{y}}{\partial x_k} dx \Big|_0^T + \int_{Q_T} \operatorname{div} m \frac{\partial \tilde{\varphi}}{\partial t} \frac{\partial \tilde{y}}{\partial t} dx dt + \int_{Q_T} \frac{\partial m_k}{\partial x_i} \cdot \frac{\partial \tilde{\varphi}}{\partial x_i} \cdot \frac{\tilde{y}}{\partial x_k} dx dt
$$

+
$$
\int_{Q_T} \frac{\partial}{\partial x_k} (m_k \frac{\tilde{\varphi}}{\partial x_i}) \frac{\partial \tilde{y}}{\partial x_i} dx dt - \int_{\Sigma_T} m_n m \frac{\partial \tilde{\varphi}}{\partial x_i} \frac{\partial \tilde{y}}{\partial x_i} d\sigma dt - \int_{\Sigma_T} \frac{\partial \tilde{\varphi}}{\partial n} m_k . n_k \frac{\partial \tilde{y}}{\partial x_k} d\sigma dt.
$$

Also

$$
L = \frac{1}{2} \int_{\Sigma_T} \frac{\partial \tilde{\varphi}}{\partial n} m.n \nabla \tilde{y} d\sigma dt,
$$

$$
= \frac{1}{2} \int_{\Sigma_T} \left(\frac{\partial \tilde{\varphi}}{\partial n} \right) m.n d\sigma dt.
$$

The two equations have the same initial and boundary conditions. Let's study ${\cal L}$?

$$
\frac{\partial \tilde{\varphi}}{\partial x_i} = \frac{\partial \tilde{\varphi}}{\partial n} . n_i + \frac{\partial \tilde{\varphi}}{\partial \tau_i},
$$

Decomposition according to the normal and the tangential. However

$$
\frac{\partial \tilde{\varphi}}{\partial n_i} = \frac{\partial \tilde{\varphi}}{\partial n} n_i \Rightarrow \sum_i \frac{\partial \tilde{\varphi}}{\partial n_i} = \sum_i \frac{\partial \tilde{\varphi}}{\partial n} n_i \Rightarrow \nabla \tilde{\varphi} = \frac{\partial \tilde{\varphi}}{\partial n} n.
$$

So we deduce that:

$$
I = \int_{\Omega} \frac{\partial \tilde{\varphi}}{\partial t} m_k \frac{\partial \tilde{y}}{\partial x_k} dx|_0^T + \int_{Q_T} \operatorname{div} m \frac{\partial \tilde{\varphi}}{\partial t} \frac{\partial \tilde{y}}{\partial t} dx dt + \int_{Q_T} \frac{\partial m_k}{\partial x_i} \cdot \frac{\partial \tilde{\varphi}}{\partial x_i} \frac{\tilde{y}}{\partial x_k} dx dt + \n\int_{Q_T} \frac{\partial}{\partial x_k} (m_k \frac{\tilde{\varphi}}{\partial x_i}) \frac{\partial \tilde{y}}{\partial x_i} dx dt - \int_{\Sigma_T} m.n \frac{\partial \tilde{\varphi}}{\partial x_i} \frac{\partial \tilde{y}}{\partial x_i} d\sigma dt - \frac{1}{2} \int_{\Sigma_T} (\frac{\partial \tilde{\varphi}}{\partial n})^2 m.n d\sigma dt.
$$
 (45)

3.3.3. Third Term I3.

$$
I_3 = \int_{Q_T} \left\{ -m \nabla \tilde{y} \sum_{i=1}^m g_i \left(\int_{\Omega} \varphi'' w_i dx \right) dx dt \right\},
$$

\n
$$
= \int_0^T \int_{\Omega} \left\{ -m \nabla \tilde{y} \sum_{i=1}^m g_i \left(\int_{\Omega} \varphi'' w_i dx \right) dx dt \right\},
$$

\n
$$
= -\int_0^T \int_{\Omega} \int_{\Omega} m \nabla \tilde{y} \sum_{i=1}^m \langle g_i, w_i \rangle \tilde{\varphi}'' dx dx dt,
$$

\n
$$
= -\int_0^T \int_{\Omega} \int_{\Omega} m \nabla \tilde{y} \tilde{\varphi}'' dx dx dt,
$$

\n
$$
= -\int_0^T \int_{\Omega} \int_{\Omega} m \nabla \tilde{y} \nabla \tilde{\varphi} dx dx dt - \underbrace{\int_{\Sigma_T} \int_{\partial \Omega} m \tilde{y} \tilde{\varphi}(\sigma) d\sigma dt}_{=0},
$$

\n
$$
= -\int_0^T \int_{\Omega} \int_{\Omega} m \nabla \tilde{y} \nabla \tilde{\varphi} dx dx dt.
$$

Let's recap $I = I_3(46) \Longleftrightarrow$ $I = I_3(46) \Longleftrightarrow$ $I = I_3(46) \Longleftrightarrow$

 $\mathbf{0}$

Ω

Ω

$$
\int_{\Omega} \frac{\partial \tilde{\varphi}}{\partial t} m_k \frac{\partial \tilde{y}}{\partial x_k} dx|_0^T + \int_{Q_T} \operatorname{div} m \frac{\partial \tilde{\varphi}}{\partial t} \frac{\partial \tilde{y}}{\partial t} dx dt + \int_{Q_T} \frac{\partial m_k}{\partial x_i} \cdot \frac{\partial \tilde{\varphi}}{\partial x_i} \frac{\tilde{y}}{\partial x_k} dx dt \n+ \int_{Q_T} \frac{\partial}{\partial x_k} (m_k \frac{\tilde{\varphi}}{\partial x_i}) \frac{\partial \tilde{y}}{\partial x_i} dx dt - \int_{\Sigma_T} m \cdot n \frac{\partial \tilde{\varphi}}{\partial x_i} \frac{\partial \tilde{y}}{\partial x_i} d\sigma dt - \frac{1}{2} \int_{\Sigma_T} (\frac{\partial \tilde{\varphi}}{\partial n})^2 m \cdot n d\sigma dt \n= - \int_0^T \int_{\Omega} \int_{\Omega} m \nabla \tilde{y} \nabla \tilde{\varphi} dx dx dt \qquad (46)
$$

$$
\frac{1}{2} \int_{\Sigma_{T}} (\frac{\partial \tilde{\varphi}}{\partial n})^2 m. n d\sigma dt = \int_{\Omega} \frac{\partial \tilde{\varphi}}{\partial t} m_k \frac{\partial \tilde{y}}{\partial x_k} dx|_{0}^{T} + \int_{Q_{T}} \operatorname{div} m \frac{\partial \tilde{\varphi}}{\partial t} \frac{\partial \tilde{y}}{\partial t} dx dt \n+ \int_{Q_{T}} \frac{\partial m_k}{\partial x_i} \cdot \frac{\partial \tilde{\varphi}}{\partial x_i} \frac{\tilde{y}}{\partial x_k} dx dt + \int_{Q_{T}} \frac{\partial}{\partial x_k} (m_k \frac{\tilde{\varphi}}{\partial x_i}) \frac{\partial \tilde{y}}{\partial x_i} dx dt \n- \int_{\Sigma_{T}} m. n \frac{\partial \tilde{\varphi}}{\partial x_i} \frac{\partial \tilde{y}}{\partial x_i} d\sigma dt + \int_{0}^{T} \int_{\Omega} \int_{\Omega} m \nabla \tilde{y} \nabla \tilde{\varphi} dx dx dt.
$$
\n(47)

3.4. Getting started with the HUM method. For $x_0 \in \mathbb{R}^2$, assume

$$
\Sigma_T^{0*} = \Gamma_0^* \times]0, T[, \Sigma_T^{1*} = \Gamma_0 \times]0, T[.
$$

Let $\Vert . \Vert$ the Euclidean in \mathbb{R}^2 and introduce the following constants.

$$
R_0 = R(x_0) = \max_{x \in \overline{\Omega}} ||x - x_0||, T_0 = 2R(x_0).
$$

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Let us define in the same way the energies (see Lions [9, 10]) associated respectively with the systems $(EOS)_5$, $\boxed{40}$ and (EOH) :

$$
E(t, \tilde{\varphi}_0, \tilde{\varphi}_1) = \frac{1}{2} \left[\int_{\Omega} ||\nabla \tilde{\varphi}(t)||_{\mathbb{R}^2}^2 dx + \int_{\Omega} (\frac{\partial \tilde{\varphi}}{\partial t}(t))^2 dx \right],
$$

$$
E(t, \tilde{y}_0, \tilde{y}_1) = \frac{1}{2} \left[\int_{\Omega} ||\nabla \tilde{y}(t)||_{\mathbb{R}^2}^2 dx + \int_{\Omega} (\frac{\partial \tilde{y}}{\partial t}(t))^2 dx \right].
$$

3.4.1. Direct Inequality. Back to the relationship [\(46\)](#page-298-0)

$$
\frac{1}{2} \int_{\Sigma_{T}} (\frac{\partial \tilde{\varphi}}{\partial n})^{2} m. n d\sigma dt = \int_{\Omega} \frac{\partial \tilde{\varphi}}{\partial t} m_{k} \frac{\partial \tilde{y}}{\partial x_{k}} dx|_{0}^{T} + \int_{Q_{T}} \operatorname{div} m \frac{\partial \tilde{\varphi}}{\partial t} \frac{\partial \tilde{y}}{\partial t} dx dt + \n+ \int_{Q_{T}} \frac{\partial m_{k}}{\partial x_{i}} \cdot \frac{\partial \tilde{\varphi}}{\partial x_{i}} \frac{\tilde{y}}{\partial x_{k}} dx dt + \int_{Q_{T}} \frac{\partial}{\partial x_{k}} (m_{k} \frac{\tilde{\varphi}}{\partial x_{i}}) \frac{\partial \tilde{y}}{\partial x_{i}} dx dt + \n- \int_{\Sigma_{T}} m. n \frac{\partial \tilde{\varphi}}{\partial x_{i}} \frac{\partial \tilde{y}}{\partial x_{i}} d\sigma dt + \int_{0}^{T} \int_{\Omega} \int_{\Omega} m \nabla \tilde{y} \nabla \tilde{\varphi} dx dx dt \qquad (48)
$$

$$
\iff \frac{1}{2} \int_{\Sigma_T} (\frac{\partial \tilde{\varphi}}{\partial n})^2 m. n d\sigma dt - \int_0^T \int_{\Omega} \int_{\Omega} m \nabla \tilde{y} \nabla \tilde{\varphi} dx dx dt = \int_{\Omega} \frac{\partial \tilde{\varphi}}{\partial t} m_k \frac{\partial \tilde{y}}{\partial x_k} dx \Big|_0^T + \int_{Q_T} \dim \frac{\partial \tilde{\varphi}}{\partial t} \frac{\partial \tilde{y}}{\partial t} dx dt + \int_{Q_T} \frac{\partial m_k}{\partial x_i} \frac{\partial \tilde{\varphi}}{\partial x_i} \frac{\tilde{y}}{\partial x_k} dx dt + \int_{Q_T} \frac{\partial}{\partial x_k} (m_k \frac{\tilde{\varphi}}{\partial x_i}) \frac{\partial \tilde{y}}{\partial x_i} dx dt - \int_{\Sigma_T} m. n \frac{\partial \tilde{\varphi}}{\partial x_i} \frac{\partial \tilde{y}}{\partial x_i} d\sigma dt.
$$
 (49)

We know that:

$$
\int_{\Omega} \frac{\partial \tilde{\varphi}}{\partial t} m_k \frac{\partial \tilde{y}}{\partial x_k} dx \vert_0^T \le R_0. \int_{\Omega} \frac{\partial \tilde{\varphi}}{\partial t} \frac{\partial \tilde{y}}{\partial x_k} dx \vert_0^T \tag{50}
$$

and noticing that: $|ab| \leq \frac{1}{2}(a^2 + b^2)$ we have:

$$
\int_{\Omega} \frac{\partial \tilde{\varphi}}{\partial t} \frac{\partial \tilde{y}}{\partial x_k} dx \le \frac{1}{2} \int_{Q_T} \left[(\frac{\partial \tilde{\varphi}}{\partial t})^2 + (\frac{\partial \tilde{y}}{\partial x_k})^2 \right] dx.
$$
 (51)

Therefore:

$$
\int_{\Omega} \frac{\partial \tilde{\varphi}}{\partial t} \frac{\partial \tilde{y}}{\partial x_k} dx \vert_0^T \le \frac{T}{2} \int_{Q_T} \left[(\frac{\partial \tilde{\varphi}}{\partial t})^2 + (\frac{\partial \tilde{y}}{\partial x_k})^2 \right] dx dt.
$$
 (52)

Assume

$$
\Sigma_T = \Sigma_T^{0*} \cup \Sigma_T^{1*}, \qquad M_1 = \max_{1 \le i,k \le 2} \max_{x \in \bar{\mathcal{B}}_i} |\frac{\partial x_k}{\partial x_i}(x)|.
$$

Consider an open ball \mathcal{B}_i which does not meet any crack vertex ie $h \equiv \eta h$ (In the general case we can recover the domain Ω by a finite union of \mathcal{B}_i ie $\Omega = \cup_{i=1}^{Ns} \mathcal{B}_i$.

$$
\int_{Q_T} \frac{\partial m_k}{\partial x_i} \cdot \frac{\partial \tilde{\varphi}}{\partial x_k} \frac{\partial \tilde{y}}{\partial x_k} dx dt = \int_{B_i \times]0,T[} \frac{\partial m_k}{\partial x_i} \cdot \frac{\partial \tilde{\varphi}}{\partial x_i} \frac{\partial \tilde{y}}{\partial x_k} dx dt,
$$

$$
\leq \frac{M_1}{2} \int_{B_i \times]0,T[} \left[(\frac{\partial \tilde{\varphi}}{\partial x_i})^2 + (\frac{\partial \tilde{y}}{\partial x_k})^2 \right] dxdt,
$$

$$
\leq \frac{M_1}{2} \int_{B_i \times]0,T[} \left[||\nabla \tilde{\varphi}||_{L^2(\mathbb{R}^2)}^2 + ||\nabla \tilde{y}||_{L^2(\mathbb{R})^2} \right] dxdt
$$
(53)

Relationships $\left(\overline{50}\right),\,\left(\overline{51}\right),$ and $\left(\overline{53}\right),$ we deduce: \mathbf{v}

$$
\frac{1}{2} \int_{\Sigma_{T}} (\frac{\partial \tilde{\varphi}}{\partial n})^{2} m. n d\sigma dt - \sum_{i=1}^{N_{s}} \int_{0}^{T} \int_{B_{i}} \int_{B_{i}} m \nabla \tilde{y} \nabla \tilde{\varphi} dx dx dt \le R_{0} \cdot \sum_{i=1}^{N_{s}} \int_{B_{i}} \frac{\partial \tilde{\varphi}}{\partial t} \frac{\partial \tilde{y}}{\partial x_{k}} dx|_{0}^{T} + \frac{T}{2} \int_{Q_{T}} \left[(\frac{\partial \tilde{\varphi}}{\partial t})^{2} + (\frac{\partial \tilde{y}}{\partial x_{k}})^{2} \right] dx dt + \sum_{i=1}^{N_{s}} \frac{M_{1}}{2} \int_{B_{i} \times]0, T[} \left[||\nabla \tilde{\varphi}||_{L^{2}(\mathbb{R}^{2})}^{2} + ||\nabla \tilde{y}||_{L^{2}(\mathbb{R})^{2}} \right] dx dt \tag{54}
$$

$$
\Rightarrow \frac{1}{2} \int_{\Sigma_{T}} (\frac{\partial \tilde{\varphi}}{\partial n})^{2} m. n d\sigma dt \leq R_{0} \cdot \sum_{i=1}^{N_{s}} \int_{B_{i}} \frac{\partial \tilde{\varphi}}{\partial t} \frac{\partial \tilde{y}}{\partial x_{k}} dx|_{0}^{T} \n+ \frac{T}{2} \int_{Q_{T}} \left[(\frac{\partial \tilde{\varphi}}{\partial t})^{2} + (\frac{\partial \tilde{y}}{\partial x_{k}})^{2} \right] dx dt \n+ \sum_{i=1}^{N_{s}} \frac{M_{1}}{2} \int_{B_{i} \times]0, T[} \left[||\nabla \tilde{\varphi}||_{L^{2}(\mathbb{R}^{2})}^{2} + ||\nabla \tilde{y}||_{L^{2}(\mathbb{R})^{2}} \right] dx dt \n+ \sum_{i=1}^{N_{s}} \int_{0}^{T} \int_{B_{i}} \int_{B_{i}} m \nabla \tilde{y} \nabla \tilde{\varphi} dx dx dt.
$$
\n(55)

Therefore

$$
\sum_{i=1}^{N_s} \int_0^T \int_{B_i} \int_{B_i} m \nabla \tilde{y} \nabla \tilde{\varphi} dx dx dt \leq R_0 \sum_{i=1}^{N_s} \int_0^T \int_{B_i} \int_{B_i} \nabla \tilde{y} \nabla \tilde{\varphi} dx dx dt
$$

\n
$$
\leq R_0 \sum_{i=1}^{N_s} \int_0^T \int_{B_i} \int_{B_i} \left[(\frac{\partial \tilde{\varphi}}{\partial x_k})^2 + (\frac{\partial \tilde{y}}{\partial t})^2 \right],
$$

\n
$$
\leq R_0 \sum_{i=1}^{N_s} \int_{B_i} \int_0^T \int_{B_i} \left[(\frac{\partial \tilde{\varphi}}{\partial x_k})^2 + (\frac{\partial \tilde{y}}{\partial t})^2 \right],
$$

\n
$$
\leq R_0 \sum_{i=1}^{N_s} mes(B_i) \left[E(t, \tilde{\varphi}_0, \tilde{\varphi}_1) + E(t, \tilde{y}_0, \tilde{y}_1) \right].
$$
 (56)

Starting from the fact that the energy associated with the wave equation is constant, we obtain:

$$
\sum_{i=1}^{N_s} \int_0^T \int_{B_i} \int_{B_i} m \nabla \tilde{y} \nabla \tilde{\varphi} dx dx dt \le 2R_0 \sum_{i=1}^{N_s} mes(B_i)E(t, \tilde{\varphi}_0, \tilde{\varphi}_1). \tag{57}
$$

So the relation [\(53\)](#page-299-2) implies

$$
\frac{1}{2} \int_{\Sigma_{T}} (\frac{\partial \tilde{\varphi}}{\partial n})^{2} m. n d\sigma dt \leq R_{0} \cdot \sum_{i=1}^{N_{s}} \int_{B_{i}} \frac{\partial \tilde{\varphi}}{\partial t} \frac{\partial \tilde{y}}{\partial x} dx|_{0}^{T} \n+ \frac{T}{2} \sum_{i=1}^{N_{s}} \int_{B_{i}} \left[(\frac{\partial \tilde{\varphi}}{\partial t})^{2} + (\frac{\partial \tilde{y}}{\partial x_{k}})^{2} \right] dx dt \n+ \sum_{i=1}^{N_{s}} \frac{M_{1}}{2} \int_{B_{i} \times]0, T[} \left[||\nabla \tilde{\varphi}||_{L^{2}(\mathbb{R}^{2})}^{2} + ||\nabla \tilde{y}||_{L^{2}(\mathbb{R})^{2}} \right] dx dt \n+ 2R_{0} \sum_{i=1}^{N_{s}} mes(B_{i}) E(t, \tilde{\varphi}_{0}, \tilde{\varphi}_{1}), \qquad (58) \n+ \frac{1}{2} \int_{\Sigma_{T}} (\frac{\partial \tilde{\varphi}}{\partial n})^{2} m. n d\sigma dt \leq \frac{T.R_{0}}{2} \cdot \sum_{i=1}^{N_{s}} \int_{B_{i}} \left[(\frac{\partial \tilde{\varphi}}{\partial t})^{2} + (\frac{\partial \tilde{y}}{\partial x_{k}})^{2} \right] dx dt \n+ \frac{T}{2} \sum_{i=1}^{N_{s}} \int_{B_{i}} \left[(\frac{\partial \tilde{\varphi}}{\partial t})^{2} + (\frac{\partial \tilde{y}}{\partial x_{k}})^{2} \right] dx dt \n+ \sum_{i=1}^{N_{s}} \frac{M_{1}}{2} \int_{B_{i} \times]0, T[} \left[||\nabla \tilde{\varphi}||_{L^{2}(\mathbb{R}^{2})}^{2} + ||\nabla \tilde{y}||_{L^{2}(\mathbb{R}^{2})} \right] dx dt
$$

+
$$
2R_0 \sum_{i=1}^{N_s} mes(B_i)E(t, \tilde{\varphi}_0, \tilde{\varphi}_1),
$$
 (59)

$$
\frac{1}{2} \int_{\Sigma_{T}} \left(\frac{\partial \tilde{\varphi}}{\partial n}\right)^{2} m. n d\sigma dt \leq \frac{T.R_{0} + 1}{2} \cdot \sum_{i=1}^{N_{s}} \int_{B_{i}} \left[\left(\frac{\partial \tilde{\varphi}}{\partial t}\right)^{2} + \left(\|\nabla \tilde{\varphi}\|_{L^{2}(\mathbb{R}^{2})}\right)^{2} \right] dx dt +
$$
\n
$$
\sum_{i=1}^{N_{s}} \frac{M_{1}}{2} \int_{B_{i} \times]0, T[} \left[\left(\frac{\partial \tilde{y}}{\partial t}\right)^{2} + \left(\|\nabla \tilde{y}\|_{L^{2}(\mathbb{R}^{2})}\right)^{2} \right] dx dt + 2R_{0} \sum_{i=1}^{N_{s}} mes(B_{i}) E(t, \tilde{\varphi}_{0}, \tilde{\varphi}_{1}), (60)
$$
\n
$$
\frac{1}{2} \int_{\Sigma_{T}} \left(\frac{\partial \tilde{\varphi}}{\partial n}\right)^{2} m. n d\sigma dt \leq \left(\frac{T.R_{0} + 1}{2} + \frac{M_{1}}{2} + 2R_{0} \sum_{i=1}^{N_{s}} mes(B_{i}) \right) E(t, \tilde{\varphi}_{0}, \tilde{\varphi}_{1})
$$
\n
$$
\leq C_{T}^{0}(\Omega) E(t, \tilde{\varphi}_{0}, \tilde{\varphi}_{1}). \tag{61}
$$

From

$$
\frac{1}{2}||\frac{\partial \tilde{\varphi}}{\partial n}||_{L^{2}(\Omega)}^{2} \leq C_{T}^{0}(\Omega)E(t, \tilde{\varphi}_{0}, \tilde{\varphi}_{1}),
$$
\n(62)

where

re
$$
C_T^0(\Omega) = \left(\frac{T.R_0 + 1}{2} + \frac{M_1}{2} + 2R_0 \sum_{i=1}^{N_s} mes(B_i)\right).
$$
 (63)

3.4.2. Inverse Inequality. Feedback on the relationship $\sqrt{47}$

$$
\frac{1}{2} \int_{\Sigma_{T}} (\frac{\partial \tilde{\varphi}}{\partial n})^{2} m. n d\sigma dt = \int_{\Omega} \frac{\partial \tilde{\varphi}}{\partial t} m_{k} \frac{\partial \tilde{y}}{\partial x_{k}} dx|_{0}^{T} + \int_{Q_{T}} \operatorname{div} m \frac{\partial \tilde{\varphi}}{\partial t} \frac{\partial \tilde{y}}{\partial t} dx dt \n+ \int_{Q_{T}} \frac{\partial m_{k}}{\partial x_{i}} \frac{\partial \tilde{\varphi}}{\partial x_{i}} dx dt + \int_{Q_{T}} \frac{\partial}{\partial x_{k}} (m_{k} \frac{\tilde{\varphi}}{\partial x_{i}}) \frac{\partial \tilde{y}}{\partial x_{i}} dx dt \n- \int_{\Sigma_{T}} m. n \frac{\partial \tilde{\varphi}}{\partial x_{i}} \frac{\partial \tilde{y}}{\partial x_{i}} d\sigma dt + \int_{0}^{T} \int_{\Omega} \int_{\Omega} m \nabla \tilde{y} \nabla \tilde{\varphi} dx dx dt. (64)
$$

$$
\int_{Q_T} \frac{\partial m_k}{\partial x_i} \cdot \frac{\partial \tilde{\varphi}}{\partial x_k} dx dt = \sum_{i=1}^{Ns} \int_{B_i} \int_0^T \frac{\partial m_k}{\partial x_i} \cdot \frac{\partial \tilde{\varphi}}{\partial x_i} \frac{\partial \tilde{y}}{\partial x_k} dx dt,
$$
\n
$$
\leq \frac{1}{2} \sum_{i=1}^{Ns} \int_{B_i} \int_0^T \frac{\partial m_k}{\partial x_i} \left[(\frac{\partial \tilde{\varphi}}{\partial x_i})^2 + (\frac{\partial \tilde{y}}{\partial x_k})^2 \right] dx dt,
$$
\n
$$
\leq \frac{1}{2} \left[||\nabla \tilde{\varphi}||_{\mathbb{R}^2}^2 + ||\nabla \tilde{y}||_{\mathbb{R}^2}^2 \right] \sum_{i=1}^{Ns} \int_{B_i} \int_0^T \frac{\partial m_k}{\partial x_i},
$$
\n
$$
\leq \frac{1}{2} \left[||\nabla \tilde{\varphi}||_{\mathbb{R}^2}^2 + ||\nabla \tilde{y}||_{\mathbb{R}^2}^2 \right] Ns. \int_0^T m(x) dt,
$$
\n
$$
\leq \frac{T}{2} \left[||\nabla \tilde{\varphi}||_{\mathbb{R}^2}^2 + ||\nabla \tilde{y}||_{\mathbb{R}^2}^2 \right] Ns.m(x),
$$
\n
$$
\leq \frac{T.R_0Ns}{2} \left[||\nabla \tilde{\varphi}||_{\mathbb{R}^2}^2 + ||\nabla \tilde{y}||_{\mathbb{R}^2}^2 \right].
$$

We deduce that:

$$
-\sum_{i=1}^{Ns} \int_{B_i} \int_0^T \frac{\partial m_k}{\partial x_i} \cdot \frac{\partial \tilde{\varphi}}{\partial x_i} \frac{\partial \tilde{y}}{\partial x_k} dx dt \ge -\frac{T.R_0Ns}{2} \left[||\nabla \tilde{\varphi}||^2_{\mathbb{R}^2} + ||\nabla \tilde{y}||^2_{\mathbb{R}^2} \right].
$$
 (65)

In addition, let us pose $M_2 = min_{x \in \bar{\Omega}} ||m(x)||_{\mathbb{R}^2}^2$:

$$
\int_{\Omega} \frac{\partial \tilde{\varphi}}{\partial t} m_k \frac{\partial \tilde{y}}{\partial x_k} dx \vert_0^T \ge M_2 \int_{\Omega} \frac{\partial \tilde{\varphi}}{\partial t} \frac{\partial \tilde{y}}{\partial x_k} dx \vert_0^T,
$$

therefore

$$
\int_{\Omega} \frac{\partial \tilde{\varphi}}{\partial t} \frac{\partial \tilde{y}}{\partial x_k} dx \vert_0^T dx \le \frac{1}{2} \int_{Q_T} \left[(\frac{\partial \tilde{\varphi}}{\partial t})^2 + (\frac{\partial \tilde{y}}{\partial x_k})^2 \right] dx dt \Rightarrow
$$

$$
-\frac{1}{2} \int_{Q_T} \left[(\frac{\partial \tilde{\varphi}}{\partial t})^2 + (\frac{\partial \tilde{y}}{\partial x_k})^2 \right] dx dt \le -\int_{\Omega} \frac{\partial \tilde{\varphi}}{\partial t} \frac{\partial \tilde{y}}{\partial x_k} dx \vert_0^T dx,
$$

So

$$
\int_{\Omega} \frac{\partial \tilde{\varphi}}{\partial t} m_k \frac{\partial \tilde{y}}{\partial x_k} dx \vert_0^T \ge -\frac{M_2.T}{2} \int_{Q_T} \left[(\frac{\partial \tilde{\varphi}}{\partial t})^2 + (\frac{\partial \tilde{y}}{\partial x_k})^2 \right] dx dt.
$$
 (66)

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Also, from the relation (65) we deduce

$$
-\sum_{i=1}^{N_s} \int_0^T \int_{B_i} \int_{B_i} m \nabla \tilde{y} \nabla \tilde{\varphi} dx dx dt \ge -2R_0 \sum_{i=1}^{N_s} mes(B_i)E(t, \tilde{\varphi}_0, \tilde{\varphi}_1). \tag{67}
$$

We also know that,

$$
\int_{Q_T} \frac{\partial}{\partial x_k} (m_k \frac{\partial \tilde{\varphi}}{\partial x_i}) \frac{\partial \tilde{y}}{\partial x_i} dx dt = \int_{Q_T} \frac{\partial m_k}{\partial x_k} \frac{\partial \tilde{\varphi}}{\partial x_i} \frac{\partial \tilde{y}}{\partial x_i} dx dt + \int_{Q_T} m_k \frac{\partial^2 \tilde{\varphi}}{\partial x_i^2} \frac{\partial \tilde{y}}{\partial x_i} dx dt
$$

$$
\int_{\Sigma_T} m.n \frac{\partial \tilde{\varphi}}{\partial x_i} \frac{\partial \tilde{y}}{\partial x_i} d\sigma dt = \underbrace{\int_{\Sigma_T^{0*}} m.n \frac{\partial \tilde{\varphi}}{\partial x_i} \frac{\partial \tilde{y}}{\partial x_i} d\sigma dt}_{m.n<0} + \underbrace{\int_{\Sigma_T^{1*}} m.n \frac{\partial \tilde{\varphi}}{\partial x_i} \frac{\partial \tilde{y}}{\partial x_i} d\sigma dt}_{m.n>0}
$$

By grouping and reducing simultaneously we have:

$$
\frac{1}{2}||\frac{\partial \tilde{\varphi}}{\partial n}||_{L^{2}(\Sigma_{T})} \geq -\frac{M_{2} \cdot T}{2} \int_{Q_{T}} \left[(\frac{\partial \tilde{\varphi}}{\partial t})^{2} + (\frac{\partial \tilde{y}}{\partial x_{k}})^{2} \right] dxdt \n- \frac{T \cdot R_{0} N s}{2} [||\nabla \tilde{\varphi}||_{\mathbb{R}^{2}}^{2} + ||\nabla \tilde{y}||_{\mathbb{R}^{2}}^{2}] \n- 2R_{0} \sum_{i=1}^{N_{s}} mes(B_{i}) E(t, \tilde{\varphi}_{0}, \tilde{\varphi}_{1}) + \int_{Q_{T}} divm \frac{\partial \tilde{\varphi}}{\partial t} \frac{\partial \tilde{y}}{\partial t} dxdt \n+ \int_{Q_{T}} \frac{\partial}{\partial x_{k}} (m_{k} \frac{\partial \tilde{\varphi}}{\partial x_{i}})^{\frac{\partial \tilde{y}}{\partial x_{i}}} dxdt - \int_{\Sigma_{T}} m.n \frac{\partial \tilde{\varphi}}{\partial x_{i}} \frac{\partial \tilde{y}}{\partial x_{i}} d\sigma dt \implies\n\frac{1}{2} ||\frac{\partial \tilde{\varphi}}{\partial n}||_{L^{2}(\Sigma_{T})} \geq \left(-\frac{M_{2} \cdot T}{2} - \frac{T \cdot R_{0} N s}{2} - 2R_{0} \sum_{i=1}^{N_{s}} mes(B_{i}) + 2 \cdot \frac{M_{2} \cdot T}{2} \right) E(t, \tilde{\varphi}_{0}, \tilde{\varphi}_{1}),
$$
(68)

$$
\geq \left(\frac{M_2.T}{2} - \frac{T.R_0Ns}{2} - 2R_0 \sum_{i=1}^{N_s} mes(B_i)\right) E(t, \tilde{\varphi}_0, \tilde{\varphi}_1). \tag{69}
$$

By posing

$$
C_T^1(\Omega) = \left(\frac{M_2.T}{2} - \frac{T.R_0Ns}{2} - 2R_0 \sum_{i=1}^{N_s} mes(B_i)\right),\tag{70}
$$

$$
\frac{1}{2}||\frac{\partial \tilde{\varphi}}{\partial n}||_{L^2(\Sigma_T)} \ge C_T^1(\Omega)E(t, \tilde{\varphi}_0, \tilde{\varphi}_1).
$$

3.5. Exact Controllability Result. Either the operator $\Lambda : H_0^1(\Omega) \times L^2(\Omega)$ Lions [9] defined by:

$$
\Lambda(\tilde{\varphi}_0, \tilde{\varphi}_1) = (\tilde{y}'(0), -\tilde{y}(0)).\tag{72}
$$

Indeed, we know that Grisvard [2]:

$$
\Lambda \in \mathcal{L}\left[H_0^1(\Omega) \times L^2(\Omega), H^{-1}(\Omega) \times L^2(\Omega)\right] \quad \text{and} \tag{73}
$$

$$
||\Lambda(\tilde{\varphi}_0, \tilde{\varphi}_1)||_{H^{-1}(\Omega)}^2 = ||\tilde{y}(0)||_{L^2(\Omega)}^2 + ||\tilde{y}'(0)||_{H^{-1}(\Omega)}^2.
$$
\n(74)

Considering $\tilde{\varphi}_n \in C(0,T;D(A)) \cap C^1(0,T;D(A^{\frac{1}{2}})) \cap C^2(0,T;L^2(\Omega))$, and also $\tilde{\varphi}_{0n} \in D(A), \tilde{\varphi}_{1n} \in D(A^{\frac{1}{2}}).$ Assume $\zeta_n = \left[u(\tilde{\varphi}) \right] \chi_{\bar{O}} = \left[\sum_{i=1}^m g_i \left(\int_{\Omega} \varphi'' w_i dx \right) \right] \chi_{\bar{O}}$ where O is an arbitrarily small part of the domain Ω not meeting any vertex of cracks.

Let $z_n \in C(0,T; H_0^2(\Omega)) \cap C^1(0,T; L^2(\Omega))$ solution of the following equation

$$
(EOS)_6: \begin{cases} z''_n - \Delta z_n = \zeta_n & \text{in } Q_T, \\ (z_n) \cdot \chi_{\bar{O}} = 0 & \text{on } \Sigma_T, \\ z_n(T) = z'_n(T) = 0 & \text{in } \Omega. \end{cases}
$$

So we have (3.5) (3.5) :

$$
\langle \Lambda(\tilde{\varphi}_{0n}, \tilde{\varphi}_{1n}), (z_{0n}, z_{1n}) \rangle = \langle z'_n(0), \tilde{\varphi}_{0n} \rangle - \langle z_n(0), \tilde{\varphi}_{1n} \rangle. \tag{75}
$$

By multiplying the equation $(EOS)_{6}$ by $\tilde{\varphi}_n$ and the equation (EOS) [\(18\)](#page-291-0) by z_n we get:

$$
-\int_{Q_T} (z_n'' - \Delta z_n)\tilde{\varphi}_n dxdt + \int_{Q_T} (\tilde{\varphi}_n'' - \Delta \tilde{\varphi}_n)z_n dxdt = \int_{Q_T} \sum_{i=1}^m ||g_i|| \left(\int_{\Omega} ||\tilde{\varphi}''||.||w_i|| dx \right) dt.
$$

In particular on the open O:

$$
-\int_{O\times]0,T[} (z_n'' - \Delta z_n)\tilde{\varphi}_n dxdt + \int_{O\times]0,T[} (\tilde{\varphi}_n'' - \Delta \tilde{\varphi}_n)z_n dxdt
$$

$$
= \int_{O\times]0,T[} \sum_{i=1}^m ||g_i|| \left(\int_O ||\tilde{\varphi}''||.||w_i||dx \right) dt,
$$

which is also written

$$
\int_{O\times]0,T[\sum_{i=1}^{m}||g_i|| \left(\int_{O} ||\tilde{\varphi}''||.||w_i||dx\right)dt = -\langle \tilde{\varphi}_n, z'_n \rangle \Big|_0^T + \langle \tilde{\varphi}'_n, z_n \rangle \Big|_0^T
$$

$$
- \underbrace{\int_{\partial O\times]0,T[\left(\frac{\partial \tilde{\varphi}_n}{\partial \nu}\right)^2 \tilde{\varphi}_n(\sigma) d\sigma dt}_{=0}
$$

$$
- \underbrace{\int_{\partial O\times]0,T[\left(\frac{\partial z_n}{\partial \nu}\right)^2 z_n(\sigma) d\sigma dt}_{=0}.
$$

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Therefore

$$
\begin{array}{rcl} \displaystyle\int_{O\times]0,T[\,\stackrel{\cdot }{i=1}} \displaystyle\sum_{i=1}^m ||g_i||\left(\int_O ||\tilde{\varphi}''||.||w_i||dx\right)dt&=&-\left<\tilde{\varphi}_n,z_n'\right>|\stackrel{\cdot }{0}+\left<\tilde{\varphi}_n',z_n\right>|\stackrel{\cdot T}{0},\\ &=&\left<\Lambda(\tilde{\varphi}_{0n},\tilde{\varphi}_{1n}),(\tilde{\varphi}_{0n},\tilde{\varphi}_{1n})\right>.
$$

Passing to the limit,

$$
\langle \Lambda(\tilde{\varphi}_0, \tilde{\varphi}_1), (\tilde{\varphi}_0, \tilde{\varphi}_1) \rangle = \int_{O \times]0,T[} \sum_{i=1}^m ||g_i|| \left(\int_O ||\tilde{\varphi}''||.||w_i|| dx \right) dt.
$$

But we also know that

$$
\int_{O\times]0,T[\sum_{i=1}^{m}||g_i||\left(\int_{O}||\tilde{\varphi}''||.||w_i||dx\right)dt = \int_{O\times]0,T[\sum_{i=1}^{m}\int_{O}||\tilde{\varphi}''||.||g_i||.||w_i||dxdt,\n\geq \int_{O\times]0,T[\prod_{i=1}^{m}\int_{O}\tilde{\varphi}''dx||.||dxdt,\n\geq 2m\left[\frac{1}{2}\int_{O}||\tilde{\varphi}'|||_{0}^{T}dx\right].
$$

By covering the domain Ω by a disjoint finite union of openings O_i ie $\Omega = \bigcup_{i=1}^m O_i$ and $O_i \cap O_j = \emptyset$ if $i \neq j$.

Consequently, we deduce that:

$$
\langle \Lambda(\varphi_0, \varphi_1), (\varphi_0, \varphi_1) \rangle \ge K_1(T - T_0)E_0. \tag{76}
$$

 Λ being linear, continuous and coercive on $H_0^1(\Omega) \times L^2(\Omega)$ for $T > T_0$, then according to a Classical Controllability Theorem, Λ is an isomorphism of $H_0^1(\Omega) \times L^2(\Omega)$ in $L^2(\Omega) \times H^{-1}(\Omega)$.

Let $(z_0, z_1) \in L^2(\Omega) \times H^{-1}(\Omega)$, the following equation

$$
\Lambda(\tilde{\varphi}_0, \tilde{\varphi}_1) = (z_1, -z_0),
$$

admit a unique solution $(\tilde{\varphi}_0, \tilde{\varphi}_1) \in H_0^1(\Omega) \times L^2(\Omega)$ for all $T > T_0$. Let us now consider $\tilde{\varphi}$ and z respective solutions of the equations $(EOS)_5$ and $(EOS)_6$ with as initial conditions:

$$
z_0 = \tilde{\varphi}_0,
$$

\n
$$
z_1 = \tilde{\varphi}_1,
$$

\n
$$
\zeta_n = \left(\sum_{i=1}^m g_i \left(\int_{\Omega} \tilde{\varphi}'' w_i dx\right)\right) \chi_O,
$$

\nand
\n
$$
\varphi = \begin{cases} \tilde{\varphi}.\chi_O & on \ \Sigma_T^{*0}, \\ 0 & on \ \Sigma_T^{*1}.\end{cases}
$$

By a uniqueness of solutions theorem, we deduce that: $\tilde{\varphi} = z$ so therefore $z(T) =$ $z'(T) = 0.$

Hence the result of exact controllability.

Remark 3. This result does not depend on any geometrical condition: consequently the crack lines may not be concurrent; and, the exact controllability result has been proven.

Figure 2. Non-convex domain with non-concurrent cracks

4. Conclusion and Perspectives

The presence of cracks, corners or angles in a mechanical device or materials always leads to the appearance of singularities. And, once the diagnosis of these cracks (desired or not) has been made, it is necessary to try to control them without major geomeric constraints.

One of the objectives that we set ourselves, within the framework of this research paper, was assess the exact controllability of the wave equation in the cracked domains without constraints on the cracks. If anything, the formulas of integrations by parts (formulas of Green in the fields with corners and/or cracks) could be done (to our knowledge) only if the lines of cracks or their support were concurrent.

Based on recent work by Dauge [13, 14], Dauge [15] and Costabel [16], we were able to establish, without additional assumptions on the nature of the cracks or their support, the exact controllability of the wave equation with more cracks. Consequently its results were obtained on a non-convex polygonal domain with non-concurrent crack lines. From the results obtained in this paper, certain questions naturally emerge. Our goal is to no longer have constraining geometric conditions ("Closer" to reality).

When it comes to the perspectives, we have a double goal that we plan on achieving in the near future. Firstly, generalize in higher dimension the results obtained in this paper. And,secondly, make numerical simulations to support its theoretical results.

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