

Functional Calculus

Concepts and Applications

Flora Duffy

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Analytical Applications on Some Hilbert Spaces

Fethi Soltani

Abstract

In this paper, we establish an uncertainty inequality for a Hilbert space H . The minimizer function associated with a bounded linear operator from H into a Hilbert space K is provided. We come up with some results regarding Hardy and Dirichlet spaces on the unit disk \mathbb{D} .

Keywords: Hilbert space, Hardy space, Dirichlet space, uncertainty inequality, minimizer function

1. Introduction

Hilbert spaces are the most important tools in the theories of partial differential equations, quantum mechanics, Fourier analysis, and ergodicity. Apart from the classical Euclidean spaces, examples of Hilbert spaces include spaces of square-integrable functions, spaces of sequences, Sobolev spaces consisting of generalized functions, and Hardy spaces of holomorphic functions. Saitoh et al. applied the theory of Hilbert spaces to the Tikhonov regularization problems [1, 2]. Matsuura et al. obtained the approximate solutions for bounded linear operator equations with the viewpoint of numerical solutions by computers [3, 4]. During the last years, the theory of Hilbert spaces has gained considerable interest in various fields of mathematical sciences [5–9]. We expect that the results of this paper will be useful when discussing (in Section 2) uncertainty inequality for Hilbert space H and minimizer function associated with a bounded linear operator T from H into a Hilbert space K . As applications, we consider Hardy and Dirichlet spaces as follows.

Let \mathbb{C} be the complex plane and $\mathbb{D} = \{z \in \mathbb{C} : |z| < 1\}$ the open unit disk. The Hardy space $H(\mathbb{D})$ is the set of all analytic functions f in the unit disk \mathbb{D} with the finite integral:

$$\int_0^{2\pi} |f(e^{i\theta})|^2 d\theta. \quad (1)$$

It is a Hilbert space when equipped with the inner product:

$$\langle f, g \rangle_{H(\mathbb{D})} = \frac{1}{2\pi} \int_0^{2\pi} f(e^{i\theta}) \overline{g(e^{i\theta})} d\theta. \quad (2)$$

Over the years, the applications of Hardy space $H(\mathbb{D})$ play an important role in various fields of mathematics [5, 10] and in certain parts of quantum mechanics

[11, 12]. And this space is the background of some applications. For example, in Section 3, we study on $H(\mathbb{D})$ the following two operators:

$$\nabla f(z) = f'(z), \quad Lf(z) = z^2 f'(z) + zf(z), \quad (3)$$

and we deduce uncertainty inequality for this space. Next, we establish the minimizer function associated with the difference operator:

$$T_1 f(z) = \frac{1}{z} (f(z) - f(0)). \quad (4)$$

In Section 4, we consider the Dirichlet space $\mathcal{D}(\mathbb{D})$, which is the set of all analytic functions f in the unit disk \mathbb{D} with the finite Dirichlet integral:

$$\int_{\mathbb{D}} |f'(z)|^2 \frac{dx dy}{\pi}, \quad z = x + iy. \quad (5)$$

It is also a Hilbert space when equipped with the inner product:

$$\langle f, g \rangle_{\mathcal{D}(\mathbb{D})} = f(0)\overline{g(0)} + \int_{\mathbb{D}} f'(z)\overline{g'(z)} \frac{dx dy}{\pi}, \quad z = x + iy. \quad (6)$$

This space is the objective of many applicable works [5, 13–17] and plays a background to our contribution. For example, we study on $\mathcal{D}(\mathbb{D})$ the following two operators:

$$\Lambda f(z) = f'(z) - f'(0), \quad Xf(z) = z^2 f'(z), \quad (7)$$

and we deduce the uncertainty inequality for this space $\mathcal{D}(\mathbb{D})$. And we establish the minimizer function associated with the difference operator:

$$T_2 f(z) = \frac{1}{z} (f(z) - zf'(0) - f(0)). \quad (8)$$

2. Generalized results

Let H be a Hilbert space equipped with the inner product $\langle \cdot, \cdot \rangle_H$. And let A and B be the two operators defined on H . We define the commutator $[A, B]$ by

$$[A, B] := AB - BA. \quad (9)$$

The adjoint of A denoted by A^* is defined by

$$\langle Af, g \rangle_H = \langle f, A^* g \rangle_H, \quad (10)$$

for $f \in \text{Dom}(A)$ and $g \in \text{Dom}(A^*)$.

Theorem 2.1. For $f \in \text{Dom}(AA^*) \cap \text{Dom}(A^*A)$, one has

$$\|A^* f\|_H^2 = \|Af\|_H^2 + \langle [A, A^*] f, f \rangle_H. \quad (11)$$

Proof. Let $f \in \text{Dom}(AA^*) \cap \text{Dom}(A^*A)$. Then AA^*f and A^*Af belong to H . Therefore $[A, A^*]f \in H$. Hence one has

$$\|A^*f\|_H^2 = \langle AA^*f, f \rangle_H = \langle A^*Af, f \rangle_H + \langle [A, A^*]f, f \rangle_H \quad (12)$$

$$= \|Af\|_H^2 + \langle [A, A^*]f, f \rangle_H. \quad \square \quad (13)$$

The following result is proved in [18, 19].

Theorem 2.2. Let A and B be the self-adjoint operators on a Hilbert space H . Then

$$\|(A - a)f\|_H \|(B - b)f\|_H \geq \frac{1}{2} |\langle [A, B]f, f \rangle_H|, \quad (14)$$

for all $f \in \text{Dom}(AB) \cap \text{Dom}(BA)$, and all $a, b \in \mathbb{R}$.

Theorem 2.3. Let $f \in \text{Dom}(AA^*) \cap \text{Dom}(A^*A)$. For all $a, b \in \mathbb{R}$, one has

$$\|(A + A^* - a)f\|_H \|(A - A^* + ib)f\|_H \geq |\|Af\|_H^2 - \|A^*f\|_H^2|, \quad (15)$$

where i is the imaginary unit.

Proof. Let us consider the following two operators on $\text{Dom}(AA^*) \cap \text{Dom}(A^*A)$ by

$$P = A + A^*, \quad Q = i(A - A^*). \quad (16)$$

It follows that, for $f \in \text{Dom}(AA^*) \cap \text{Dom}(A^*A)$, we have $Pf, Qf \in H$. The operators P and Q are self-adjoint and $[P, Q] = -2i[A, A^*]$. Thus the inequality (15) follows from Theorems 2.1 and 2.2. \square

Theorem 2.4. Let $f \in \text{Dom}(AA^*) \cap \text{Dom}(A^*A)$. Then

$$\Delta_H^+(f)\Delta_H^-(f) \geq \|f\|_H^4 (\|Af\|_H^2 - \|A^*f\|_H^2)^2, \quad (17)$$

where

$$\Delta_H^\pm(f) = \|f\|_H^2 \|(A \pm A^*)f\|_H^2 - |\langle (A \pm A^*)f, f \rangle_H|^2. \quad (18)$$

Proof. Let $f \in \text{Dom}(AA^*) \cap \text{Dom}(A^*A)$. The operator P given by (16) is self-adjoint; then for any real a , we have

$$\|(P - a)f\|_H^2 = \|Pf\|_H^2 + a^2\|f\|_H^2 - 2a\langle Pf, f \rangle_H. \quad (19)$$

This shows that

$$\min_{a \in \mathbb{R}} \|(P - a)f\|_H^2 = \|Pf\|_H^2 - \frac{|\langle Pf, f \rangle_H|^2}{\|f\|_H^2}, \quad (20)$$

and the minimum is attained when $a = \frac{\langle Pf, f \rangle_H}{\|f\|_H^2}$. In other words, we have

$$\min_{a \in \mathbb{R}} \|(A + A^* - a)f\|_H^2 = \|(A + A^*)f\|_H^2 - \frac{|\langle (A + A^*)f, f \rangle_H|^2}{\|f\|_H^2}. \quad (21)$$

Similarly

$$\min_{b \in \mathbb{R}} \|(A - A^* + ib)f\|_H^2 = \|(A - A^*)f\|_H^2 - \frac{|\langle (A - A^*)f, f \rangle_H|^2}{\|f\|_H^2}. \quad (22)$$

Then by (15), (21), and (22), we deduce the inequality (17). \square

Let $\lambda > 0$ and let $T : H \rightarrow K$ be a bounded linear operator from H into a Hilbert space K . Building on the ideas of Saitoh [2], we examine the minimizer function associated with the operator T .

Theorem 2.5. For any $k \in K$ and for any $\lambda > 0$, the problem

$$\inf_{f \in H} \{ \lambda \|f\|_H^2 + \|Tf - k\|_K^2 \} \tag{23}$$

has a unique minimizer given by

$$f_{\lambda,k}^* = (\lambda I + T^* T)^{-1} T^* k. \tag{24}$$

Proof. The problem (23) is solved elementarily by finding the roots of the first derivative $D\Phi$ of the quadratic and strictly convex function $\Phi(f) = \lambda \|f\|_H^2 + \|Tf - k\|_K^2$. Note that for convex functions, the equation $D\Phi(f) = 0$ is a necessary and sufficient condition for the minimum at f . The calculation provides

$$D\Phi(f) = 2\lambda f + 2T^*(Tf - k), \tag{25}$$

and the assertion of the theorem follows at once. \square

Theorem 2.6. If $T : H \rightarrow K$ is an isometric isomorphism; then for any $k \in K$ and for any $\lambda > 0$, the problem

$$\inf_{f \in H} \{ \lambda \|f\|_H^2 + \|Tf - k\|_K^2 \} \tag{26}$$

has a unique minimizer given by

$$f_{\lambda,h}^* = \frac{1}{\lambda + 1} T^{-1} k. \tag{27}$$

Proof. We have $T^* = T^{-1}$ and $T^* T = I$. Thus, by (24), we deduce the result. \square

3. The Hardy space $H(\mathbb{D})$

Let \mathbb{C} be the complex plane and $\mathbb{D} = \{z \in \mathbb{C} : |z| < 1\}$ the open unit disk. The Hardy space $H(\mathbb{D})$ is the set of all analytic functions f in the unit disk \mathbb{D} with the finite integral:

$$\int_0^{2\pi} |f(e^{i\theta})|^2 d\theta. \tag{28}$$

It is a Hilbert space when equipped with the inner product:

$$\langle f, g \rangle_{H(\mathbb{D})} = \frac{1}{2\pi} \int_0^{2\pi} f(e^{i\theta}) \overline{g(e^{i\theta})} d\theta. \tag{29}$$

If $f, g \in H(\mathbb{D})$ with $f(z) = \sum_{n=0}^{\infty} a_n z^n$ and $g(z) = \sum_{n=0}^{\infty} b_n z^n$, then

$$\langle f, g \rangle_{H(\mathbb{D})} = \sum_{n=0}^{\infty} a_n \overline{b_n}. \tag{30}$$

The set $\{z^n\}_{n=0}^{\infty}$ forms an Hilbert's basis for the space $H(\mathbb{D})$.
The Szegő kernel S_z given for $z \in \mathbb{D}$, by

$$S_z(w) = \sum_{n=0}^{\infty} \bar{z}^n w^n = \frac{1}{1 - \bar{z}w}, \quad w \in \mathbb{D}, \quad (31)$$

is a reproducing kernel for the Hardy space $H(\mathbb{D})$, meaning that $S_z \in H(\mathbb{D})$, and for all $f \in H(\mathbb{D})$, we have $\langle f, S_z \rangle_{H(\mathbb{D})} = f(z)$.

For $z \in \mathbb{D}$, the function $u(z) = S_{\bar{z}}(w)$ is the unique analytic solution on \mathbb{D} of the initial problem:

$$u'(z) = w(zu'(z) + u(z)), \quad w \in \mathbb{D}, \quad u(0) = 1. \quad (32)$$

In the next of this section, we define the operators ∇ , \mathfrak{R} , and L on $H(\mathbb{D})$ by

$$\nabla f(z) = f'(z), \quad \mathfrak{R}f(z) = zf'(z), \quad Lf(z) = z^2 f'(z) + zf(z). \quad (33)$$

These operators satisfy the commutation rule:

$$[\nabla, L] = \nabla L - L\nabla = 2\mathfrak{R} + I, \quad (34)$$

where I is the identity operator.

We define the Hilbert space $U(\mathbb{D})$ as the space of all analytic functions f in the unit disk \mathbb{D} such that

$$\|f\|_{U(\mathbb{D})}^2 = \frac{1}{2\pi} \int_0^{2\pi} |f'(e^{i\theta})|^2 d\theta < \infty. \quad (35)$$

If $f \in U(\mathbb{D})$ with $f(z) = \sum_{n=0}^{\infty} a_n z^n$, then

$$\|f\|_{U(\mathbb{D})}^2 = \sum_{n=1}^{\infty} n^2 |a_n|^2. \quad (36)$$

Thus, the space $U(\mathbb{D})$ is a subspace of the Hardy space $H(\mathbb{D})$.

Theorem 3.1.

- i. For $f \in U(\mathbb{D})$, then ∇f , $\mathfrak{R}f$ and Lf belong to $H(\mathbb{D})$.
- ii. $\nabla^* = L$.
- iii. For $f \in U(\mathbb{D})$, one has

$$\|Lf\|_{H(\mathbb{D})}^2 = \|\nabla f\|_{H(\mathbb{D})}^2 + \|f\|_{H(\mathbb{D})}^2 + 2\langle \mathfrak{R}f, f \rangle_{H(\mathbb{D})}. \quad (37)$$

Proof.

- i. Let $f \in U(\mathbb{D})$ with $f(z) = \sum_{n=0}^{\infty} a_n z^n$. Then

$$\nabla f(z) = \sum_{n=0}^{\infty} (n+1)a_{n+1}z^n, \quad \mathfrak{R}f(z) = \sum_{n=1}^{\infty} na_n z^n, \quad (38)$$

and

$$Lf(z) = \sum_{n=1}^{\infty} na_{n-1}z^n. \quad (39)$$

Therefore

$$\|\nabla f\|_{H(\mathbb{D})}^2 = \sum_{n=0}^{\infty} (n+1)^2 |a_{n+1}|^2 = \|f\|_{U(\mathbb{D})}^2, \quad (40)$$

$$\|\Re f\|_{H(\mathbb{D})}^2 = \sum_{n=1}^{\infty} n^2 |a_n|^2 = \|f\|_{U(\mathbb{D})}^2, \quad (41)$$

and

$$\|Lf\|_{H(\mathbb{D})}^2 = \sum_{n=0}^{\infty} (n+1)^2 |a_n|^2 \leq |f(0)|^2 + 4\|f\|_{U(\mathbb{D})}^2. \quad (42)$$

Consequently ∇f , $\Re f$, and Lf belong to $H(\mathbb{D})$.

ii. For $f, g \in U(\mathbb{D})$ with $f(z) = \sum_{n=0}^{\infty} a_n z^n$ and $g(z) = \sum_{n=0}^{\infty} b_n z^n$, one has

$$\langle \nabla f, g \rangle_{H(\mathbb{D})} = \sum_{n=0}^{\infty} (n+1) a_{n+1} \overline{b_n} = \sum_{n=1}^{\infty} n a_n \overline{b_{n-1}} = \langle f, Lg \rangle_{H(\mathbb{D})}. \quad (43)$$

Thus $\nabla^* = L$.

iii. Let $f \in U(\mathbb{D})$. By (ii) and (34), we deduce that

$$\|Lf\|_{H(\mathbb{D})}^2 = \langle \nabla Lf, f \rangle_{H(\mathbb{D})} \quad (44)$$

$$= \langle L\nabla f, f \rangle_{H(\mathbb{D})} + \langle [\nabla, L]f, f \rangle_{H(\mathbb{D})} \quad (45)$$

$$= \|\nabla f\|_{H(\mathbb{D})}^2 + \|f\|_{H(\mathbb{D})}^2 + 2\langle \Re f, f \rangle_{H(\mathbb{D})}. \quad \square \quad (46)$$

Theorem 3.2. Let $f \in U(\mathbb{D})$. For all $a, b \in \mathbb{R}$, one has

$$\|(\nabla + L - a)f\|_{H(\mathbb{D})} \|(\nabla - L + ib)f\|_{H(\mathbb{D})} \geq \|f\|_{H(\mathbb{D})}^2 + 2\langle \Re f, f \rangle_{H(\mathbb{D})}. \quad (47)$$

Theorem 3.3. Let T_1 be the difference operator defined on $H(\mathbb{D})$ by

$$T_1 f(z) = \frac{1}{z} (f(z) - f(0)). \quad (48)$$

i. The operator T_1 maps continuously from $H(\mathbb{D})$ to $H(\mathbb{D})$, and

$$\|T_1 f\|_{H(\mathbb{D})} \leq \|f\|_{H(\mathbb{D})}. \quad (49)$$

ii. For $f \in H(\mathbb{D})$ and $z \in \mathbb{D}$, we have

$$T_1^* f(z) = zf(z), \quad T_1^* T_1 f(z) = f(z) - f(0). \quad (50)$$

iii. For any $h \in H(\mathbb{D})$ and for any $\lambda > 0$, the problem

$$\inf_{f \in H(\mathbb{D})} \left\{ \lambda \|f\|_{H(\mathbb{D})}^2 + \|T_1 f - h\|_{H(\mathbb{D})}^2 \right\} \quad (51)$$

has a unique minimizer given by

$$f_{\lambda, h}^*(z) = \frac{1}{\lambda + 1} zh(z), \quad z \in \mathbb{D}. \quad (52)$$

Proof.

i. If $f \in H(\mathbb{D})$ with $f(z) = \sum_{n=0}^{\infty} a_n z^n$, then $T_1 f(z) = \sum_{n=0}^{\infty} a_{n+1} z^n$ and

$$\|T_1 f\|_{H(\mathbb{D})}^2 = \sum_{n=1}^{\infty} |a_n|^2 \leq \|f\|_{H(\mathbb{D})}^2. \quad (53)$$

ii. If $f, g \in H(\mathbb{D})$ with $f(z) = \sum_{n=0}^{\infty} a_n z^n$ and $g(z) = \sum_{n=0}^{\infty} b_n z^n$, then

$$\langle T_1 f, g \rangle_{H(\mathbb{D})} = \sum_{n=0}^{\infty} a_{n+1} \bar{b}_n = \sum_{n=1}^{\infty} a_n \bar{b}_{n-1} = \langle f, T_1^* g \rangle_{H(\mathbb{D})}, \quad (54)$$

where $T_1^* g(z) = zg(z)$, for $z \in \mathbb{D}$. And therefore

$$T_1^* T_1 f(z) = zT_1 f(z) = f(z) - f(0). \quad (55)$$

iii. From Theorem 2.5 we have

$$(\lambda I + T_1^* T_1) f_{\lambda, h}^*(z) = T_1^* h(z). \quad (56)$$

By (ii) we deduce that

$$(\lambda + 1) f_{\lambda, h}^*(z) - f_{\lambda, h}^*(0) = zh(z). \quad (57)$$

And from this equation, $f_{\lambda, h}^*(0) = 0$. Hence

$$f_{\lambda, h}^*(z) = \frac{1}{\lambda + 1} zh(z). \quad \square \quad (58)$$

4. The Dirichlet space $\mathcal{D}(\mathbb{D})$

The Dirichlet space $\mathcal{D}(\mathbb{D})$ is the set of all analytic functions f in the unit disk \mathbb{D} with the finite Dirichlet integral:

$$\int_{\mathbb{D}} |f'(z)|^2 \frac{dx dy}{\pi}, \quad z = x + iy. \quad (59)$$

It is a Hilbert space when equipped with the inner product:

$$\langle f, g \rangle_{\mathcal{D}(\mathbb{D})} = f(0)\bar{g}(0) + \int_{\mathbb{D}} f'(z)\bar{g}'(z) \frac{dx dy}{\pi}, \quad z = x + iy. \quad (60)$$

If $f, g \in \mathcal{D}(\mathbb{D})$ with $f(z) = \sum_{n=0}^{\infty} a_n z^n$ and $g(z) = \sum_{n=0}^{\infty} b_n z^n$, then

$$\langle f, g \rangle_{\mathcal{D}(\mathbb{D})} = a_0 \bar{b}_0 + \sum_{n=1}^{\infty} n a_n \bar{b}_n. \quad (61)$$

The set $\left\{1, \frac{z^n}{\sqrt{n}}\right\}_{n=1}^{\infty}$ forms an Hilbert's basis for the space $\mathcal{D}(\mathbb{D})$.

The function K_z given for $z \in \mathbb{D}$, by

$$K_z(w) = 1 + \log \left(\frac{1}{1 - \bar{z}w} \right), \quad w \in \mathbb{D}, \quad (62)$$

is a reproducing kernel for the Dirichlet space $\mathcal{D}(\mathbb{D})$, meaning that $K_z \in \mathcal{D}(\mathbb{D})$, and for all $f \in \mathcal{D}(\mathbb{D})$, we have $\langle f, K_z \rangle_{\mathcal{D}(\mathbb{D})} = f(z)$.

For $z \in \mathbb{D}$, the function $u(z) = K_{\bar{z}}(w)$ is the unique analytic solution on \mathbb{D} of the initial problem:

$$\frac{u'(z) - u'(0)}{z} = wu'(z), \quad w \in \mathbb{D}, \quad u(0) = 1. \quad (63)$$

In the next of this section, we define the operators Λ , \mathfrak{R} , and X on $\mathcal{D}(\mathbb{D})$ by

$$\Lambda f(z) = f'(z) - f'(0), \quad \mathfrak{R}f(z) = zf'(z), \quad Xf(z) = z^2f'(z). \quad (64)$$

These operators satisfy the following commutation relation:

$$[\Lambda, X] = \Lambda X - X\Lambda = 2\mathfrak{R}. \quad (65)$$

We define the Hilbert space $V(\mathbb{D})$ as the space of all analytic functions f in the unit disk \mathbb{D} such that

$$\|f\|_{V(\mathbb{D})}^2 = \int_{\mathbb{D}} |f'(z)|^2 |z|^2 \frac{dx dy}{\pi} < \infty, \quad z = x + iy. \quad (66)$$

If $f \in V(\mathbb{D})$ with $f(z) = \sum_{n=0}^{\infty} a_n z^n$, then

$$\|f\|_{V(\mathbb{D})}^2 = \sum_{n=1}^{\infty} n^3 |a_n|^2. \quad (67)$$

Thus, the space $V(\mathbb{D})$ is a subspace of the Dirichlet space $\mathcal{D}(\mathbb{D})$.

Theorem 4.1.

i. For $f \in V(\mathbb{D})$, then Λf , $\mathfrak{R}f$, and Xf belong to $\mathcal{D}(\mathbb{D})$.

ii. $\Lambda^* = X$.

iii. For $f \in V(\mathbb{D})$, one has

$$\|Xf\|_{\mathcal{D}(\mathbb{D})}^2 = \|\Lambda f\|_{\mathcal{D}(\mathbb{D})}^2 + 2\langle \mathfrak{R}f, f \rangle_{\mathcal{D}(\mathbb{D})}. \quad (68)$$

Proof.

i. Let $f \in V(\mathbb{D})$ with $f(z) = \sum_{n=0}^{\infty} a_n z^n$. Then

$$\Lambda f(z) = \sum_{n=1}^{\infty} (n+1)a_{n+1}z^n, \quad \mathfrak{R}f(z) = \sum_{n=1}^{\infty} na_n z^n, \quad (69)$$

and

$$Xf(z) = \sum_{n=2}^{\infty} (n-1)a_{n-1}z^n. \quad (70)$$

Therefore

$$\|\Lambda f\|_{\mathcal{D}(\mathbb{D})}^2 = \sum_{n=1}^{\infty} n(n+1)^2 |a_{n+1}|^2 \leq \sum_{n=2}^{\infty} n^3 |a_n|^2 \leq \|f\|_{V(\mathbb{D})}^2, \quad (71)$$

$$\|\Re f\|_{\mathcal{D}(\mathbb{D})}^2 = \sum_{n=1}^{\infty} n^3 |a_n|^2 = \|f\|_{V(\mathbb{D})}^2, \quad (72)$$

and

$$\|Xf\|_{\mathcal{D}(\mathbb{D})}^2 = \sum_{n=1}^{\infty} (n+1)n^2 |a_n|^2 \leq 2\|f\|_{V(\mathbb{D})}^2. \quad (73)$$

Consequently Λf , $\Re f$, and Xf belong to $\mathcal{D}(\mathbb{D})$.

ii. For $f, g \in V(\mathbb{D})$ with $f(z) = \sum_{n=0}^{\infty} a_n z^n$ and $g(z) = \sum_{n=0}^{\infty} b_n z^n$, one has

$$\langle \Lambda f, g \rangle_{\mathcal{D}(\mathbb{D})} = \sum_{n=1}^{\infty} n(n+1) a_{n+1} \bar{b}_n = \sum_{n=2}^{\infty} n(n-1) a_n \bar{b}_{n-1} = \langle f, Xg \rangle_{\mathcal{D}(\mathbb{D})}. \quad (74)$$

iii. Let $f \in V(\mathbb{D})$. By (ii) and (65), we deduce that

$$\|Xf\|_{\mathcal{D}(\mathbb{D})}^2 = \langle \Lambda Xf, f \rangle_{\mathcal{D}(\mathbb{D})} \quad (75)$$

$$= \langle X\Lambda f, f \rangle_{\mathcal{D}(\mathbb{D})} + \langle [\Lambda, X]f, f \rangle_{\mathcal{D}(\mathbb{D})} \quad (76)$$

$$= \|\Lambda f\|_{\mathcal{D}(\mathbb{D})}^2 + 2\langle \Re f, f \rangle_{\mathcal{D}(\mathbb{D})}. \quad \square \quad (77)$$

Theorem 4.2. Let $f \in V(\mathbb{D})$. For all $a, b \in \mathbb{R}$, one has

$$\|(\Lambda + X - a)f\|_{\mathcal{D}(\mathbb{D})} \|(\Lambda - X + ib)f\|_{\mathcal{D}(\mathbb{D})} \geq 2\langle \Re f, f \rangle_{\mathcal{D}(\mathbb{D})}. \quad (78)$$

Theorem 4.3. Let T_2 be the difference operator defined on $\mathcal{D}(\mathbb{D})$ by

$$T_2 f(z) = \frac{1}{z} (f(z) - zf'(0) - f(0)). \quad (79)$$

i. The operator T_2 maps continuously from $\mathcal{D}(\mathbb{D})$ to $\mathcal{D}(\mathbb{D})$, and

$$\|T_2 f\|_{\mathcal{D}(\mathbb{D})} \leq \|f\|_{\mathcal{D}(\mathbb{D})}. \quad (80)$$

ii. For $f \in \mathcal{D}(\mathbb{D})$ with $f(z) = \sum_{n=0}^{\infty} a_n z^n$, we have

$$T_2^* f(z) = \sum_{n=2}^{\infty} \frac{n-1}{n} a_{n-1} z^n, \quad T_2^* T_2 f(z) = \sum_{n=2}^{\infty} \frac{n-1}{n} a_n z^n. \quad (81)$$

iii. For any $d \in \mathcal{D}(\mathbb{D})$ and for any $\lambda > 0$, the problem

$$\inf_{f \in \mathcal{D}(\mathbb{D})} \left\{ \lambda \|f\|_{\mathcal{D}(\mathbb{D})}^2 + \|T_2 f - d\|_{\mathcal{D}(\mathbb{D})}^2 \right\} \quad (82)$$

has a unique minimizer given by

$$f_{\lambda, d}^*(z) = \langle d, \Psi_z \rangle_{\mathcal{D}(\mathbb{D})}, \quad z \in \mathbb{D}, \quad (83)$$

$$\Psi_z(w) = \sum_{n=1}^{\infty} \frac{\bar{z}^{n+1}}{\lambda(n+1) + n} w^n, \quad w \in \mathbb{D}. \quad (84)$$

Proof.

i. If $f \in \mathcal{D}(\mathbb{D})$ with $f(z) = \sum_{n=0}^{\infty} a_n z^n$, then $T_2 f(z) = \sum_{n=1}^{\infty} a_{n+1} z^n$ and

$$\|T_2 f\|_{\mathcal{D}(\mathbb{D})}^2 = \sum_{n=2}^{\infty} (n-1) |a_n|^2 \leq \sum_{n=2}^{\infty} n |a_n|^2 \leq \|f\|_{\mathcal{D}(\mathbb{D})}^2. \quad (85)$$

ii. If $f, g \in \mathcal{D}(\mathbb{D})$ with $f(z) = \sum_{n=0}^{\infty} a_n z^n$ and $g(z) = \sum_{n=0}^{\infty} b_n z^n$, then

$$\langle T_2 f, g \rangle_{\mathcal{D}(\mathbb{D})} = \sum_{n=1}^{\infty} n a_{n+1} \bar{b}_n = \sum_{n=2}^{\infty} (n-1) a_n \bar{b}_{n-1} = \langle f, T_2^* g \rangle_{\mathcal{D}(\mathbb{D})}, \quad (86)$$

where

$$T_2^* g(z) = \sum_{n=2}^{\infty} \frac{n-1}{n} b_{n-1} z^n, \quad z \in \mathbb{D}. \quad (87)$$

And therefore

$$T_2^* T_2 f(z) = \sum_{n=2}^{\infty} \frac{n-1}{n} a_n z^n. \quad (88)$$

iii. We put $d(z) = \sum_{n=0}^{\infty} d_n z^n$ and

$$f_{\lambda, d}^*(z) = \sum_{n=0}^{\infty} c_n z^n. \quad (89)$$

From (ii) and the equation

$$(\lambda I + T_2^* T_2) f_{\lambda, d}^*(z) = T_2^* d(z), \quad (90)$$

we deduce that

$$c_1 = c_0 = 0, \quad c_n = \frac{n-1}{\lambda n + n - 1} d_{n-1}, \quad n \geq 2. \quad (91)$$

Thus

$$f_{\lambda, d}^*(z) = \sum_{n=1}^{\infty} \frac{n d_n}{\lambda(n+1) + n} z^{n+1} = \langle d, \Psi_z \rangle_{\mathcal{D}(\mathbb{D})}, \quad z \in \mathbb{D}. \quad \square \quad (92)$$



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Fixed Point Theorems of a New Generalized Nonexpansive Mapping

Shi Jie

Abstract

This paper introduces a $T - (D_a)$ mapping that is weaker than the nonexpansive mapping. It introduces several iterations for the fixed point of the $T - (D_a)$ mapping. It gives fixed point theorems and convergence theorems for the $T - (D_a)$ mapping in Banach space, instead of uniformly convex Banach space. This paper gives some basic properties on the $T - (D_a)$ mapping and gives the example to show the existence of $T - (D_a)$ mapping. The results of this paper are obtained in general Banach spaces. It considers some sufficient conditions for convergence of fixed points of mappings in general Banach spaces under higher iterations.

Keywords: iteration, convergence theorems, nonexpansive mapping, fixed point
2010 MSC: 47H09, 47H10

1. Introduction

In this paper, E is a Banach space, C is a nonempty closed convex subset of E , and $Fix(T) = \{x \in C : Tx = x\}$.

Definition 1. T is contraction mapping if there is $r \in [0, 1)$

$$\|Tx - Ty\| \leq r\|x - y\| \text{ for all } x, y \in C.$$

Definition 2. T is nonexpansive mapping if

$$\|Tx - Ty\| \leq \|x - y\| \text{ for all } x, y \in C.$$

Definition 3. T is quasinonexpansive mapping if

$$\|Tx - Ty\| \leq \|x - y\| \text{ for all } x \in C, y \in F(T).$$

Definition 4. $T : C \rightarrow C$ is a $T - (D_a)$ mapping on a subset C , if there is $a \in (\frac{1}{2}, 1)$, $\|Tx - Ty\| \leq \|x - y\|$ for all $\alpha \in [a, 1]$, $x \in C, y \in C(T, x, \alpha)$, where $C(T, x, \alpha) = \{y \in C | y = (1 - \alpha)p + \alpha Tp, p \in C, \|Tp - p\| \leq \|Tx - x\|\}$.

In 2008 Suzuki [1] defined a mapping T in Banach space: $\frac{1}{2}\|Tx - Ty\| \leq \|x - y\|$ implies $\|Tx - Ty\| \leq \|x - y\|$. And T is said to satisfy condition (C). Suzuki [1] showed that the mapping satisfying condition (C) is weaker than nonexpansive mapping and stronger than quasinonexpansive mapping.

Suzuki [1] proved the theorem T is a mapping in Banach space, T satisfies condition (C), and $\{x_n\}$ is the sequence defined by the iteration process:

$$\begin{cases} x_1 = x \in C, \\ x_{n+1} = (1 - \alpha_n)x_n + \alpha_n Tx_n, \end{cases} \quad (1)$$

then $\{x_n\}$ converges to a fixed point of T .

Suzuki [1] gave this convergence theorem in an ordinary Banach space, and the mapping satisfying condition (C) is weaker than nonexpansive mapping.

In 2016, Thakur [2] proved the theorem T is a mapping in uniformly convex Banach space, T satisfies condition (C), and $\{x_n\}$ is the sequence defined by iteration process:

$$\begin{cases} x_1 = x \in C, \\ x_{n+1} = Ty_n, \\ y_n = Tz_n, \\ z_n = (1 - \alpha_n)x_n + \alpha_n Tx_n, \end{cases} \quad (2)$$

then $\{x_n\}$ converges to a fixed point of T .

Thakur [2] claimed that the rate of iteration is fastest of known iterations. However, the disadvantage is that their results must be in uniformly convex Banach space, instead of the ordinary Banach space.

The aim of this article is there exists a generalized nonexpansive mapping, which makes the sequence generated by Thakur's iteration converge to a fixed point in a general Banach space.

The following propositions are obvious:

Proposition 1. If T is nonexpansive, then T satisfies condition (D_a) .

Proposition 2. If T is $T - (D_a)$ mapping, then T is quasinonexpansive.

Proposition 3. Suppose $T : C \rightarrow C$ is a $T - (D_a)$ mapping. Then, for $x, y \in C$:

- (1) $\|T^2x - Tx\| \leq \|Tx - x\|$ for all $x \in C$.
- (2) $\|T^2x - Ty\| \leq \|Tx - y\|$ or $\|T^2y - Tx\| \leq \|Ty - x\|$ for all $x, y \in C$.

Proof:

- (1) Since $\|Tx - x\| \leq \|Tx - x\|$, $Tx \in C(T, x, 1)$, we have $\|T^2x - Tx\| \leq \|Tx - x\|$.
- (2) For all $x, y \in C$, $\|Tx - x\| \leq \|Ty - y\|$ or $\|Ty - y\| \leq \|Tx - x\|$.

Then $Tx \in C(T, y, \alpha)$ or $Ty \in C(T, x, \alpha)$.

It follows that $\|T^2x - Ty\| \leq \|Tx - y\|$ or $\|T^2y - Tx\| \leq \|Ty - x\|$.

Example 1

$$Tx = \begin{cases} \begin{pmatrix} 1.1 & x_2 \\ x_3 & x_4 \end{pmatrix}, & x_1 = 3, \\ \begin{pmatrix} 0 & x_2 \\ x_3 & x_4 \end{pmatrix}, & x_1 \neq 3, \end{cases}$$

where

$$x = \begin{pmatrix} x_1 & x_2 \\ x_3 & x_4 \end{pmatrix}, x_1 \in [0, 3], x_2 \in [0, 0.01], x_3 \in [0, 0.01], x_4 \in [0, 0.01].$$

$$\|x\|_1 = \max\{|x_1| + |x_3|, |x_2| + |x_4|\}$$

Set

$$x = \begin{pmatrix} 3 & 0 \\ 0 & 0 \end{pmatrix}$$

and

$$y = \begin{pmatrix} 2.5 & 0 \\ 0 & 0 \end{pmatrix}$$

We see that

$$\|Tx - Ty\|_1 = 1.1 > \|x - y\|_1.$$

Hence, T is not a nonexpansive mapping.

To verify that T is a $T - (D_a)$ mapping, consider the following cases:

Case 1:

$$\alpha \in \left[\frac{11}{19}, 1\right], x = \begin{pmatrix} x_1 & x_2 \\ x_3 & x_4 \end{pmatrix}, x_1 \neq 3.$$

$$y = \begin{pmatrix} y_1 & y_2 \\ y_3 & y_4 \end{pmatrix} \in C(T, x, \alpha),$$

then $y_1 \neq 3$. We have

$$\|Ty - Tx\| = \left\| \begin{pmatrix} 0 & y_2 - x_2 \\ y_3 - x_3 & y_4 - x_4 \end{pmatrix} \right\| \leq \left\| \begin{pmatrix} y_1 - x_1 & y_2 - x_2 \\ y_3 - x_3 & y_4 - x_4 \end{pmatrix} \right\| = \|y - x\|$$

Case 2:

$$\alpha \in \left[\frac{11}{19}, 1\right], x = \begin{pmatrix} x_1 & x_2 \\ x_3 & x_4 \end{pmatrix}, x_1 = 3.$$

$$y = \begin{pmatrix} y_1 & y_2 \\ y_3 & y_4 \end{pmatrix} \in C(T, x, \alpha),$$

then $y_1 \in [0, 1.9]$. We have

$$\|Ty - Tx\| = \left\| \begin{pmatrix} 1.1 & y_2 - x_2 \\ y_3 - x_3 & y_4 - x_4 \end{pmatrix} \right\| \leq 1.11 \leq \|y - x\|$$

Hence, T is a $T - (D_a)$ mapping, and T is not nonexpansive.

2. Fixed point

In this section, we prove convergence theorems for fixed point of the $T - (D_a)$ mapping in Banach space.

Lemma 1. Let C be bounded convex subset of a Banach space B . Assume that $T : C \rightarrow C$ is $T - (D_a)$ mapping and $\{x_n\}, \{y_n\}, \{z_n\}$ are sequences generated by iteration:

$$\begin{cases} x_1 = x \in C, \\ x_{n+1} = Ty_n, \\ y_n = Tz_n, \\ z_n = (1 - \alpha_n)x_n + \alpha_n Tx_n, \end{cases} \quad (3)$$

where $\frac{1}{2} < a \leq \alpha_n \leq b < 1$. Then

- (1) $\|Tx_{n+1} - x_{n+1}\| \leq \|Ty_n - y_n\| \leq \|Tz_n - z_n\| \leq \|Tx_n - x_n\|$.
- (2) $\lim_{n \rightarrow \infty} \|Tx_n - x_n\| = \lim_{n \rightarrow \infty} \|Ty_n - y_n\| = \lim_{n \rightarrow \infty} \|Tz_n - z_n\| = r \geq 0$.

Proof: (1) From Proposition 3 and $z_n = (1 - \alpha_n)x_n + \alpha_n Tx_n$, we have

$$\begin{aligned} \|Tx_{n+1} - x_{n+1}\| &\leq \|T^2y_n - Ty_n\| \\ &\leq \|Ty_n - y_n\| = \|T^2z_n - Tz_n\| \\ &\leq \|Tz_n - z_n\| \\ &= \|Tz_n - Tx_n + (1 - \alpha_n)(Tx_n - x_n)\| \\ &\leq \|z_n - x_n\| + (1 - \alpha_n)\|Tx_n - x_n\| \\ &= \|Tx_n - x_n\|. \end{aligned}$$

(2) From (1), we have $0 \leq \|Tx_{n+1} - x_{n+1}\| \leq \|Tx_n - x_n\|$. So $\lim_{n \rightarrow \infty} \|Tx_n - x_n\| = r \geq 0$. Now, we have $\lim_{n \rightarrow \infty} \|Tx_n - x_n\| = \lim_{n \rightarrow \infty} \|Ty_n - y_n\| = \lim_{n \rightarrow \infty} \|Tz_n - z_n\| = r \geq 0$.

Lemma 2. Assume that $T : C \rightarrow C$ is a $T - (D_a)$ mapping and $\{x_n\}, \{y_n\}, \{z_n\}$ are sequences generated by iteration (3). $\frac{1}{2} < a \leq \alpha_n \leq b < 1$. Let $\{u_m\}$ satisfy $u_{3n-2} = x_n, u_{3n-1} = z_n, u_{3n} = y_n$. Then, for all $n \geq 1, p \geq 1$

$$\begin{aligned} &\left(1 + \sum_{k=3n-2}^{3n+p-3} \beta_k\right) \|Tu_{3n-2} - u_{3n-2}\| \leq \|Tu_{3n-2+p} - u_{3n-2}\| \\ &+ \left(\prod_{k=n}^{n+p-1} \frac{2}{1 - \alpha_k}\right) (\|Tu_{3n-2} - u_{3n-2}\| - \|Tu_{3n-2+3p} - u_{3n-2+3p}\|), \end{aligned} \quad (4)$$

where

$$\beta_k = \begin{cases} \alpha_n & k = 3n - 2 \\ 1 & k \neq 3n - 2 \end{cases}$$

Proof: From Lemma 1, we have

$$\begin{aligned} &\|Tx_{n+1} - x_{n+1}\| \\ &\leq \|Tz_n - z_n\| \\ &= \|Tz_n - (1 - \alpha_n)x_n - \alpha_n Tx_n\| \\ &\leq (1 - \alpha_n)\|Tz_n - x_n\| + \alpha_n\|Tz_n - Tx_n\| \\ &\leq (1 - \alpha_n)\|Tz_n - x_n\| + \alpha_n\|z_n - x_n\| \\ &= (1 - \alpha_n)\|Tz_n - x_n\| + \alpha_n^2\|Tx_n - x_n\|. \end{aligned}$$

So, for $p = 1$ and all $n \geq 1$

$$\begin{aligned} & (1 + \beta_{3n-2}) \|Tu_{3n-2} - u_{3n-2}\| \\ &= (1 + \alpha_n) \|Tx_n - x_n\| \\ &\leq \|Tx_n - x_n\| + \left(\frac{1}{1 - \alpha_n}\right) (\|Tx_n - x_n\| - \|Tx_{n+1} - x_{n+1}\|) \\ &= \|Tu_{3n-1} - u_{3n-2}\| + \left(\frac{1}{1 - \alpha_n}\right) (\|Tu_{3n-2} - u_{3n-2}\| - \|Tu_{3n+1} - u_{3n+1}\|) \\ &\leq \|Tu_{3n-1} - u_{3n-2}\| + \left(\frac{2}{1 - \alpha_n}\right) (\|Tu_{3n-2} - u_{3n-2}\| - \|Tu_{3n+1} - u_{3n+1}\|). \end{aligned}$$

(4) holds.

We make the inductive assumption that (4) holds for a given $p > 1$ and all $n > 0$ and obtain, upon replacing n with $n + 1$

$$\begin{aligned} & \left(1 + \sum_{k=3n+1}^{3n+p} \beta_k\right) \|Tu_{3n+1} - u_{3n+1}\| \leq \|Tu_{3n+1+p} - u_{3n+1}\| \\ & + \left(\prod_{k=n+1}^{n+p} \frac{2}{1 - \alpha_k}\right) (\|Tu_{3n+1} - u_{3n+1}\| - \|Tu_{3n+1+3p} - u_{3n+1+3p}\|). \end{aligned} \tag{5}$$

And obviously

$$k \geq 3n - 2, \|Tu_{k+1} - Tu_k\| \leq \beta_k \|Tu_{3n-2} - u_{3n-2}\|, \tag{6}$$

$$k > t, \|Tu_k - Tu_t\| \leq \|u_k - u_t\|. \tag{7}$$

Case 1: $p = 3m, m \geq 1$. From (6) and (7)

$$\begin{aligned} & \|Tu_{3n+1+p} - u_{3n+1}\| \\ &= \|Tx_{n+m+1} - x_{n+1}\| \\ &= \|Tx_{n+m+1} - Ty_n\| \\ &\leq \|x_{n+m+1} - y_n\| \\ &= \|Ty_{n+m} - Tz_n\| \\ &\leq \|y_{n+m} - z_n\| \\ &= \|Tz_{n+m} - (1 - \alpha_n)x_n - \alpha_n Tx_n\| \\ &\leq (1 - \alpha_n) \|Tz_{n+m} - x_n\| + \alpha_n \|Tz_{n+m} - Tx_n\| \\ &\leq (1 - \alpha_n) \|Tz_{n+m} - x_n\| \\ &+ \alpha_n (\|Tz_{n+m} - Tx_{n+m}\| + \|Tx_{n+m} - Ty_{n+m-1}\| + \dots + \|Ty_n - Tz_n\| + \|Tz_n - Tx_n\|) \\ &= (1 - \alpha_n) \|Tu_{3n-1+p} - u_{3n-2}\| + \alpha_n \sum_{k=3n-2}^{3n-2+p} \|Tu_{k+1} - Tu_k\| \\ &\leq (1 - \alpha_n) \|Tu_{3n-1+p} - u_{3n-2}\| + \alpha_n \sum_{k=3n-2}^{3n-2+p} \beta_k \|Tu_{3n-2} - u_{3n-2}\|. \end{aligned}$$

It follows that

$$\|Tu_{3n+1+p} - u_{3n+1}\| \leq (1 - \alpha_n) \|Tu_{3n-1+p} - u_{3n-2}\| + \alpha_n \sum_{k=3n-2}^{3n-2+p} \beta_k \|Tu_{3n-2} - u_{3n-2}\| \tag{8}$$

Using (5) and (8), we have

$$\begin{aligned} & \left(1 + \sum_{k=3n+1}^{3n+p} \beta_k\right) \|Tu_{3n+1} - u_{3n+1}\| \\ & \leq (1 - \alpha_n) \|Tu_{3n-1+p} - u_{3n-2}\| + \alpha_n \sum_{k=3n-2}^{3n-2+p} \beta_k \|Tu_{3n-2} - u_{3n-2}\| \\ & + \left(\prod_{k=n+1}^{n+p} \frac{2}{1 - \alpha_k}\right) (\|Tu_{3n+1} - u_{3n+1}\| - \|Tu_{3n+1+3p} - u_{3n+1+3p}\|). \end{aligned}$$

From $\left(1 + \sum_{k=3n+1}^{3n+p} \beta_k\right) \leq \left(\prod_{k=n+1}^{n+p} \frac{1}{1 - \alpha_k}\right)$ and $\|Tu_{3n+1} - u_{3n+1}\| \leq \|Tu_{3n-2} - u_{3n-2}\|$, we have

$$\begin{aligned} & \left(1 + \sum_{k=3n+1}^{3n+p} \beta_k\right) \|Tu_{3n-2} - u_{3n-2}\| \\ & \leq (1 - \alpha_n) \|Tu_{3n-1+p} - u_{3n-2}\| + \alpha_n \sum_{k=3n-2}^{3n-2+p} \beta_k \|Tu_{3n-2} - u_{3n-2}\| \\ & + \left(\prod_{k=n+1}^{n+p} \frac{2}{1 - \alpha_k}\right) (\|Tu_{3n-2} - u_{3n-2}\| - \|Tu_{3n+1+3p} - u_{3n+1+3p}\|). \end{aligned}$$

Then

$$\begin{aligned} & \frac{\left(1 + \sum_{k=3n+1}^{3n+p} \beta_k - \alpha_n \sum_{k=3n-2}^{3n-2+p} \beta_k\right)}{1 - \alpha_n} \|Tu_{3n-2} - u_{3n-2}\| \\ & \leq \|Tu_{3n-1+p} - u_{3n-2}\| \\ & + \left(\prod_{k=n}^{n+p} \frac{2}{1 - \alpha_k}\right) (\|Tu_{3n-2} - u_{3n-2}\| - \|Tu_{3n+1+3p} - u_{3n+1+3p}\|). \end{aligned}$$

It follows that

$$\begin{aligned} & \left(1 + \sum_{k=3n-2}^{3n-2+p} \beta_k\right) \|Tu_{3n-2} - u_{3n-2}\| \\ & \leq \|Tu_{3n-1+p} - u_{3n-2}\| \\ & + \left(\prod_{k=n}^{n+p} \frac{2}{1 - \alpha_k}\right) (\|Tu_{3n-2} - u_{3n-2}\| - \|Tu_{3n+1+3p} - u_{3n+1+3p}\|). \end{aligned}$$

Thus, for $n, p + 1$, (4) holds.

Case 2: $p = 3m + 1, m \geq 0$. From (6) and (7), we have

$$\begin{aligned} & \|Tu_{3n+1+p} - u_{3n+1}\| \\ & = \|Tz_{n+m+1} - x_{n+1}\| \\ & = \|Tz_{n+m+1} - Ty_n\| \\ & \leq \|z_{n+m+1} - y_n\| \\ & = \|(1 - \alpha_{m+n+1})x_{m+n+1} + \alpha_{m+n+1}Tx_{m+n+1} - Tz_n\| \end{aligned}$$

$$\begin{aligned}
 &\leq (1 - \alpha_{m+n+1}) \|x_{m+n+1} - Tz_n\| + \alpha_{m+n+1} \|Tx_{m+n+1} - Tz_n\| \\
 &\leq (1 - \alpha_{m+n+1}) \|Ty_{m+n} - Tz_n\| + \alpha_{m+n+1} \|x_{m+n+1} - z_n\| \\
 &\leq (1 - \alpha_{m+n+1}) (\|Ty_{m+n} - Tz_{m+n}\| + \|Tz_{m+n} - Tx_{n+m}\| + \dots + \|Tx_{n+1} - Ty_n\| \\
 &\quad + \|Ty_n - Tz_n\|) + \alpha_{m+n+1} \|x_{m+n+1} - z_n\| \\
 &= (1 - \alpha_{m+n+1}) \sum_{k=3n-1}^{3n-2+p} \|Tu_{k+1} - Tu_k\| + \alpha_{m+n+1} \|Ty_{m+n} - (1 - \alpha_n)x_n - \alpha_n Tx_n\| \\
 &\leq (1 - \alpha_{m+n+1}) \sum_{k=3n-1}^{3n-2+p} \beta_k \|Tu_{3n-2} - u_{3n-2}\| \\
 &\quad + \alpha_{m+n+1} ((1 - \alpha_n) \|Ty_{m+n} - x_n\| + \alpha_n \|Ty_{m+n} - Tx_n\|) \\
 &\leq (1 - \alpha_{m+n+1}) \sum_{k=3n-1}^{3n-2+p} \beta_k \|Tu_{3n-2} - u_{3n-2}\| + \alpha_{m+n+1} (1 - \alpha_n) \|Ty_{m+n} - x_n\| \\
 &\quad + \alpha_{m+n+1} \alpha_n (\|Ty_{m+n} - Tz_{m+n}\| + \|Tz_{m+n} - Tx_{m+n}\| + \dots + \|Ty_n - Tz_n\| + \|Tz_n - Tx_n\|) \\
 &= (1 - \alpha_{m+n+1}) \sum_{k=3n-1}^{3n-2+p} \beta_k \|Tu_{3n-2} - u_{3n-2}\| + \alpha_{m+n+1} (1 - \alpha_n) \|Ty_{m+n} - x_n\| \\
 &\quad + \alpha_{m+n+1} \alpha_n \sum_{k=3n-2}^{3n-2+p} \|Tu_{k+1} - Tu_k\| \\
 &\leq (1 - \alpha_{m+n+1}) \sum_{k=3n-1}^{3n-2+p} \beta_k \|Tu_{3n-2} - u_{3n-2}\| + \alpha_{m+n+1} (1 - \alpha_n) \|Ty_{m+n} - x_n\| \\
 &\quad + \alpha_{m+n+1} \alpha_n \sum_{k=3n-2}^{3n-2+p} \beta_k \|Tu_k - u_k\| \\
 &= (1 - \alpha_{m+n+1}) \sum_{k=3n-1}^{3n-2+p} \beta_k \|Tu_{3n-2} - u_{3n-2}\| + \alpha_{m+n+1} (1 - \alpha_n) \|Tu_{3n-1+p} - u_{3n-2}\| \\
 &\quad + \alpha_{m+n+1} \alpha_n \sum_{k=3n-2}^{3n-2+p} \beta_k \|Tu_k - u_k\| \\
 &= (1 - \alpha_{m+n+1} + \alpha_{m+n+1} \alpha_n) \sum_{k=3n-2}^{3n-2+p} \beta_k \|Tu_{3n-2} - u_{3n-2}\| \\
 &\quad - \alpha_n (1 - \alpha_{m+n+1}) \|Tu_{3n-2} - u_{3n-2}\| + \alpha_{m+n+1} (1 - \alpha_n) \|Tu_{3n-1+p} - u_{3n-2}\|.
 \end{aligned}$$

It follows that

$$\begin{aligned}
 &\|Tu_{3n+1+p} - u_{3n+1}\| \\
 &\leq (1 - \alpha_{m+n+1} + \alpha_{m+n+1} \alpha_n) \sum_{k=3n-2}^{3n-2+p} \beta_k \|Tu_{3n-2} - u_{3n-2}\| \\
 &\quad - \alpha_n (1 - \alpha_{m+n+1}) \|Tu_{3n-2} - u_{3n-2}\| + \alpha_{m+n+1} (1 - \alpha_n) \|Tu_{3n-1+p} - u_{3n-2}\|
 \end{aligned} \tag{9}$$

Using (5) and (9), we have

$$\begin{aligned}
 &\left(1 + \sum_{k=3n+1}^{3n+p} \beta_k\right) \|Tu_{3n+1} - u_{3n+1}\| \\
 &\leq (1 - \alpha_{m+n+1} + \alpha_{m+n+1} \alpha_n) \sum_{k=3n-2}^{3n-2+p} \beta_k \|Tu_{3n-2} - u_{3n-2}\|
 \end{aligned}$$

$$-\alpha_n(1 - \alpha_{m+n+1})\|Tu_{3n-2} - u_{3n-2}\| + \alpha_{m+n+1}(1 - \alpha_n)\|Tu_{3n-1+p} - u_{3n-2}\|$$

$$+ \left(\prod_{k=n+1}^{n+p} \frac{2}{1 - \alpha_k} \right) (\|Tu_{3n+1} - u_{3n+1}\| - \|Tu_{3n+1+3p} - u_{3n+1+3p}\|).$$

From $\left(1 + \sum_{k=3n+1}^{3n+p} \beta_k\right) \leq \left(\prod_{k=n+1}^{n+p} \frac{1}{1 - \alpha_k}\right)$ and $\|Tu_{3n+1} - u_{3n+1}\| \leq \|Tu_{3n-2} - u_{3n-2}\|$, we have

$$\left(1 + \sum_{k=3n+1}^{3n+p} \beta_k\right) \|Tu_{3n-2} - u_{3n-2}\|$$

$$\leq (1 - \alpha_{m+n+1} + \alpha_{m+n+1}\alpha_n) \sum_{k=3n-2}^{3n-2+p} \beta_k \|Tu_{3n-2} - u_{3n-2}\|$$

$$-\alpha_n(1 - \alpha_{m+n+1})\|Tu_{3n-2} - u_{3n-2}\| + \alpha_{m+n+1}(1 - \alpha_n)\|Tu_{3n-1+p} - u_{3n-2}\|$$

$$+ \left(\prod_{k=n+1}^{n+p} \frac{2}{1 - \alpha_k} \right) (\|Tu_{3n-2} - u_{3n-2}\| - \|Tu_{3n+1+3p} - u_{3n+1+3p}\|).$$

Then

$$\frac{\left(1 + \sum_{k=3n+1}^{3n+p} \beta_k + \alpha_n(1 - \alpha_{m+n+1}) - (1 - \alpha_{m+n+1} + \alpha_{m+n+1}\alpha_n) \sum_{k=3n-2}^{3n-2+p} \beta_k\right)}{\alpha_{m+n+1}(1 - \alpha_n)}$$

$$\|Tu_{3n-2} - u_{3n-2}\| \leq \|Tu_{3n-1+p} - u_{3n-2}\|$$

$$+ \left(\prod_{k=n}^{n+p} \frac{2}{1 - \alpha_k} \right) (\|Tu_{3n-2} - u_{3n-2}\| - \|Tu_{3n+1+3p} - u_{3n+1+3p}\|).$$

It follows that

$$\left(1 + \sum_{k=3n-2}^{3n-2+p} \beta_k\right) \|Tu_{3n-2} - u_{3n-2}\|$$

$$\leq \|Tu_{3n-1+p} - u_{3n-2}\|$$

$$+ \left(\prod_{k=n}^{n+p} \frac{2}{1 - \alpha_k} \right) (\|Tu_{3n-2} - u_{3n-2}\| - \|Tu_{3n+1+3p} - u_{3n+1+3p}\|).$$

Thus, for $n, p + 1$, (4) holds.

Case 3: $p = 3m + 2, m \geq 0$. From (6) and (7), we have

$$\|Tu_{3n+1+p} - u_{3n+1}\|$$

$$= \|Ty_{n+m+1} - Ty_n\|$$

$$\leq \|y_{n+m+1} - y_n\|$$

$$= \|Tz_{n+m+1} - Tz_n\|$$

$$\leq \|z_{n+m+1} - z_n\|$$

$$\leq \|z_{n+m+1} - (1 - \alpha_n)x_n - \alpha_n Tx_n\|$$

$$\leq (1 - \alpha_n)\|z_{n+m+1} - x_n\| + \alpha_n\|z_{n+m+1} - Tx_n\|$$

$$= (1 - \alpha_n)\|(1 - \alpha_{n+m+1})x_{n+m+1} + \alpha_{n+m+1}Tx_{n+m+1} - x_n\|$$

$$+ \alpha_n\|(1 - \alpha_{n+m+1})x_{n+m+1} + \alpha_{n+m+1}Tx_{n+m+1} - Tx_n\|$$

$$\begin{aligned}
 &\leq (1 - \alpha_n)((1 - \alpha_{m+n+1})\|x_{n+m+1} - x_n\| + \alpha_{m+n+1}\|Tx_{n+m+1} - x_n\|) \\
 &+ \alpha_n((1 - \alpha_{m+n+1})\|x_{n+m+1} - Tx_n\| + \alpha_{m+n+1}\|Tx_{n+m+1} - Tx_n\|) \\
 &\leq (1 - \alpha_n)\alpha_{m+n+1}\|Tx_{n+m+1} - x_n\| \\
 &+ ((1 - \alpha_{m+n+1})(1 - \alpha_n) + \alpha_{m+n+1}\alpha_n)\|x_{n+m+1} - x_n\| \\
 &+ \alpha_n(1 - \alpha_{m+n+1})\|Ty_{n+m+1} - Tx_n\| \\
 &\leq (1 - \alpha_n)\alpha_{m+n+1}\|Tx_{n+m+1} - x_n\| \\
 &+ ((1 - \alpha_{m+n+1})(1 - \alpha_n) + \alpha_{m+n+1}\alpha_n)(\|Ty_{n+m} - Tz_{n+m}\| + \dots + \|Tz_n - Tx_n\| + \|Tx_n - x_n\|) \\
 &+ \alpha_n(1 - \alpha_{m+n+1})(\|Ty_{n+m+1} - Tz_{n+m+1}\| + \dots + \|Ty_n - Tz_n\| + \|Tz_n - Tx_n\|) \\
 &\leq (1 - \alpha_n)\alpha_{m+n+1}\|Tx_{n+m+1} - x_n\| \\
 &+ ((1 - \alpha_{m+n+1})(1 - \alpha_n) + \alpha_{m+n+1}\alpha_n) \left(\sum_{k=3n-2}^{3n-3+p} \|Tu_{k+1} - Tu_k\| + \|Tu_{3n-2} - x_{3n-2}\| \right) \\
 &+ (1 - \alpha_{m+n+1})\alpha_n \sum_{k=3n-2}^{3n-3+p} \|Tu_{k+1} - Tu_k\| \\
 &\leq (1 - \alpha_n)\alpha_{m+n+1}\|Tx_{n+m+1} - x_n\| \\
 &+ ((1 - \alpha_{m+n+1})(1 - \alpha_n) + \alpha_{m+n+1}\alpha_n) \sum_{k=3n-2}^{3n-2+p} \beta_k \|Tx_n - x_n\| \\
 &+ (1 - \alpha_{m+n+1})\alpha_n \sum_{k=3n-2}^{3n-3+p} \beta_k \|Tx_n - x_n\| \\
 &\leq (1 - \alpha_n)\alpha_{m+n+1}\|Tu_{3n-1+p} - u_{3n-2}\| \\
 &+ \left((1 - \alpha_{m+n+1} + \alpha_n\alpha_{m+n+1}) \sum_{k=3n-2}^{3n-2+p} \beta_k - \alpha_n(1 - \alpha_{m+n+1}) \right) \|Tu_{3n-2} - u_{3n-2}\|.
 \end{aligned}$$

It follows that

$$\begin{aligned}
 &\|Tu_{3n+1+p} - u_{3n+1}\| \\
 &\leq (1 - \alpha_n)\alpha_{m+n+1}\|Tu_{3n-1+p} - u_{3n-2}\| \\
 &+ \left((1 - \alpha_{m+n+1} + \alpha_n\alpha_{m+n+1}) \sum_{k=3n-2}^{3n-2+p} \beta_k - \alpha_n(1 - \alpha_{m+n+1}) \right) \|Tu_{3n-2} - u_{3n-2}\|
 \end{aligned} \tag{10}$$

Using (5) and (10), we have

$$\begin{aligned}
 &\left(1 + \sum_{k=3n+1}^{3n+p} \beta_k \right) \|Tu_{3n+1} - u_{3n+1}\| \\
 &\leq (1 - \alpha_n)\alpha_{m+n+1}\|Tu_{3n-1+p} - u_{3n-2}\| \\
 &+ \left((1 - \alpha_{m+n+1} + \alpha_n\alpha_{m+n+1}) \sum_{k=3n-2}^{3n-2+p} \beta_k - \alpha_n(1 - \alpha_{m+n+1}) \right) \|Tu_{3n-2} - u_{3n-2}\| \\
 &+ \left(\prod_{k=n+1}^{n+p} \frac{2}{1 - \alpha_k} \right) (\|Tu_{3n+1} - u_{3n+1}\| - \|Tu_{3n+1+3p} - u_{3n+1+3p}\|).
 \end{aligned}$$

From $\left(1 + \sum_{k=3n+1}^{3n+p} \beta_k \right) \leq \left(\prod_{k=n+1}^{n+p} \frac{2}{1 - \alpha_k} \right)$ and $\|Tu_{3n+1} - u_{3n+1}\| \leq \|Tu_{3n-2} - u_{3n-2}\|$, we have

$$\begin{aligned}
 & \left(1 + \sum_{k=3n+1}^{3n+p} \beta_k\right) \|Tu_{3n-2} - u_{3n-2}\| \\
 & \leq (1 - \alpha_n)\alpha_{m+n+1} \|Tu_{3n-1+p} - u_{3n-2}\| \\
 & + \left((1 - \alpha_{m+n+1} + \alpha_n\alpha_{m+n+1}) \sum_{k=3n-2}^{3n-2+p} \beta_k - \alpha_n(1 - \alpha_{m+n+1})\right) \|Tu_{3n-2} - u_{3n-2}\| \\
 & + \left(\prod_{k=n+1}^{n+p} \frac{2}{1 - \alpha_k}\right) (\|Tu_{3n-2} - u_{3n-2}\| - \|Tu_{3n+1+3p} - u_{3n+1+3p}\|).
 \end{aligned}$$

Then

$$\begin{aligned}
 & \frac{(1 + \sum_{k=3n+1}^{3n+p} \beta_k - ((1 - \alpha_{m+n+1} + \alpha_n\alpha_{m+n+1}) \sum_{k=3n-2}^{3n-2+p} \beta_k - \alpha_n(1 - \alpha_{m+n+1})))}{(1 - \alpha_n)\alpha_{m+n+1}} \|Tu_{3n-2} - u_{3n-2}\| \\
 & \leq \|Tu_{3n-1+p} - u_{3n-2}\| \\
 & + \left(\prod_{k=n}^{n+p} \frac{2}{1 - \alpha_k}\right) (\|Tu_{3n-2} - u_{3n-2}\| - \|Tu_{3n+1+3p} - u_{3n+1+3p}\|).
 \end{aligned}$$

It follows that

$$\begin{aligned}
 & \left(1 + \sum_{k=3n-2}^{3n-2+p} \beta_k\right) \|Tu_{3n-2} - u_{3n-2}\| \\
 & \leq \|Tu_{3n-1+p} - u_{3n-2}\| \\
 & + \left(\prod_{k=n}^{n+p} \frac{2}{1 - \alpha_k}\right) (\|Tu_{3n-2} - u_{3n-2}\| - \|Tu_{3n+1+3p} - u_{3n+1+3p}\|).
 \end{aligned}$$

Thus, for $n, p + 1$, (4) holds. This completes the induction.

Lemma 3. $T : C \rightarrow C$ is a $T - (D_a)$ mapping, $\|Tx - x\| \leq \|Ty - y\|$. Then

$$\|x - Ty\| \leq 3\|Tx - x\| + \|x - y\|.$$

Proof: Since $\|Tx - x\| \leq \|Ty - y\|$, we have $Tx \in C(T, y, \alpha)$. Then

$$\|T^2x - Ty\| \leq \|Tx - y\|.$$

It follows that

$$\|x - Ty\| \leq \|x - Tx\| + \|T^2x - Tx\| + \|T^2x - Ty\|.$$

From **Proposition 3**, we have

$$\|x - Ty\| \leq 2\|Tx - x\| + \|Tx - y\| \leq 2\|Tx - x\| + \|Tx - x\| + \|x - y\| = 3\|Tx - x\| + \|x - y\|.$$

Theorem 1. Assume that $T : C \rightarrow C$ is a $T - (D_a)$ mapping and $\{x_n\}, \{y_n\}, \{z_n\}$ are sequences generated by iteration (3), $\frac{1}{2} < a \leq \alpha_n \leq b < 1$. Then

$$\lim_{n \rightarrow \infty} \|Tx_n - x_n\| = 0.$$

Proof: Since C is bounded, there must exist $d > 0$, for every $x \in C$, $\|x\| \leq d$. Let $\{u_m\}$ satisfy $u_{3n-2} = x_n, u_{3n-1} = z_n, u_{3n} = y_n$. From Lemma 1, $\lim_{k \rightarrow \infty} \|Tu_k - u_k\| = r \geq 0$. Assume $r > 0$. Let ε satisfy

$$e^{\frac{6}{1-b}\left(\frac{d}{r}+1\right)} \varepsilon < r$$

and choose n so that for every $p > 0$

$$\|Tu_{3n-2} - u_{3n-2}\| - \|Tu_{3n-2+3p} - u_{3n-2+3p}\| < \varepsilon.$$

Now choose p so that $r \left(\sum_{k=3n-2}^{3n+p-4} \beta_k \right) \leq d \leq r \left(\sum_{k=3n-2}^{3n+p-3} \beta_k \right)$.

Since $\frac{1}{2} < a \leq \alpha_n \leq b < 1$, for every k, t , we have $1 + \alpha_k < 3\alpha_k$, $\alpha_t < 2\alpha_k$. From **Lemma 2** and $r \leq \|Tu_{3n-2} - u_{3n-2}\|$, we have

$$\begin{aligned} d + r &\leq r \left(1 + \sum_{k=3n-2}^{3n+p-3} \beta_k \right) \\ &\leq \left(1 + \sum_{k=3n-2}^{3n+p-3} \beta_k \right) \|Tu_{3n-2} - u_{3n-2}\| \\ &\leq \|Tu_{3n-2+p} - u_{3n-2}\| \\ &\quad + \left(\prod_{k=n}^{n+p-1} \frac{2}{1-\alpha_k} \right) (\|Tu_{3n-2} - u_{3n-2}\| - \|Tu_{3n-2+3p} - u_{3n-2+3p}\|) \\ &< d + \left(\prod_{k=n}^{n+p-1} \frac{2}{1-\alpha_k} \right) \varepsilon \\ &= d + e^{\sum_{k=n}^{n+p-1} \ln \left(1 + \frac{1+\alpha_k}{1-\alpha_k} \right)} \varepsilon \\ &\leq d + e^{\sum_{k=n}^{n+p-1} \frac{1+\alpha_k}{1-\alpha_k}} \varepsilon \\ &\leq d + e^{\frac{3}{1-b} \sum_{k=n}^{n+p-1} \alpha_k} \varepsilon \\ &\leq d + e^{\frac{6}{1-b} \sum_{k=3n-2}^{3n+p-3} \beta_k} \varepsilon \\ &\leq d + e^{\frac{6}{1-b} \left(\sum_{k=3n-2}^{3n+p-4} \beta_k + 1 \right)} \varepsilon \\ &< d + e^{\frac{6}{1-b} \left(\frac{d}{r} + 1 \right)} \varepsilon < d + r. \end{aligned}$$

This is a contradiction. So $\lim_{k \rightarrow \infty} \|Tu_k - u_k\| = 0$. That is to say, $\lim_{n \rightarrow \infty} \|Tx_n - x_n\| = 0$. This completes the proof.

Theorem 2. Assume that $T : C \rightarrow C$ is a $T - (D_a)$ mapping and $\{x_n\}$ is generated by iteration (3), $\frac{1}{2} < a \leq \alpha_n \leq b < 1$. Then the sequence $\{x_n\}$ converges to a fixed point of T .

Proof: Since C is compact, there exists a subsequence $\{x_{n_k}\} \subset \{x_n\}$ which converges to some $z \in C$. By **Lemma 3**, we have

$\|x_{n_k} - Tz\| \leq 3\|Tx_{n_k} - x_{n_k}\| + \|x_{n_k} - z\|$. Since $\lim_{n_k \rightarrow \infty} \|Tx_{n_k} - x_{n_k}\| = 0$ and $\lim_{n_k \rightarrow \infty} \|x_{n_k} - z\| = 0$, we have $\lim_{n_k \rightarrow \infty} \|x_{n_k} - Tz\| = 0$. This implies that $z = Tz$. On the other hand, from **Proposition 3**

$$\begin{aligned} \|x_{n+1} - z\| &\leq \|y_n - z\| \leq \|z_n - z\| \\ &\leq \alpha_n \|Tx_n - z\| + (1 - \alpha_n) \|x_n - z\| \\ &\leq \|x_n - z\|. \end{aligned}$$

So, $\lim_{n \rightarrow \infty} \|x_n - z\|$ exists. Therefore, $\lim_{n \rightarrow \infty} \|x_n - z\| = 0$. This completes the proof.



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Chapter 3

A Survey on Hilbert Spaces and Reproducing Kernels

Baver Okutmuştur

Abstract

The main purpose of this chapter is to provide a brief review of Hilbert space with its fundamental features and introduce reproducing kernels of the corresponding spaces. We separate our analysis into two parts. In the first part, the basic facts on the inner product spaces including the notion of norms, pre-Hilbert spaces, and finally Hilbert spaces are presented. The second part is devoted to the reproducing kernels and the related Hilbert spaces which is called the reproducing kernel Hilbert spaces (RKHS) in the complex plane. The operations on reproducing kernels with some important theorems on the Bergman kernel for different domains are analyzed in this part.

Keywords: Hilbert spaces, norm spaces, reproducing kernels, reproducing kernel Hilbert spaces (RKHS), operations on reproducing kernels, sesqui-analytic kernels, analytic functions, Bergman kernel

1. Framework

This chapter consists of introductory concept on the Hilbert space theory and reproducing kernels. We start by presenting basic definitions, propositions, and theorems from functional analysis related to Hilbert spaces. The notion of linear space, norm, inner product, and pre-Hilbert spaces are in the first part. The second part is devoted to the fundamental properties of the reproducing kernels and the related Hilbert spaces. The operations with reproducing kernels, inclusion property, Bergman kernel, and further properties with examples of the reproducing kernels are analyzed in the latter section.

2. Introduction to Hilbert spaces

We start by the definition of a vector space and related topics. Let \mathbb{C} be the complex field. The following preliminaries can be considered as fundamental concepts of the Hilbert spaces.

2.1 Vector spaces and inner product spaces

Vector space. A vector space is a linear space that is closed under vector addition and scalar multiplication. More precisely, if we denote our linear space by \mathcal{H} over the field \mathbb{C} , then it follows that

i. if $x, y, z \in \mathcal{H}$, then

$$x + y = y + x \in \mathcal{H}, \quad x + (y + z) = (x + y) + z \in \mathcal{H};$$

ii. if k is scalar, then $kx \in \mathcal{H}$.

Inner product. Let \mathcal{H} be a linear space over the complex field \mathbb{C} . An *inner product* on \mathcal{H} is a two variable function

$$\langle \cdot, \cdot \rangle : \mathcal{H} \times \mathcal{H} \rightarrow \mathbb{C}, \text{ satisfying}$$

i. $\langle f, g \rangle = \overline{\langle g, f \rangle}$ for $f, g \in \mathcal{H}$.

ii. $\langle \alpha f + \beta g, h \rangle = \alpha \langle f, h \rangle + \beta \langle g, h \rangle$ and $\langle f, \alpha g + \beta h \rangle = \bar{\alpha} \langle f, g \rangle + \bar{\beta} \langle f, h \rangle$ for $\alpha, \beta \in \mathbb{C}$ and $f, g, h \in \mathcal{H}$.

iii. $\langle f, f \rangle \geq 0$ for $f \in \mathcal{H}$ and $\langle f, f \rangle = 0 \Leftrightarrow f = 0$.

Pre-Hilbert space. A *pre-Hilbert space* \mathcal{H} is a linear space over the complex field \mathbb{C} with an inner product defined on it.

Norm space or inner product space. A norm on an inner product space \mathcal{H} denoted by $\| \cdot \|$ is defined by

$$\|f\| = \langle f, f \rangle^{1/2} \text{ or } \|f\|_{\mathcal{H}} = \langle f, f \rangle_{\mathcal{H}}^{1/2}$$

where $f \in \mathcal{H}$ and $\langle \cdot, \cdot \rangle = \langle \cdot, \cdot \rangle_{\mathcal{H}}$ denote the inner product on \mathcal{H} . The corresponding space is called as the inner product space or the norm space.

Properties of norm. For all $f, g \in \mathcal{H}$, and $\lambda \in \mathbb{C}$, we have

- $\|f\| \geq 0$. (Observe that the equality occurs only if $f = 0$).
- $\|\lambda f\| = |\lambda| \|f\|$.

Schwarz inequality. For all $f, g \in \mathcal{H}$, it follows that

$$|\langle f, g \rangle| \leq \|f\| \|g\|. \quad (1)$$

In case if f and g are linearly dependent, then the inequality becomes equality.

Triangle inequality. For all $f, g \in \mathcal{H}$, it follows that

$$\|f + g\| \leq \|f\| + \|g\|. \quad (2)$$

In case if f and g are linearly dependent, then the inequality becomes equality.

Polarization identity. For all $f, g \in \mathcal{H}$, it follows that

$$\langle f, g \rangle = \frac{1}{4} (\|f + g\|^2 - \|f - g\|^2 + i\|f + ig\|^2 - \|f - ig\|^2) \text{ for } f, g \in \mathcal{H}. \quad (3)$$

Parallelogram identity. For all $f, g \in \mathcal{H}$, it follows that

$$\|f + g\|^2 + \|f - g\|^2 = 2\|f\|^2 + 2\|g\|^2. \quad (4)$$

Metric. A metric on a set X is a function $d: X \times X \rightarrow \mathbb{R}$ satisfying the properties.

- $d(x, y) \geq 0$ and $d(x, y) = 0$ only if $x = y$;
- $d(x, y) = d(y, x)$;
- $d(x, y) \leq d(x, z) + d(z, y)$;

for all $x, y, z \in X$. Moreover the space (X, d) is the associated metric space. If we rearrange the metric with its properties for the inner product space \mathcal{H} , then it follows that for all $f, g, h \in \mathcal{H}$ and for all $\lambda \in \mathbb{C}$, where d satisfies all requirements to be a metric, we have

- $d(f, g) \geq 0$ and equality occurs only if $f = g$.
- $d(f, g) = d(g, f)$.
- $d(f, g) \leq d(f, h) + d(h, g)$.
- $d(f - h, g - h) = d(f, g)$.
- $d(\lambda f, \lambda g) = |\lambda| \cdot d(f, g)$.

Note. The binary function d given in the metric definition above represents the metric topology in \mathcal{H} which is called *strong topology* or *norm topology*. As a result, a sequence $(f_n)_{n \geq 0}$ in the pre-Hilbert space \mathcal{H} converges strongly to f if the condition

$$\|f_n - f\| \rightarrow 0 \text{ whenever } n \rightarrow \infty$$

is satisfied.

2.2 Introduction to linear operators

Linear operator. A map L from a linear space to another linear space is called *linear operator* if

$$L(\alpha f + \beta g) = \alpha Lf + \beta Lg$$

is satisfied for all $\alpha, \beta \in \mathbb{C}$ and for all $f, g \in \mathcal{H}$.

Continuous operator. An operator L is said to be continuous if it is continuous at each point of its domain. Notice that the domain and range spaces must be convenient for appropriate topologies.

Lipschitz constant of a linear operator. If L is a linear operator from \mathcal{H} to \mathcal{G} where \mathcal{H} and \mathcal{G} are pre-Hilbert spaces, then the Lipschitz constant for L is its norm $\|L\|$ and it is defined by

$$\|L\| = \sup\{\|Lf\|_{\mathcal{G}} / \|f\|_{\mathcal{H}} : 0 \neq f \in \mathcal{H}\}. \quad (5)$$

Theorem 1. Let L be a linear operator from the pre-Hilbert spaces \mathcal{H} to \mathcal{G} . Then the followings are mutually equivalent:

- L is continuous.
- L is bounded, that is,

$$\sup\{\|Lf\|_{\mathcal{G}} : \|f\|_{\mathcal{H}} \leq k\} < \infty$$

for $0 \leq k < \infty$.

iii. L is Lipschitz continuous, that is,

$$\|Lf - Lg\|_{\mathcal{G}} \leq \lambda \|f - g\|_{\mathcal{H}},$$

where $0 \leq \lambda < \infty$ and $f, g \in \mathcal{H}$.

Some properties of linear operators. Let $B(\mathcal{H}, \mathcal{G})$ be the collection of all continuous linear operators from the pre-Hilbert spaces \mathcal{H} to \mathcal{G} . Then

- $B(\mathcal{H}, \mathcal{G})$ is a linear space with respect to the natural addition and scalar multiplication satisfying

$$(\alpha L + \beta M)f = \alpha Lf + \beta Mf,$$

where L and M are linear operators, $f \in \mathcal{H}$ and $\alpha, \beta \in \mathbb{C}$.

- Whenever $\mathcal{H} = \mathcal{G}$, then $B(\mathcal{H}, \mathcal{G})$ is denoted by $B(\mathcal{H})$.
- If \mathcal{K} is another pre-Hilbert space, $L \in B(\mathcal{H}, \mathcal{G})$ and $K \in B(\mathcal{G}, \mathcal{K})$. Then the product

$$(KL)f = K(Lf) \text{ for } f \in \mathcal{H} \in B(\mathcal{H}, \mathcal{K}).$$

In addition,

- $K(\xi L + \zeta M) = \xi KL + \zeta KM$
- $\|\xi L\| = |\xi| \cdot \|L\|$
- $\|L + M\| \leq \|L\| + \|M\|$ and
- $\|KL\| \leq \|K\| \|L\|$.

are also satisfied.

2.3 Hilbert spaces and linear operators

Linear form (or linear functional). A linear operator from the pre-Hilbert space \mathcal{H} to the scalar field \mathbb{C} is called a *linear form* (or *linear functional*).

Hilbert spaces. A pre-Hilbert space \mathcal{H} is said to be a *Hilbert space* if it is complete in metric. In other words if f_n is a Cauchy sequence in \mathcal{H} , that is, if

$$\|f_n - f_m\| \rightarrow 0 \text{ whenever } n, m \rightarrow \infty,$$

then there is $f \in \mathcal{H}$ such that

$$\|f_n - f\| \rightarrow 0 \text{ whenever } n \rightarrow \infty.$$

Note. Every subspace of a pre-Hilbert space is also a pre-Hilbert space with respect to the induced inner product. However, the reverse is not always true. For a subspace of a Hilbert space to be also a Hilbert space, it must be closed.

Completion. The canonical method for which a pre-Hilbert space \mathcal{H} is embedded as a dense subspace of a Hilbert space $\tilde{\mathcal{H}}$ so that

$$\langle f, g \rangle_{\tilde{\mathcal{H}}} = \langle f, g \rangle_{\mathcal{H}} \text{ for } f, g \in \mathcal{H}$$

is called *completion*.

Note. If L is a continuous linear operator from a dense subspace \mathcal{M} of a Hilbert space \mathcal{H} to a Hilbert space \mathcal{G} , then it can be extended uniquely to a continuous linear operator from \mathcal{H} to \mathcal{G} with preserving norm.

Theorem 2. Let \mathcal{M} and \mathcal{N} be dense subspaces of the Hilbert spaces \mathcal{H} and \mathcal{G} , respectively. For $f \in \mathcal{H}, g \in \mathcal{M}$ and $0 \leq \lambda < \infty$, if a linear operator L from \mathcal{M} to \mathcal{G} satisfies

$$|\langle Lf, g \rangle_{\mathcal{G}}| \leq \lambda \|f\|_{\mathcal{H}} \|g\|_{\mathcal{G}}, \quad (6)$$

then L is uniquely extended to a continuous linear operator from \mathcal{M} to \mathcal{G} with norm $\leq \lambda$ where the norm coincides with the minimum of such λ .

Theorem 3. Let (Ω, μ) denotes a measure space so that Ω is the union of subsets of finite positive measure and $L^2(\Omega, \mu)$ consists of all measurable functions $f(\omega)$ on Ω such that

$$\int_{\Omega} |f(\omega)|^2 d\mu(\omega) < \infty. \quad (7)$$

Then $L^2(\Omega, \mu)$ is a Hilbert space with respect to the inner product

$$\langle f, g \rangle := \int_{\Omega} f(\omega) \overline{g(\omega)} d\mu(\omega). \quad (8)$$

Theorem 4 (F. Riesz). For each continuous linear functional φ on a Hilbert space \mathcal{H} , there exists uniquely $g \in \mathcal{H}$ such that

$$\varphi(f) = \langle f, g \rangle \text{ for } f \in \mathcal{H}. \quad (9)$$

Theorem 5. Let \mathcal{M} be a closed subspace of a Hilbert space \mathcal{H} . Then the algebraic direct sum relation

$$\mathcal{H} = \mathcal{M} \oplus \mathcal{M}^{\perp}$$

is satisfied. In other words, $\forall f \in \mathcal{H}$ can be uniquely written by

$$f = f_{\mathcal{M}} + f_{\mathcal{M}^{\perp}} \text{ with } f_{\mathcal{M}} \in \mathcal{M}, f_{\mathcal{M}^{\perp}} \in \mathcal{M}^{\perp}. \quad (10)$$

In addition, $\|f_{\mathcal{M}}\|$ coincides with the distance from f to \mathcal{M}^{\perp}

$$\|f_{\mathcal{M}}\| = \min \{ \|f - g\| : g \in \mathcal{M}^{\perp} \}. \quad (11)$$

Remark. In a Hilbert space, the closed linear span of any subset A of a Hilbert space \mathcal{H} coincides with $(A^{\perp})^{\perp}$.

Total subset of a Hilbert space. A subset \mathcal{A} of a Hilbert space \mathcal{H} is called *total* in \mathcal{H} if 0 is the only element that is orthogonal to all elements of \mathcal{A} . In other words,

$$\mathcal{A}^{\perp} = \{0\}.$$

As a result, \mathcal{A} is total if and only if every element of \mathcal{H} can be approximated by linear combinations of elements of \mathcal{A} .

Orthogonal projection. If \mathcal{M} is a closed subspace of \mathcal{H} , the map $f \mapsto f_{\mathcal{M}}$ gives a linear operator from \mathcal{H} to \mathcal{M} with norm ≤ 1 . We call this operator as the *orthogonal projection* to \mathcal{M} and denote it by $P_{\mathcal{M}}$.

Note. If I is the identity operator on \mathcal{H} , then $I - P_{\mathcal{M}}$ denotes the orthogonal projection to \mathcal{M}^\perp , and the relation

$$\|f\|^2 = \|P_{\mathcal{M}}f\|^2 + \|(I - P_{\mathcal{M}})f\|^2 \quad (12)$$

is satisfied for all $f \in \mathcal{H}$.

Weak topology. The weakest topology that makes continuous all linear functionals of the form $f \mapsto \langle f, g \rangle$ is called *the weak topology* of a Hilbert space \mathcal{H} .

Note. If $f \in \mathcal{H}$, then with respect to the weak topology, a fundamental system of neighborhoods of f is composed of subsets of the form

$$U(f; A, \epsilon) = \{h : |\langle f, g \rangle - \langle h, g \rangle| < \epsilon \text{ for } g \in A\},$$

where A is a finite subset of \mathcal{H} and $\epsilon > 0$. Then a directed net $\{f_\lambda\}$ converges weakly to f if and only if

$$\langle f_\lambda, g \rangle \xrightarrow{\lambda} \langle f, g \rangle \text{ for all } g \in \mathcal{H}.$$

Operator weak topology. The weakest topology that makes continuous all linear functionals of the form

$$L \mapsto \langle Lf, g \rangle \text{ for } f \in \mathcal{H}, g \in \mathcal{G}$$

is called *the operator weak topology* in the space $B(\mathcal{H}, \mathcal{G})$ of continuous linear operators from \mathcal{H} to \mathcal{G} . In addition, a directed net $\{L_\lambda\}$ converges weakly to L if

$$\langle L_\lambda f, g \rangle \xrightarrow{\lambda} \langle Lf, g \rangle.$$

Operator strong topology. The weakest topology that makes continuous all linear operators of the form

$$L \mapsto Lf \text{ for } f \in \mathcal{H}$$

is called *the operator strong topology*. Moreover a directed net $\{L_\lambda\}$ converges strongly to L if

$$\|L_\lambda f - Lf\| \xrightarrow{\lambda} 0 \text{ for all } f \in \mathcal{H}.$$

Theorem 6. Let \mathcal{H} and \mathcal{G} be Hilbert spaces and $B(\mathcal{H}, \mathcal{G})$ be a continuous linear operator from \mathcal{H} to \mathcal{G} . Then

- the closed unit ball $U := \{f : \|f\| \leq 1\}$ of \mathcal{H} is weakly compact;
- the closed unit ball $\{L : \|L\| \leq 1\}$ of $B(\mathcal{H}, \mathcal{G})$ is weakly compact.

Theorem 7. Let \mathcal{H} be a Hilbert space and $A \subseteq \mathcal{H}$. Then if A is weakly bounded in the sense

$$\sup_{f \in A} |\langle f, g \rangle| < \infty \text{ for } g \in \mathcal{H}, \quad (13)$$

then it is strongly bounded, that is, $\sup_{f \in A} \|f\| < \infty$.

Theorem 8. If \mathcal{H} and \mathcal{G} are Hilbert spaces and L is a linear operator from \mathcal{H} to \mathcal{G} , then the strong continuity and weak continuity for L are equivalent.

Theorem 9. Let \mathcal{H} and \mathcal{G} be Hilbert spaces. Then the following statements for $\mathbf{L} \subseteq B(\mathcal{H}, \mathcal{G})$ are mutually equivalent:

(i) \mathbf{L} is weakly bounded; that is, for $f \in \mathcal{H}$, $g \in \mathcal{G}$, we have

$$\sup_{L \in \mathbf{L}} |\langle Lf, g \rangle| < \infty$$

(ii) \mathbf{L} is strongly bounded; that is, for $f \in \mathcal{H}$, we have

$$\sup_{L \in \mathbf{L}} \|Lf\| < \infty.$$

(iii) \mathbf{L} is norm bounded (or uniformly bounded); that is,

$$\sup_{L \in \mathbf{L}} \|L\| < \infty.$$

Theorem 10. A linear operator L from the Hilbert spaces \mathcal{H} to \mathcal{G} is said to be *closed* if its *graph*

$$G_L := \{f \oplus Lf : f \in \mathcal{H}\} \quad (14)$$

is a closed subspace of the direct sum space $\mathcal{H} \oplus \mathcal{G}$, that is, whenever $n \rightarrow \infty$,

$$\|f_n - f\| \rightarrow 0 \text{ in } \mathcal{H} \text{ and } \|Lf_n - g\| \rightarrow 0 \text{ in } \mathcal{G} \Rightarrow g = Lf.$$

Theorem 11. If L is a closed linear operator with a domain of a Hilbert space \mathcal{H} to another Hilbert space \mathcal{G} , then it is continuous.

Sesqui-linear form. A function $\Phi : \mathcal{H} \times \mathcal{G} \rightarrow \mathbb{C}$ is a *sesqui-linear form* (or *sesqui-linear function*) if for $f, h \in \mathcal{H}$, $g, k \in \mathcal{G}$ and $\alpha, \beta \in \mathbb{C}$,

$$(i) \quad \Phi(\alpha f + \beta h, g) = \alpha \Phi(f, g) + \beta \Phi(h, g) \quad (15)$$

$$(ii) \quad \Phi(f, \alpha g + \beta k) = \bar{\alpha} \Phi(f, g) + \bar{\beta} \Phi(f, k) \quad (16)$$

are satisfied where \mathcal{H} and \mathcal{G} are Hilbert spaces.

Remark. If $L \in B(\mathcal{H}, \mathcal{G})$, then the sesqui-linear form Φ defined by

$$\Phi(f, g) = \langle Lf, g \rangle_{\mathcal{G}} \quad (17)$$

is bounded in the sense that

$$|\Phi(f, g)| \leq \lambda \|f\|_{\mathcal{H}} \|g\|_{\mathcal{G}} \text{ for } f \in \mathcal{H}, g \in \mathcal{G}, \quad (18)$$

where $\lambda \geq \|L\|$.

Remark. If a sesqui-linear form Φ satisfies the condition (18), then for $f \in \mathcal{H}$, the linear functional

$$g \mapsto \overline{\Phi(f, g)}$$

is continuous on \mathcal{G} . If we apply the Riesz theorem, then there exists uniquely $f' \in \mathcal{G}$ satisfying

$$\|f'\|_{\mathcal{G}} \leq \lambda \|f\|_{\mathcal{H}} \text{ and } \Phi(f, g) = \langle f', g \rangle_{\mathcal{G}} \text{ for } g \in \mathcal{G}.$$

Hence $f \mapsto f'$ becomes linear, and as a result we obtain

$$\Phi(f, g) = \langle f', g \rangle_{\mathcal{G}} = \langle Lf, g \rangle_{\mathcal{G}}.$$

Adjoint operator. If $L \in B(\mathcal{H}, \mathcal{G})$, then the unique operator $L^* \in B(\mathcal{G}, \mathcal{H})$ satisfying

$$\Phi(f, g) = \langle f, L^*g \rangle_{\mathcal{H}} \text{ for } f \in \mathcal{H}, g \in \mathcal{G} \tag{19}$$

is called the *adjoint* of L .

Remark. By the definitions of L and L^* , it follows that

$$\langle Lf, g \rangle_{\mathcal{G}} = \langle f, L^*g \rangle_{\mathcal{H}} \text{ for } f \in \mathcal{H}, g \in \mathcal{G}. \tag{20}$$

Isometric property. The adjoint operation is isometric if

$$\|L\| = \|L^*\| \text{ is satisfied.} \tag{21}$$

Remark. Let \mathcal{H}, \mathcal{G} , and \mathcal{K} be Hilbert spaces and $K \in B(\mathcal{G}, \mathcal{K})$ and $L \in B(\mathcal{H}, \mathcal{G})$ be given. Then

$$KL \in B(\mathcal{H}, \mathcal{K}) \text{ and } (KL)^* = L^*K^* \tag{22}$$

$$\text{Ker}(L) = (\text{Ran}(L^*))^\perp \text{ and } (\text{Ker}(L))^\perp = \text{Clos}\{\text{Ran}(L)^*\} \tag{23}$$

where $\text{Ker}(L)$ is the kernel of L and $\text{Ran}(L)$ is the range of L .

Theorem 12. If $L, M \in B(\mathcal{H}, \mathcal{G})$, then the following statements are mutually equivalent.

- i. $\text{Ran}(M) \subseteq \text{Ran}(L)$.
- ii. There exists $K \in B(\mathcal{H})$ such that $M = LK$.
- iii. There exists $0 \leq \lambda < \infty$ such that

$$\|M^*g\| \leq \lambda \|L^*g\| \text{ for } g \in \mathcal{G}.$$

Quadric form. Let \mathcal{H} be a Hilbert space. A function

$$\varphi : \mathcal{H} \rightarrow \mathbb{C}$$

is a *quadratic form* if for all $f \in \mathcal{H}$ and $\zeta \in \mathbb{C}$,

$$\varphi(\zeta f) = |\zeta|^2 \varphi(f) \tag{24}$$

and

$$\varphi(f + g) + \varphi(f - g) = 2\{\varphi(f) + \varphi(g)\} \tag{25}$$

are satisfied.

Note. If $L \in B(\mathcal{H})$, the quadratic form φ on \mathcal{H} is defined by

$$\varphi(f) = \langle Lf, f \rangle \text{ for } f \in \mathcal{H}, \quad (26)$$

and it is bounded

$$|\varphi(f)| \leq \lambda \|f\|^2 \text{ for } f \in \mathcal{H}, \quad (27)$$

where $\lambda \geq \|L\|$.

Remark. The sesqui-linear form Φ associated with L can be recovered from the quadratic form φ by the equation

$$\Phi(f, g) = \frac{1}{4} \{ \varphi(f + g) - \varphi(f - g) \} + \{ \varphi(f + ig) - \varphi(f - ig) \} \quad (28)$$

for all $f, g \in \mathcal{H}$.

Self-adjoint operator. A continuous linear operator L on a Hilbert space \mathcal{H} is said to be *self-adjoint* if $L = L^*$.

Remark. L is self-adjoint if and only if the associated sesqui-linear form Φ is Hermitian.

Remark. If L is self-adjoint, then the norm of L coincides with the minimum of λ given in (27) for the related quadratic form

Theorem 13. If L is a continuous self-adjoint operator, then

$$\|L\| = \sup \{ |\langle Lf, f \rangle| : \|f\| \leq 1 \}. \quad (29)$$

Positive definite operator. A self-adjoint operator $L \in B(\mathcal{H})$ is said to be *positive* (or *positive definite*) if

$$\langle Lf, f \rangle \geq 0 \text{ for all } f \in \mathcal{H}.$$

If $\langle Lf, f \rangle = 0$ only when $f = 0$, then L is said to be *strictly positive* (or, *strictly positive definite*).

Note. For any positive operator $L \in B(\mathcal{H})$, the Schwarz inequality holds in the following sense

$$|\langle Lf, g \rangle|^2 \leq \langle Lf, f \rangle \cdot \langle Lg, g \rangle. \quad (30)$$

Theorem 14. Let L and M be continuous positive operators on \mathcal{H} and \mathcal{G} , respectively. Then a continuous linear operator K from \mathcal{H} to \mathcal{G} satisfies the inequality

$$|\langle Kf, g \rangle_{\mathcal{G}}|^2 \leq \langle Lf, f \rangle_{\mathcal{H}} \langle Mg, g \rangle_{\mathcal{G}} \text{ for } f \in \mathcal{H}, g \in \mathcal{G} \quad (31)$$

if and only if the continuous linear operator

$$\begin{bmatrix} L & K^* \\ K & M \end{bmatrix}$$

on the direct sum Hilbert space $\mathcal{H} \oplus \mathcal{G}$ with

$$f \oplus g \mapsto (Lf + K^*g) \oplus (Kf + Mg)$$

is positive definite.

Theorem 15. Let L be a continuous positive definite operator. Then there exists a unique positive definite operator called the *square root* of L , denoted by $L^{1/2}$, such that $(L^{1/2})^2 = L$.

Modulus operator. The square root of the positive definite operator L^*L is called the *modulus (operator)* of L if L is a continuous linear operator.

Isometry. A linear operator U between Hilbert spaces \mathcal{H} and \mathcal{G} is called *isometric* or an *isometry* if

$$\|Uf\|_{\mathcal{G}} = \|f\|_{\mathcal{H}} \text{ for } f \in \mathcal{H} \quad (32)$$

is satisfied, that is, it preserves the norm.

Note. Eq. (32) implies that a continuous linear operator U is isometric if and only if $U^*U = I_{\mathcal{H}}$; in other words,

$$\langle Uf, Ug \rangle_{\mathcal{G}} = \langle f, g \rangle_{\mathcal{H}} \text{ for } f, g \in \mathcal{H}, \quad (33)$$

that is, U preserves the inner product.

Unitary operator. A surjective isometry linear operator $U : \mathcal{H} \rightarrow \mathcal{H}$ is called a *unitary (operator)*.

Note. Observe that if $U \in B(H)$ is a unitary operator, then $U^* = U^{-1}$.

Partial isometry. A continuous linear operator U between Hilbert spaces \mathcal{H} and \mathcal{G} is called a *partial isometry* if

$$f \in (\text{Ker}U)^{\perp} = \text{Ran}(U^*) \Rightarrow \|Uf\| = \|f\|.$$

The spaces $(\text{Ker}U)^{\perp}$ and $\text{Ran}(U)$ are called the *initial space* of U and the *final space* of U , respectively.

Note. If U is a partial isometry, then its adjoint U^* is also a partial isometry.

Theorem 17. Every continuous linear operator L on \mathcal{H} admits a unique decomposition

$$L = U\tilde{L}, \quad (34)$$

where \tilde{L} is a positive definite operator and U is a partial isometry with initial space the closure of $\text{Ran}(\tilde{L})$.

3. Reproducing kernels and RKHS

We continue our analysis on the abstract theory of reproducing kernels.

3.1 Definition and fundamental properties

Reproducing kernels. Let \mathcal{H} be a Hilbert space of functions on a nonempty set X with the inner product $\langle f, g \rangle$ and norm $\|f\| = \langle f, f \rangle^{1/2}$ for f and $g \in \mathcal{H}$. Then the complex valued function $K(y, x)$ of y and x in X is called a *reproducing kernel* of \mathcal{H} if

i. For all $x \in X$, it follows that $K_x(\cdot) = K(\cdot, x) \in \mathcal{H}$,

ii. For all $x \in X$ and all $f \in \mathcal{H}$,

$$f(x) = \langle f, K_x \rangle, \quad (35)$$

are satisfied.

Note. Let K be a reproducing kernel. Applying (35) to the function K_x at y , we get

$$K_x(y) = K(y, x) = \langle K_y, K_x \rangle, \text{ for } x, y \in X. \quad (36)$$

Then, for any $x \in X$, we obtain

$$\|K_x\| = \langle K_x, K_x \rangle^{1/2} = K(x, x)^{1/2}. \quad (37)$$

Note. Observe that the subset $\{K_x\}_{x \in X}$ is total in \mathcal{H} , that is, its closed linear span coincides with \mathcal{H} . This follows from the fact that, if $f \in \mathcal{H}$ and $f \perp K_x$ for all $x \in X$, then

$$f(x) = \langle f, K_x \rangle = 0 \text{ for all } x \in X,$$

and hence f is the 0 element in \mathcal{H} . As a result, $\{0\}^\perp = \mathcal{H}$.

RKHS. A Hilbert space \mathcal{H} of functions on a set X is called a RKHS if there exists a reproducing kernel K of \mathcal{H} .

Theorem 18. If a Hilbert space \mathcal{H} of functions on a set X admits a reproducing kernel K , then this reproducing kernel K is unique.

Theorem 19. There exists a reproducing kernel K for \mathcal{H} if and only if for all $x \in X$, the linear functional $\mathcal{H} \ni f \mapsto f(x)$ of evaluation at x is bounded on \mathcal{H} .

Hermitian and positive definite kernel. Let X be an arbitrary set and K be a kernel on X , that is, $K : X \times X \rightarrow \mathbb{C}$. The kernel K is called *Hermitian* if for any finite set of points $\{y_1, \dots, y_n\} \subseteq X$, we have

$$\sum_{i,j=1}^n \bar{\epsilon}_j \epsilon_i K(y_j, y_i) \in \mathbb{R}.$$

It is called *positive definite*, if for any complex numbers $\epsilon_1, \dots, \epsilon_n$, we have

$$\sum_{i,j=1}^n \bar{\epsilon}_j \epsilon_i K(y_j, y_i) \geq 0.$$

Note. From the previous inequality, it follows that for any finitely supported family of complex numbers $\{\epsilon_x\}_{x \in X}$, we have

$$\sum_{x,y \in X} \bar{\epsilon}_y \epsilon_x K(y, x) \geq 0. \quad (38)$$

Theorem 20. The reproducing kernel K of a reproducing kernel Hilbert space \mathcal{H} is a positive definite matrix in the sense of E.H. Moore.

Properties of RKHS. Given a reproducing kernel Hilbert space \mathcal{H} and its kernel $K(y, x)$ on X , then for all $x, y \in X$, we have

- i. $K(y, y) \geq 0$.
- ii. $K(y, x) = \overline{K(x, y)}$.
- iii. $|K(y, x)|^2 \leq K(y, y)K(x, x)$ (Schwarz inequality).

iv. Let $x_0 \in X$. Then the following statements are equivalent:

- a. $K(x_0, x_0) = 0$.
- b. $K(y, x_0) = 0$ for all $y \in X$.
- c. $f(x_0) = 0$ for all $f \in \mathcal{H}$.

Theorem 21. For any positive definite kernel K on X , there exists a unique Hilbert space \mathcal{H}_K of functions on X with reproducing kernel K .

Theorem 22. Every sequence of functions $(f_n)_{n \geq 1}$ that converges strongly to a function f in $\mathcal{H}_K(X)$ converges also in the pointwise sense, i.e., for any point $x \in X$,

$$\lim_{n \rightarrow \infty} f_n(x) = f(x).$$

In addition, this convergence is uniform on every subset of X on which $x \mapsto K(x, x)$ is bounded.

Theorem 23. A complex valued function g on X belongs to the reproducing kernel Hilbert space $\mathcal{H}_K(X)$ if and only if there exists $0 \leq \lambda < \infty$ such that,

$$\left[g(y)\overline{g(x)} \right] \leq \lambda^2 [K(y, x)] \text{ on } X. \quad (39)$$

$\|g\|$ coincides with the minimum of all such λ .

Theorem 24. If $K^{(1)}(y, x)$ and $K^{(2)}(y, x)$ are two positive definite kernels on X , then the following statements are mutually equivalent:

- i. $\mathcal{H}_{K^{(1)}}(X) \subseteq \mathcal{H}_{K^{(2)}}(X)$.
- ii. There exists $0 \leq \lambda < \infty$ such that

$$\left[K^{(1)}(y, x) \right] \leq \lambda^2 \left[K^{(2)}(y, x) \right].$$

Note. For any map φ from a set X to a Hilbert space \mathcal{H} , with the notation $x \mapsto \varphi_x$, a kernel K can be defined by

$$K(y, x) = \langle \varphi_x, \varphi_y \rangle \text{ for } x, y \in X. \quad (40)$$

Theorem 25. Let $\varphi : X \rightarrow \mathcal{H}$ be an arbitrary map and for $x, y \in X$ let K be defined as

$$K(y, x) = \langle \varphi_x, \varphi_y \rangle.$$

Then K is a positive definite kernel.

Theorem 26. Let T be the linear operator from \mathcal{H} to the space of functions on X , defined by

$$(Tf)(x) = \langle f, \varphi_x \rangle \text{ for } x \in X, f \in \mathcal{H}.$$

Then $\text{Ran}(T)$ coincides with $\mathcal{H}_K(X)$ and

$$\|Tf\|_K = \|P_{\mathcal{M}}f\| \text{ for } f \in \mathcal{H},$$

where \mathcal{M} is the orthogonal complement of $\text{Ker}(T)$, $P_{\mathcal{M}}$ is the orthogonal projection onto \mathcal{M} , and $\|\cdot\|_K$ denotes the norm in $\mathcal{H}_K(X)$.

Kolmogorov decomposition. Let $K(y, x)$ be a positive definite kernel on an abstract set X . Then there exists a Hilbert space \mathcal{H} and a function $\varphi : X \rightarrow \mathcal{H}$ such that

$$K(y, x) = \langle \varphi_x, \varphi_y \rangle \text{ for } x, y \in X.$$

3.2 Operations with RKHSs

Theorem 27. Let $K^{(0)}$ be the restriction of the positive definite kernel K to a nonempty subset X_0 of X and let $\mathcal{H}_{K^{(0)}}(X_0)$ and $\mathcal{H}_K(X)$ be the RKHS corresponding to $K^{(0)}$ and K , respectively. Then

$$\mathcal{H}_{K^{(0)}}(X_0) = \{f|_{X_0} : f \in \mathcal{H}_K(X)\} \quad (41)$$

and

$$\|h\|_{K^{(0)}} = \min \left\{ \|f\|_K : f|_{X_0} = h \right\} \text{ for all } h \in \mathcal{H}_{K^{(0)}}(X_0). \quad (42)$$

Remark. If $K^{(1)}(y, x)$ and $K^{(2)}(y, x)$ are two positive definite kernels, then

$$K(y, x) = K^{(1)}(y, x) + K^{(2)}(y, x)$$

is also a positive definite kernel.

Remark. Let $\mathcal{H}_{K^{(1)}}$, $\mathcal{H}_{K^{(2)}}$, and \mathcal{H}_K be RKHSs with reproducing kernels $K^{(1)}(y, x)$, $K^{(2)}(y, x)$, and $K(y, x)$, respectively, and let $K = K^{(1)} + K^{(2)}$. Then

$$\mathcal{H}_K(X) = \mathcal{H}_{K^{(1)}}(X) + \mathcal{H}_{K^{(2)}}(X),$$

and for $f \in \mathcal{H}_{K^{(1)}}(X)$ and $g \in \mathcal{H}_{K^{(2)}}(X)$, it follows that

$$\|f + g\|_K^2 = \min \left\{ \|f + h\|_{K^{(1)}}^2 + \|g - h\|_{K^{(2)}}^2 : h \in \mathcal{H}_{K^{(1)}}(X) \cap \mathcal{H}_{K^{(2)}}(X) \right\}. \quad (43)$$

Theorem 28. The intersection $\mathcal{H}_{K^{(1)}}(X) \cap \mathcal{H}_{K^{(2)}}(X)$ of Hilbert spaces $\mathcal{H}_{K^{(1)}}(X)$ and $\mathcal{H}_{K^{(2)}}(X)$ is again a Hilbert space of functions on X with respect to the norm

$$\|f\|^2 := \|f\|_{K^{(1)}}^2 + \|f\|_{K^{(2)}}^2.$$

In addition the intersection Hilbert space is a RKHS.

Theorem 29. The reproducing kernel of the space

$$\mathcal{H}_K(X) = \mathcal{H}_{K^{(1)}}(X) \cap \mathcal{H}_{K^{(2)}}(X)$$

is determined, as a quadratic form, by

$$\begin{aligned} \sum_{x,y} \bar{\varepsilon}_y \varepsilon_x K(y, x) &= \inf \left\{ \sum_{x,y} \bar{\eta}_y \eta_x K^{(1)}(y, x) + \sum_{x,y} \bar{\zeta}_y \zeta_x K^{(2)}(y, x) : [\varepsilon_x] \right. \\ &= [\eta_x] + [\zeta_x] \left. \right\}, \end{aligned}$$

where $[\varepsilon_x]$, $[\eta_x]$, $[\zeta_x]$ are an arbitrary complex valued function on X with finite support.

Theorem 30. The tensor product Hilbert space

$$\mathcal{H}_{K^{(1)}}(X) \otimes \mathcal{H}_{K^{(2)}}(X)$$

is a RKHS on $X \times X$.

Theorem 31. The RKHS $\mathcal{H}_K(X)$ of the kernel $K(y, x) = K^{(1)}(y, x) \cdot K^{(2)}(y, x)$ consists of all functions f on X for which there are sequences $(g_n)_{n \geq 0}$ of functions in $\mathcal{H}_{K^{(1)}}(X)$ and $(h_n)_{n \geq 0}$ of functions in $\mathcal{H}_{K^{(2)}}(X)$ so that

$$\sum_1^\infty \|g_n\|_{K^{(1)}}^2 \|h_n\|_{K^{(2)}}^2 < \infty, \quad \sum_1^\infty g_n(x)h_n(x) = f(x), \quad x \in X, \quad (44)$$

and the norm is given by

$$\|f\|_K^2 = \min \left\{ \sum_1^\infty \|g_n\|_{K^{(1)}}^2 \|h_n\|_{K^{(2)}}^2 \right\},$$

where the minimum is taken over the set of all sequences $(g_n)_{n \geq 0}$ and $(h_n)_{n \geq 0}$ satisfying (44).

3.3 Examples of RKHS. Bergman and Hardy spaces

Bergman space. The space of all analytic functions f on Ω for which

$$\iint_{\Omega} |f(z)|^2 dx dy < \infty, \quad (z = x + iy)$$

is satisfied is called the *Bergman space* on Ω and denoted by $A^2(\Omega)$.

Remark. $A^2(\Omega)$ is a RKHS with respect to the inner product

$$\langle f, g \rangle \equiv \langle f, g \rangle_{\Omega} := \iint_{\Omega} f(z) \overline{g(z)} dx dy,$$

and its kernel is called the *Bergman kernel* on Ω and denoted by $B^{(\Omega)}(w, z)$.

Bergman kernel for the unit disc. The Bergman kernel for the open unit disc \mathbb{D} is given by

$$B^{(\mathbb{D})}(w, z) = \frac{1}{\pi} \frac{1}{(1 - w\bar{z})^2} \quad \text{for } w, z \in \mathbb{D}. \quad (45)$$

Bergman kernel of a simply connected domain. The Bergman kernel of a simply connected domain $\Omega (\neq \mathbb{C})$ is given by

$$B^{(\Omega)}(w, z) = \frac{1}{\pi} \frac{\varphi'(w) \overline{\varphi'(z)}}{\left(1 - \varphi(w) \overline{\varphi(z)}\right)^2} \quad \text{for } w, z \in \Omega, \quad (46)$$

where φ is any conformal mapping function from Ω onto \mathbb{D} .

Theorem 32. A conformal mapping from Ω to \mathbb{D} can be recovered from the Bergman kernel of Ω .

Jordan curve. A *Jordan curve* is a continuous 1 – 1 image of $\{|\xi| = 1\}$ in \mathbb{C} .

Green function. A *Green function* $G(w, z)$ of Ω is a function harmonic in Ω except at z , where it has logarithmic singularity, and continuous in the closure $\bar{\Omega}$, with boundary values $G(w, z) = 0$ for all $w \in \partial\Omega$, where Ω is a finitely connected domain of the complex plane.

Theorem 33. Let Ω be a finitely connected domain bounded by analytic Jordan curves, and let $G(w, z)$ be the Green's function of Ω . Then the Bergman kernel function is

$$B^{(\Omega)}(w, z) = -\frac{2}{\pi} \frac{\partial^2 G}{\partial w \partial \bar{z}}(w, z), \quad w \neq z. \tag{47}$$

Hardy space. The closed linear span of $\{\varphi_n : n = 0, 1, \dots\}$ in $L^2(\mathbb{T})$ is called the (Hilbert type) *Hardy space* on \mathbb{T} and is denoted by $H^2(\mathbb{T})$. Here $\varphi_n(\xi) = \xi^n$.

Remark. $f \in L^2(\mathbb{T})$ belongs to the Hardy space $H^2(\mathbb{T})$ if and only if it is orthogonal to all φ_n ($n < 0$), that is, all Fourier coefficients of f with negative indices vanish. Then we have

$$\langle f, g \rangle_{L^2} = \sum_{n=0}^{\infty} a_n \bar{b}_n \text{ for } f, g \in H^2(\mathbb{T}), \tag{48}$$

where

$$a_n = \langle f, \varphi_n \rangle_{L^2} \text{ and } b_n = \langle g, \varphi_n \rangle_{L^2} \quad (n = 0, 1, \dots).$$

Szegő kernel. The kernel $S(\xi, z) := \frac{1}{1-\xi\bar{z}}$ for $\xi \in \mathbb{T}, z \in \mathbb{D}$, or its analytic extension $\tilde{S}(w, z) := \frac{1}{1-w\bar{z}}$ for $w, z \in \mathbb{D}$ is called the *Szegő kernel*.

Notes

This chapter intends to offer a sample survey for the fundamental concepts of Hilbert spaces and provide an introductory theory of reproducing kernels. We present the basic properties with important theorems and sometimes with punctual notes and remarks to support the subject. However, due to the limit of content and pages, we skipped the proofs of the theorems. The proofs of the first part can be found in [1, 2] and in most of the basic functional analysis books. Besides, the proofs of the second part (related with the reproducing kernels) can easily be found in [3]. The Hilbert space and functional analysis parts of this chapter are based on the books by J.B. Conway [1] and R.G. Douglas [2]. On the other hand, the reproducing kernel part is based on the lecture notes of T. Ando [4] and N. Aronszajn [5], the book of S. Saitoh and Y. Sawano [6], and the book of B. Okutmustur and A. Gheondea [3]. Moreover, the details of Bergman and Hardy spaces are widely explained in the books [7–9].



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Approximate Solutions of Some Boundary Value Problems by Using Operational Matrices of Bernstein Polynomials

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Abstract

In this chapter, we develop an efficient numerical scheme for the solution of boundary value problems of fractional order differential equations as well as their coupled systems by using Bernstein polynomials. On using the mentioned polynomial, we construct operational matrices for both fractional order derivatives and integrations. Also we construct a new matrix for the boundary condition. Based on the suggested method, we convert the considered problem to algebraic equation, which can be easily solved by using Matlab. In the last section, numerical examples are provided to illustrate our main results.

Keywords: Bernstein polynomials, coupled systems, fractional order differential equations, operational matrices of integration, approximate solutions

2010 MSC: 34L05, 65L05, 65T99, 34G10

1. Introduction

Generalization of classical calculus is known as fractional calculus, which is one of the fastest growing area of research, especially the theory of fractional order differential equations because this area has wide range of applications in real-life problems. Differential equations of fractional order provide an excellent tool for the description of many physical biological phenomena. The said equations play important roles for the description of hereditary characteristics of various materials and genetical problems in biological models as compared with integer order differential equations in the form of mathematical models. Nowadays, most of its applications are found in bio-medical engineering as well as in other scientific and engineering disciplines such as mechanics, chemistry, viscoelasticity, control theory, signal and image processing phenomenon, economics, optimization theory, etc.; for details, we refer the reader to study [1–9] and the references there in. Due to these important applications of fractional order differential equations, mathematicians are taking interest in the study of these equations because their models are more realistic and practical. In the last decade, many researchers have studied the existence and uniqueness of solutions to boundary value problems and their

coupled systems for fractional order differential equations (see [10–17]). Hence the area devoted to existence theory has been very well explored. However, every fractional differential equation cannot be solved for its analytical solutions easily due to the complex nature of fractional derivative; so, in such a situation, approximate solutions to such a problem is most efficient and helpful. Recently, many methods such as finite difference method, Fourier series method, Adomian decomposition method (ADM), inverse Laplace technique (ILT), variational iteration methods (VIM), fractional transform method (FTM), differential transform method (DTM), homotopy analysis method (HAM), method of radial base function (MRBM), wavelet techniques (WT), spectral methods and many more (for more details, see [9, 18–38]) have been developed for obtaining numerical solutions of such differential equations. These methods have their own merits and demerits. Some of them provide a very good approximation. However, in the last few years, some operational matrices were constructed to achieve good approximation as in [39]. After this, a variety of operational matrices were developed for different wavelet methods. This method uses operational matrices, where every operation, for example differentiation and integration, involved in these equations is performed with the help of a matrix. A large variety of operational matrices are available in the literature for different orthogonal polynomials like Legendre, Laguerre, Jacobi and Bernstein polynomials [40–48]. Motivated by the above applications and uses of fractional differential equations, in this chapter, we developed a numerical scheme based on operational matrices via Bernstein polynomials. Our proof is more generalized and there is no need to convert the Bernstein polynomial function vector to another basis like block pulse function or Legendre polynomials. To the best of our knowledge, the method we consider provides a very good approximation to the solution. By the use of these operational matrices, we apply our scheme to a single fractional order differential equation with given boundary conditions as

$$\begin{cases} D^\alpha y(t) + AD^\mu y(t) + By(t) = f(t), & 1 < \alpha \leq 2, \quad 0 < \mu \leq 1, \\ y(0) = a, \quad y(1) = b, \end{cases} \quad (1)$$

where $f(t)$ is the source term, A, B are any real numbers; then we extend our method to solve a boundary value problem of coupled system of fractional order differential equations of the form

$$\begin{cases} D^\alpha x(t) + A_1 D^{\mu_1} x(t) + B_1 D^{\nu_1} y(t) + C_1 x(t) + D_1 y(t) = f(t), & 1 < \alpha \leq 2, \quad 0 < \mu_1, \nu_1 \leq 1, \\ D^\beta y(t) + A_2 D^{\mu_2} x(t) + B_2 D^{\nu_2} y(t) + C_2 x(t) + D_2 y(t) = g(t), & 1 < \beta \leq 2, \quad 0 < \mu_2, \nu_2 \leq 1, \\ y(0) = a, \quad y(1) = b, \quad y(0) = c, \quad y(1) = d, \end{cases} \quad (2)$$

where $f(t), g(t)$ are source terms of the system, $A_i, B_i, C_i, D_i (i = 1, 2)$ are any real constants. Also we compare our approximations to exact values and approximations of other methods like Jacobi polynomial approximations and Haar wavelets methods to evaluate the efficiency of the proposed method. We also provide some examples for the illustration of our main results.

This chapter is designed in five sections. In the first section of the chapter, we have cited some basic works related to the numerical and analytical solutions of differential equations of arbitrary order by various methods. The necessary definitions and results related to fractional calculus and Bernstein polynomials along with the construction of some operational matrices are given in Section 2. In Section 3, we have discussed the main theory for the numerical procedure. Section 4 contains

some interesting practical examples and their images. Section 5 describes the conclusion of the chapter.

2. Basic definitions and results

In this section, we recall some fundamental definitions and results from the literature, which can be found in [10–16].

Definition 2.1. *The fractional integral of order $\gamma \in \mathbb{R}_+$ of a function $y \in L^1([0, 1], \mathbb{R})$ is defined as*

$$I_{0+}^\gamma y(t) = \frac{1}{\Gamma(\gamma)} \int_0^t (t - \tau)^{\gamma-1} y(\tau) d\tau.$$

Definition 2.2. *The Caputo fractional order derivative of a function y on the interval $[0, 1]$ is defined by*

$$D_{0+}^\gamma y(t) = \frac{1}{\Gamma(n - \gamma)} \int_0^t (t - \tau)^{n-\gamma-1} y^{(n)}(\tau) d\tau,$$

where $n = [\gamma] + 1$ and $[\gamma]$ represents the integer part of γ .

Lemma 2.1. *The fractional differential equation of order $\gamma > 0$*

$$D^\gamma y(t) = 0, n - 1 < \gamma \leq n,$$

has a unique solution of the form $y(t) = d_0 + d_1 t + d_2 t^2 + \dots + d_{n-1} t^{n-1}$, where $d_k \in \mathbb{R}$ and $k = 0, 1, 2, 3, \dots, n - 1$.

Lemma 2.2. *The following result holds for fractional differential equations*

$$I^\gamma D^\gamma y(t) = y(t) + d_0 + c_1 t + d_2 t^2 + \dots + d_{n-1} t^{n-1},$$

for arbitrary $d_k \in \mathbb{R}, k = 0, 1, 2, \dots, n - 1$.

Hence it follows that

$$D^\gamma t^k = \frac{\Gamma(k + 1)}{\Gamma(k - \gamma + 1)} t^{k-\gamma}, \quad I^\gamma t^k = \frac{\Gamma(k + 1)}{\Gamma(k + \gamma + 1)} t^{k+\gamma} \quad \text{and} \quad D^\gamma [\text{constant}] = 0.$$

2.1 The Bernstein polynomials

The Bernstein polynomials $B_{i,m}(t)$ on $[0, 1]$ can be defined as

$$B_{i,m}(t) = \binom{m}{i} t^i (1 - t)^{m-i}, \text{ for } i = 0, 1, 2 \dots m,$$

where $\binom{m}{i} = \frac{m!}{(m-i)!i!}$, which on further simplification can be written in the most simplified form as

$$B_{i,m}(t) = \sum_{k=0}^{m-i} \Theta_{(i,k,m)} t^{k+i}, \quad i = 0, 1, 2 \dots m, \tag{3}$$

where

$$\Theta_{(i,k,m)} = (-1)^k \binom{m}{i} \binom{m-i}{k}.$$

Note that the sum of the Bernstein polynomials converges to 1.

Lemma 2.3. Convergence Analysis: Assume that the function $g \in C^{m+1}[0, 1]$ that is $m + 1$ times continuously differentiable function and let $X = \langle B_{0,m}, B_{1,m}, \dots, B_{m,m} \rangle$. If $C^T \Psi(x)$ is the best approximation of g out of X , then the error bound is presented as

$$\|g - C^T \Psi\|_2 \leq \frac{\sqrt{2MS^{\frac{2m+3}{2}}}}{\Gamma(m+2)\sqrt{2m+3}},$$

where $M = \max_{x \in [0,1]} |g^{(m+1)}(x)|$, $S = \max\{1 - x_0, x_0\}$.

Proof. In view of Taylor polynomials, we have

$$F(x) = g(x_0) + (x - x_0)g^{(1)}(x_0) + \frac{(x - x_0)^2}{\Gamma 3}g^{(2)} + \dots + \frac{(x - x_0)^m}{\Gamma(m+1)}g^{(m)},$$

from which we know that

$$|g - F(x)| = |g^{(m+1)}(\eta)| \frac{(x - x_0)^{m+1}}{\Gamma(m+2)}, \text{ there exist } \eta \in (0, 1).$$

Due the best approximation $C^T \Psi(x)$ of g , we have

$$\begin{aligned} \|g - C^T \Psi(x)\|_2^2 &\leq \|g - F\|_2^2 \\ &= \int_0^1 (g(x) - F(x))^2 dx \\ &= \int_0^1 \left[|g^{(m+1)}(\eta)| \frac{(x - x_0)^{m+1}}{\Gamma(m+2)} \right]^2 dx \\ &\leq \frac{M^2}{\Gamma^2(m+2)} \int_0^1 (x - x_0)^{2m+2} dx \\ &\leq \frac{2M^2 S^{2m+3}}{\Gamma^2(m+2)(2m+3)}. \end{aligned}$$

Hence we have

$$\|g - C^T \Psi(x)\|_2 \leq \frac{\sqrt{2MS^{\frac{2m+3}{2}}}}{\Gamma(m+2)\sqrt{(2m+3)}}. \quad \square$$

Let $H = L^2[0, 1]$ be a Hilbert space, then the inner product can be defined as

$$\langle f, g \rangle = \int_0^1 f(x)g(x)dx$$

and

$$Y = span\{B_{0,m}, B_{1,m}, \dots, B_{m,m}\}$$

is a finite dimensional and closed subspace. So if $f \in H$ is an arbitrary element then its best approximation is unique in Y . This terminology can be achieved by using $y_0 \in Y$ and for all $y \in Y$, we have $\|f - y_0\| \leq \|f - y\|$. Thus any function can be approximated in terms of Bernstein polynomials as

$$f(t) = \sum_{i=0}^m c_i B_{(i,m)}, \tag{4}$$

where coefficient c_i can easily be calculated by multiplying (4) by $B_{(j,m)}(t)$, $j = 0, 1, 2, \dots, m$ and integrating over $[0, 1]$ by using inner product and $d_i = \int_0^1 B_{(i,m)}(t)f(t)dt$, $\theta_{(i,j)} = \int_0^1 B_{(i,m)}(t)B_{(j,m)}(t)dt$, $i, j = 0, 1, 2, \dots, m$, we have

$$\int_0^1 f(t)B_{(j,m)}(t)dt = \int_0^1 \sum_{i=0}^m c_i B_{(i,m)}(t) \cdot B_{(j,m)}(t)dt, \quad j = 0, 1, 2, \dots, m,$$

which implies that $\int_0^1 f(t)B_{(j,m)}(t)dt = \sum_{i=0}^m c_i \int_0^1 B_{(i,m)}(t) \cdot B_{(j,m)}(t)dt$, $j = 0, 1, 2, \dots, m$

which implies that $[d_0 \ d_1 \ \dots \ d_m] = [c_0 \ c_1 \ \dots \ c_m]$ $\begin{bmatrix} \theta_{(0,0)} & \theta_{(0,1)} & \dots & \theta_{(0,r)} & \dots & \theta_{(0,m)} \\ \theta_{(1,0)} & \theta_{(1,1)} & \dots & \theta_{(1,r)} & \dots & \theta_{(1,m)} \\ \vdots & \vdots & \vdots & \vdots & \vdots & \vdots \\ \theta_{(r,0)} & \theta_{(r,1)} & \dots & \theta_{(r,r)} & \dots & \theta_{(r,m)} \\ \vdots & \vdots & \vdots & \vdots & \vdots & \vdots \\ \theta_{(m,0)} & \theta_{(m,1)} & \dots & \theta_{(m,r)} & \dots & \theta_{(m,m)} \end{bmatrix}$.

(5)

Let $X_M = [d_0 \ d_1 \ \dots \ d_m]$, $C_M = [c_0 \ c_1 \ \dots \ c_m]$, where $M = m + 1$ where M is the scale level and $\Phi_{M \times M} = \begin{bmatrix} \theta_{(0,0)} & \theta_{(0,1)} & \dots & \theta_{(0,r)} & \dots & \theta_{(0,m)} \\ \theta_{(1,0)} & \theta_{(1,1)} & \dots & \theta_{(1,r)} & \dots & \theta_{(1,m)} \\ \vdots & \vdots & \vdots & \vdots & \vdots & \vdots \\ \theta_{(r,0)} & \theta_{(r,1)} & \dots & \theta_{(r,r)} & \dots & \theta_{(r,m)} \\ \vdots & \vdots & \vdots & \vdots & \vdots & \vdots \\ \theta_{(m,0)} & \theta_{(m,1)} & \dots & \theta_{(m,r)} & \dots & \theta_{(m,m)} \end{bmatrix}$, so

$$X_M = C_M \Phi_{M \times M} \Rightarrow C_M = X_M \Phi_{M \times M}^{-1}. \tag{6}$$

where $\Phi_{m \times m}$ is called the dual matrix of the Bernstein polynomials. After calculating c_i , (4) can be written as

$$f(t) = C_M B_M^T(t), \quad C_M \text{ is coefficient matrix}$$

where

$$B_M(t) = [B_{(0,m)}, B_{(1,m)}, \dots, B_{(m,m)}]. \tag{7}$$

Lemma 2.4. Let $B_M^T(t)$ be the function vector defined in (3), then the fractional order integration of $B_M^T(t)$ is given by

$$I^\alpha B_M^T(t) = P_{M \times M}^\alpha B_M^T(t), \tag{8}$$

where $P_{M \times M}^\alpha$ is the fractional integration's operational matrix defined as

$$P_{M \times M}^\alpha = \hat{P}_{M \times M}^\alpha \Phi_{M \times M}^{-1}$$

and $\Phi_{M \times M}^{-1}$ is given in (3) and $P_{M \times M}^\alpha$ is given by

$$\hat{P}_{M \times M}^\alpha = \begin{bmatrix} \Psi_{(0,0)} & \Psi_{(0,1)} & \cdots & \Psi_{(0,r)} & \cdots & \Psi_{(0,m)} \\ \Psi_{(1,0)} & \Psi_{(1,1)} & \cdots & \Psi_{(1,r)} & \cdots & \Psi_{(1,m)} \\ \vdots & \vdots & \vdots & \vdots & \vdots & \vdots \\ \Psi_{(r,0)} & \Psi_{(r,1)} & \cdots & \Psi_{(r,r)} & \cdots & \Psi_{(r,m)} \\ \vdots & \vdots & \vdots & \vdots & \vdots & \vdots \\ \Psi_{(m,0)} & \Psi_{(m,1)} & \cdots & \Psi_{(m,r)} & \cdots & \Psi_{(m,m)} \end{bmatrix}, \tag{9}$$

where

$$\Psi_{ij} = \sum_{k=0}^{m-i} \sum_{l=0}^{m-j} \theta_{(i,k,m)} \theta_{(j,l,m)} \frac{\Gamma(k+i+1)}{(i+j+k+l+\alpha+1)\Gamma(k+i+\alpha+1)}. \tag{10}$$

Proof. Consider

$$B_{i,m}(t) = \sum_{k=0}^{m-i} \theta_{(i,k,m)} t^{k+i} \tag{11}$$

taking fractional integration of both sides, we have

$$I^\alpha B_{i,m}(t) = \sum_{k=0}^{m-i} \theta_{(i,k,m)} I^\alpha t^{k+i} = \sum_{k=0}^{m-i} \theta_{(i,k,m)} \frac{\Gamma(k+i+\alpha)}{\Gamma(k+i+\alpha+1)} t^{k+i+\alpha}. \tag{12}$$

Now to approximate right-hand sides of above

$$\sum_{k=0}^{m-i} \theta_{(i,k,m)} \frac{\Gamma(k+i+\alpha)}{\Gamma(k+i+\alpha+1)} t^{k+i+\alpha} = C_M^{(i)} B_M^T(t) \tag{13}$$

where $C_M^{(i)}$ can be approximated by using (3) as

$$C_M^{(i)} = X_M^{(i)} \Phi_{M \times M}^{-1}, \tag{14}$$

where entries of the vector $X_M^{(i)}$ can be calculated in generalized form as

$$\begin{aligned} X_M^{(j)} &= \int_0^1 \sum_{k=0}^{m-i} \theta_{(i,k,m)} \frac{\Gamma(k+i+\alpha)}{\Gamma(k+i+\alpha+1)} t^{k+i+\alpha} B_{j,m}(t) dt, \quad j = 0, 1, 2, \dots, m \\ \Rightarrow X_M^{(j)} &= \int_0^1 \sum_{k=0}^{m-i} \theta_{(i,k,m)} \frac{\Gamma(k+i+\alpha)}{\Gamma(k+i+\alpha+1)} \sum_{l=0}^{m-j} \theta_{(j,l,m)} t^{k+i+\alpha} \cdot t^{l+j} dt, \quad j = 0, 1, 2, \dots, m \\ &= \sum_{k=0}^{m-i} \theta_{(i,k,m)} \sum_{l=0}^{m-j} \theta_{(j,l,m)} \frac{\Gamma(k+i+\alpha)}{\Gamma(k+i+\alpha+1)} \frac{1}{(k+l+j+i+\alpha+1)}, \quad j = 0, 1, 2, \dots, m \end{aligned} \tag{15}$$

evaluating this result for $i = 0, 1, 2, \dots, m$, we have

$$\begin{bmatrix} I^\alpha B_{0,m}(t) \\ I^\alpha B_{1,m}(t) \\ \vdots \\ \vdots \\ \vdots \\ I^\alpha B_{m,m}(t) \end{bmatrix} = \begin{bmatrix} X_M^{(0)} \Phi_{M \times M}^{-1} B_M^T(t) \\ X_M^{(1)} \Phi_{M \times M}^{-1} B_M^T(t) \\ \vdots \\ \vdots \\ \vdots \\ X_M^{(m)} \Phi_{M \times M}^{-1} B_M^T(t) \end{bmatrix} \quad (16)$$

further writing

$$\Psi_{i,j} = \sum_{k=0}^{m-i} \sum_{l=0}^{m-j} \theta_{(i,k,m)} \cdot \theta_{(j,l,m)} \frac{\Gamma(k+i+\alpha)}{\Gamma(k+i+\alpha+1)} \frac{1}{(k+l+j+i+\alpha+1)}$$

we get

$$I^\alpha B_M^T(t) = \hat{P}_{M \times M}^\alpha \cdot \Phi_{M \times M}^{-1} \cdot B_M^T(t). \quad (17)$$

Let us represent

$$\hat{P}_{M \times M}^\alpha \cdot \Phi_{M \times M}^{-1} = P_{M \times M}^\alpha$$

thus

$$I^\alpha B_M^T(t) = P_{M \times M}^\alpha \cdot B_M^T(t). \quad (18)$$

□

Lemma 2.5. Let $B_M^T(t)$ be the function vector as defined in (3), then fractional order derivative is defined as

$$D^\alpha B_M^T(t) = G_{M \times M}^\alpha \cdot B_M^T(t) \quad (19)$$

where $G_{M \times M}^\alpha$ is the operational matrix of fractional order derivative given by

$$G_{M \times M}^\alpha = \hat{G}_{M \times M}^\alpha \Phi_{M \times M}^{-1}, \quad (20)$$

where $\Phi_{M \times M}$ is the dual matrix given in (3) and

$$\hat{G}_{M \times M}^\alpha = \begin{bmatrix} \Psi_{(0,0)} & \Psi_{(0,1)} & \cdots & \Psi_{(0,r)} & \cdots & \Psi_{(0,m)} \\ \Psi_{(1,0)} & \Psi_{(1,1)} & \cdots & \Psi_{(1,r)} & \cdots & \Psi_{(1,m)} \\ \vdots & \vdots & \vdots & \vdots & \vdots & \vdots \\ \Psi_{(r,0)} & \Psi_{(r,1)} & \cdots & \Psi_{(r,r)} & \cdots & \Psi_{(r,m)} \\ \vdots & \vdots & \vdots & \vdots & \vdots & \vdots \\ \Psi_{(m,0)} & \Psi_{(m,1)} & \cdots & \Psi_{(m,r)} & \cdots & \Psi_{(m,m)} \end{bmatrix}, \quad (21)$$

where $\Psi_{(i,j)}$ is defined for two different cases as

Case I: ($i < [\alpha]$)

$$\Psi_{i,j} = \sum_{k=[\alpha]}^{m-i} \sum_{l=0}^{m-j} \theta_{(i,k,m)} \cdot \theta_{(j,l,m)} \frac{\Gamma(k+i-\alpha)}{\Gamma(k+i-\alpha+1)} \frac{1}{(k+l+j+i-\alpha+1)} \quad (22)$$

Case II: ($i \geq [\alpha]$)

$$\Psi_{i,j} = \sum_{k=0}^{m-i} \sum_{l=0}^{m-j} \theta_{(i,k,m)} \cdot \theta_{(j,l,m)} \frac{\Gamma(k+i-\alpha)}{\Gamma(k+i-\alpha+1)} \frac{1}{(k+l+j+i-\alpha+1)}. \quad (23)$$

Proof. Consider the general element as

$$D^\alpha B_{i,m}(t) = D^\alpha \left(\sum_{k=0}^{m-i} \theta_{(i,k,m)} \cdot t^{k+i} \right) = \sum_{k=0}^{m-i} \theta_{(i,k,m)} D^\alpha t^{k+i}. \quad (24)$$

It is to be noted in the polynomial function $B_{i,m}$ the power of the variable ‘ t ’ is an ascending order and the lowest power is ‘ i ’ therefore the first $[\alpha - 1]$ terms becomes zero when we take derivative of order α .

Case I: ($i < [\alpha]$) By the use of definition of fractional derivative

$$D^\alpha B_{i,m}(t) = \sum_{k=[\alpha]}^{m-i} \theta_{(i,k,m)} \frac{\Gamma(k+i+1)}{\Gamma(k+i-\alpha+1)} t^{k+i-\alpha}. \quad (25)$$

Now approximating RHS of (25) as

$$\sum_{k=[\alpha]}^{m-i} \theta_{(i,k,m)} \frac{\Gamma(k+i+1)}{\Gamma(k+i-\alpha+1)} t^{k+i-\alpha} = C_M^{(i)} B_M^T(t) \quad (26)$$

further implies that

$$\begin{aligned} X_M^{(j)} &= \int_0^1 \sum_{k=[\alpha]}^{m-i} \theta_{(i,k,m)} \frac{\Gamma(k+i+1)}{\Gamma(k+i-\alpha+1)} t^{k+i-\alpha} B_{j,m}(t) dt, \quad j = 0, 1, 2, \dots, m \\ \Rightarrow X_M^{(j)} &= \sum_{k=[\alpha]}^{m-i} \theta_{(i,k,m)} \sum_{l=0}^{m-j} \theta_{(j,l,m)} \frac{\Gamma(k+i+1)}{\Gamma(k+i-\alpha+1)(k+i+l+j-\alpha+1)}, \quad j = 0, 1, 2, \dots, m \end{aligned} \quad (27)$$

Case II: ($i \geq [\alpha]$) if $i \leq [\alpha]$ then

$$X_M^{(j)} = \sum_{k=0}^{m-i} \theta_{(i,k,m)} \sum_{l=0}^{m-j} \theta_{(j,l,m)} \frac{\Gamma(k+i+1)}{\Gamma(k+i-\alpha+1)(k+i+l+j-\alpha+1)}, \quad j = 0, 1, 2, \dots, m. \quad (28)$$

After careful simplification, we get

$$\begin{bmatrix} D^\alpha B_{0,m}(t) \\ D^\alpha B_{1,m}(t) \\ \vdots \\ \vdots \\ \vdots \\ D^\alpha B_{m,m}(t) \end{bmatrix} = \begin{bmatrix} X_M^{(0)} \Phi_{M \times M}^{-1} B_M^T(t) \\ X_M^{(1)} \Phi_{M \times M}^{-1} B_M^T(t) \\ \vdots \\ \vdots \\ \vdots \\ X_M^{(m)} \Phi_{M \times M}^{-1} B_M^T(t) \end{bmatrix}. \tag{29}$$

On further simplification, we have

$$\Psi_{i,j} = \sum_{k=[\alpha]}^{m-i} \sum_{l=0}^{m-j} \theta_{(i,k,m)} \cdot \theta_{(j,l,m)} \frac{\Gamma(k+i+1)}{\Gamma(k+i-\alpha+1)} \frac{1}{(k+l+j+i-\alpha+1)} \quad (i < [\alpha])$$

$$\Psi_{i,j} = \sum_{k=0}^{m-i} \sum_{l=0}^{m-j} \theta_{(i,k,m)} \cdot \theta_{(j,l,m)} \frac{\Gamma(k+i+1)}{\Gamma(k+i-\alpha+1)} \frac{1}{(k+l+j+i-\alpha+1)}$$

we get $D^\alpha B_M^T(t) = \hat{G}_{M \times M}^\alpha \Phi_{M \times M}^{-1} \cdot B_M^T(t).$ (30)

Let

$$\hat{G}_{M \times M}^\alpha \Phi_{M \times M}^{-1} = G_{M \times M}^\alpha$$

so

$$D^\alpha B_M^T(t) = G_{M \times M}^\alpha B_M^T(t)$$

which is the desired result. □

Lemma 2.6. *An operational matrix corresponding to the boundary condition by taking $B_M^T(t)$ is function vector and K is coefficient vector by taking the approximation*

$$u(t) = K\hat{B}(t)$$

is given by

$$Q_{M \times M}^{\alpha, \phi} = \begin{bmatrix} \Omega_{(0,0)} & \Omega_{(0,1)} & \cdots & \Omega_{(0,r)} & \cdots & \Omega_{(0,m)} \\ \Omega_{(1,0)} & \Omega_{(1,1)} & \cdots & \Omega_{(1,r)} & \cdots & \Omega_{(1,m)} \\ \vdots & \vdots & \vdots & \vdots & \vdots & \vdots \\ \Omega_{(r,0)} & \Omega_{(r,1)} & \cdots & \Omega_{(r,r)} & \cdots & \Omega_{(r,m)} \\ \vdots & \vdots & \vdots & \vdots & \vdots & \vdots \\ \Omega_{(m,0)} & \Omega_{(m,1)} & \cdots & \Omega_{(m,r)} & \cdots & \Omega_{(m,m)} \end{bmatrix}, \tag{31}$$

where

$$\Omega_{i,j} = \int_0^1 \Delta_{i,m} \phi(t) B_j(t) dt, \quad i, j = 0, 1, 2, \dots, m.$$

Proof. Let us take $u(t) = K\hat{B}(t)$, then

$${}_0I_1^\alpha K\hat{B}(t) = K_0I_1^\alpha \hat{B}(t) = K \begin{bmatrix} {}_0I_1^\alpha B_0(t) \\ {}_0I_1^\alpha B_1(t) \\ \vdots \\ \vdots \\ \vdots \\ {}_0I_1^\alpha B_m(t) \end{bmatrix}.$$

Let us evaluate the general terms

$$\begin{aligned} {}_0I_1^\alpha B_i(t)dt &= \frac{1}{\Gamma\alpha} \int_0^1 (1-s)^{\alpha-1} B_{i,m}(s) ds \\ &= \frac{1}{\Gamma\alpha} \sum_{k=0}^{m-i} \Theta_{i,k,m} \int_0^1 (1-s)^{\alpha-1} s^{k+i} ds. \end{aligned} \tag{32}$$

Since by

$$L\left(\int_0^1 (1-s)^{\alpha-1} s^{k+i} ds\right) = \frac{\Gamma\alpha\Gamma(k+i+1)}{\tau^{k+\alpha+i}}$$

taking inverse Laplace of both sides, we get

$$\int_0^1 (1-s)^{\alpha-1} s^{k+i} ds = L^{-1}\left[\frac{\Gamma\alpha\Gamma(k+i+1)}{\tau^{k+\alpha+i}}\right] = \frac{\Gamma\alpha\Gamma(k+i+1)}{\Gamma(k+i+\alpha+1)}$$

now Eq. (32) implies that

$${}_0I_1^\alpha B_i(t)dt = \sum_{k=0}^{m-i} \Theta_{i,k,m} \frac{\Gamma(k+i+1)}{\Gamma(k+i+\alpha+1)} = \Delta_{i,m}. \tag{33}$$

Now using the approximation $\Delta_{i,m}\phi(t) = \sum_{i=0}^m \hat{c}_i B_i(t) = C_M^i B_M^T$, and using Eq. (3) we get $C_M^i = K_M^i \Phi_{M \times M}^{-1} B_M^T$ and using $c_j = \int_0^1 \phi(t) B_j(t) dt$,

$$\begin{aligned} \phi(t)KI^\alpha \hat{B}(t) &= K \begin{bmatrix} \Delta_{0,m}\phi(t) \\ \Delta_{1,m}\phi(t) \\ \vdots \\ \vdots \\ \vdots \\ \Delta_{m,m}\phi(t) \end{bmatrix} = K \begin{bmatrix} C_M^0 \Phi_{M \times M}^{-1} B_M^T(t) \\ C_M^1 \Phi_{M \times M}^{-1} B_M^T(t) \\ \vdots \\ \vdots \\ \vdots \\ C_M^m \Phi_{M \times M}^{-1} B_M^T(t) \end{bmatrix} \\ &= K \begin{bmatrix} c_0^0 & c_1^0 & \dots & c_r^0 & \dots & c_m^0 \\ c_0^1 & c_1^1 & \dots & c_r^1 & \dots & c_m^1 \\ \vdots & \vdots & \vdots & \vdots & \vdots & \vdots \\ c_0^r & c_1^r & \dots & c_r^r & \dots & c_m^r \\ \vdots & \vdots & \vdots & \vdots & \vdots & \vdots \\ c_0^m & c_1^m & \dots & c_r^m & \dots & c_m^m \end{bmatrix} \begin{bmatrix} \Phi_{M \times M}^{-1} B_M^T(t) \\ \Phi_{M \times M}^{-1} B_M^T(t) \\ \vdots \\ \vdots \\ \vdots \\ \Phi_{M \times M}^{-1} B_M^T(t) \end{bmatrix}. \end{aligned} \tag{34}$$

On further simplification, we get

$$\phi(t)KI^\alpha\hat{B}(t) = K \begin{bmatrix} \Omega_{(0,0)} & \Omega_{(0,1)} & \cdots & \Omega_{(0,r)} & \cdots & \Omega_{(0,m)} \\ \Omega_{(1,0)} & \Omega_{(1,1)} & \cdots & \Omega_{(1,r)} & \cdots & \Omega_{(1,m)} \\ \vdots & \vdots & \vdots & \vdots & \vdots & \vdots \\ \Omega_{(r,0)} & \Omega_{(r,1)} & \cdots & \Omega_{(r,r)} & \cdots & \Omega_{(r,m)} \\ \vdots & \vdots & \vdots & \vdots & \vdots & \vdots \\ \Omega_{(m,0)} & \Omega_{(m,1)} & \cdots & \Omega_{(m,r)} & \cdots & \Omega_{(m,m)} \end{bmatrix} \begin{bmatrix} B_0(t) \\ B_1(t) \\ \vdots \\ \vdots \\ \vdots \\ B_m(t) \end{bmatrix}. \quad (35)$$

So

$$\phi(t)_0I_1^\alpha u(t) = KQ_{M \times M}^{\alpha, \phi} B_M^T(t),$$

and

$$\Omega_{i,j} = \int_0^1 \Delta_{i,m} \phi(t) B_j(t) dt, \quad i, j = 0, 1, 2, \dots, m. \quad (36)$$

which is the required result. □

3. Applications of operational matrices

In this section, we are going to approximate a boundary value problem of fractional order differential equation as well as a coupled system of fractional order boundary value problem. The application of obtained operational matrices is shown in the following procedure.

3.1 Fractional differential equations

Consider the following problem in generalized form of fractional order differential equation

$$D^\alpha y(t) + AD^\mu y(t) + By(t) = f(t), \quad 1 < \alpha \leq 2, \quad 0 < \mu \leq 1, \quad (37)$$

subject to the boundary conditions $y(0) = a, \quad y(1) = b,$

where $f(t)$ is a source term; A, B are any real constants and $y(t)$ is an unknown solution which we want to determine. To obtain a numerical solution of the above problem in terms of Bernstein polynomials, we proceed as

$$\text{Let } D^\alpha y(t) = K_M P_{M \times M}^\alpha B_M^T(t). \quad (38)$$

Applying fractional integral of order α we have

$$y(t) = K_M P_{M \times M}^\alpha B_M^T(t) - c_0 + c_1 t$$

using boundary conditions, we have

$$c_0 = a, \quad c_1 = b - a - K_M P_{M \times M}^\alpha B_M^T(t) \Big|_{t=1}.$$

Using the approximation and Lemma 2.2

$$a + t(b - a) = F_M^{(1)} B_M^T(t), \quad tP_{M \times M}^\alpha B_M^T(t) \Big|_{t=1} = Q_{M \times M}^{\alpha, \phi} B_M^T(t).$$

Hence

$$\begin{aligned} y(t) &= K_M P_{M \times M}^\alpha B_M^T(t) + a + t(b - a) - tK_M P_{M \times M}^\alpha B_M^T(t) \Big|_{t=1}, \\ \text{which gives } y(t) &= K_M P_{M \times M}^\alpha B_M^T(t) + F_M^{(1)} B_M^T(t) - Q_{M \times M}^{\alpha, \phi} B_M^T(t) \\ &= K_M \left(P_{M \times M}^\alpha - Q_{M \times M}^{\alpha, \phi} \right) B_M^T(t) + F_M^{(1)} B_M^T(t). \end{aligned} \quad (39)$$

Now

$$\begin{aligned} D^\mu y(t) &= D^\mu \left[K_M \left(P_{M \times M}^\alpha - Q_{M \times M}^{\alpha, \phi} \right) B_M^T(t) + F_M^{(1)} B_M^T(t) \right] \\ &= K_M \left(P_{M \times M}^\alpha - Q_{M \times M}^{\alpha, \phi} \right) G_{M \times M}^\mu B_M^T(t) + F_M^{(1)} G_{M \times M}^\mu B_M^T(t) \end{aligned} \quad (40)$$

and

$$f(t) = F_M^{(2)} B_M^T(t). \quad (41)$$

Putting Eqs. (38)–(41) in Eq. (37), we get

$$\begin{aligned} K_M B_M^T(t) + AK_M \left(P_{M \times M}^\alpha - Q_{M \times M}^{\alpha, \phi} \right) G_{M \times M}^\mu B_M^T(t) + AF_M^{(1)} G_{M \times M}^\mu B_M^T(t) \\ + BK_M \left(P_{M \times M}^\alpha - Q_{M \times M}^{\alpha, \phi} \right) B_M^T(t) + BF_M^{(1)} B_M^T(t) = F_M^{(2)} B_M^T(t). \end{aligned} \quad (42)$$

In simple form, we can write (42) as

$$\begin{aligned} K_M B_M^T(t) + AK_M \left(P_{M \times M}^\alpha - Q_{M \times M}^{\alpha, \phi} \right) G_{M \times M}^\mu B_M^T(t) + AF_M^{(1)} G_{M \times M}^\mu B_M^T(t) \\ + BK_M \left(P_{M \times M}^\alpha - Q_{M \times M}^{\alpha, \phi} \right) B_M^T(t) + BF_M^{(1)} B_M^T(t) - F_M^{(2)} B_M^T(t) = 0 \end{aligned} \quad (43)$$

$$K_M + K_M \left(A\hat{P}_{M \times M}^\alpha G_{M \times M}^\mu + B\hat{P}_{M \times M}^\alpha \right) + AF_M^{(1)} G_{M \times M}^\mu + BF_M^{(1)} - F_M^{(2)},$$

where

$$\hat{P}_{M \times M}^\alpha = P_{M \times M}^\alpha - Q_{M \times M}^{\alpha, \phi}.$$

Eq. (43) is an algebraic equation of Lyapunov type, which can be easily solved for the unknown coefficient vector K_M . When we find K_M , then putting this in Eq. (39), we get the required approximate solution of the problem.

3.2 Coupled system of boundary value problem of fractional order differential equations

Consider a coupled system of a fractional order boundary value problem

$$\begin{aligned} D^\alpha x(t) + A_1 D^{\mu_1} x(t) + B_1 D^{\nu_1} y(t) + C_1 x(t) + D_1 y(t) &= f(t), \quad 1 < \alpha \leq 2, 0 < \mu_1, \nu_1 \leq 1, \\ D^\beta y(t) + A_2 D^{\mu_2} x(t) + B_2 D^{\nu_2} y(t) + C_2 x(t) + D_2 y(t) &= g(t), \quad 1 < \beta \leq 2, 0 < \mu_2, \nu_2 \leq 1, \end{aligned} \quad (44)$$

subject to the boundary conditions

$$x(0) = a, \quad x(1) = b \quad y(0) = c, \quad y(1) = d, \quad (45)$$

where $A_i, B_i, C_i, D_i (i = 1, 2)$ are any real constants, $f(t), g(t)$ are given source terms. We approximate the solution of the above system in terms of Bernstein polynomials such as

$$D^\alpha x(t) = K_M B_M^T(t), \quad D^\beta y(t) = L_M B_M^T(t)$$

$$x(t) = K_M P_{M \times M}^\alpha B_M^T(t) + c_0 + c_1 t, \quad y(t) = L_M \left(P_{M \times M}^\beta B_M^T(t) + d_0 + d_1 t \right)$$

applying boundary conditions, we have

$$x(t) = K_M \left(P_{M \times M}^\alpha B_M^T(t) + a + t(b - a) - t K_M P_{M \times M}^\alpha B_M^T(t) \right) \Big|_{t=1},$$

$$y(t) = K_M \left(P_{M \times M}^\beta B_M^T(t) + c + t(d - c) - t K_M P_{M \times M}^\beta B_M^T(t) \right) \Big|_{t=1}.$$

let us approximate

$$a + t(b - a) = F_M^1 B_M^T(t), \quad c + t(d - c) = F_M^2 B_M^T(t)$$

$$t P_{M \times M}^\alpha B_M^T(t) \Big|_{t=1} = Q_{M \times M}^{\alpha, \phi} B_M^T(t), \quad t P_{M \times M}^\beta B_M^T(t) \Big|_{t=1} = Q_{M \times M}^{\beta, \phi} B_M^T(t)$$

then

$$x(t) = K_M P_{M \times M}^\alpha B_M^T(t) + F_M^{(1)} B_M^T(t) - K_M Q_{M \times M}^{\alpha, \phi} B_M^T(t)$$

$$y(t) = L_M P_{M \times M}^\beta B_M^T(t) + F_M^{(2)} B_M^T(t) - L_M Q_{M \times M}^{\beta, \phi} B_M^T(t)$$

$$D^{\mu_1} x(t) = \left[K_M P_{M \times M}^\alpha B_M^T(t) + F_M^{(1)} B_M^T(t) - K_M Q_{M \times M}^{\alpha, \phi} B_M^T(t) \right]$$

$$= K_M \left(P_{M \times M}^\alpha - Q_{M \times M}^{\alpha, \phi} \right) G_{M \times M}^{\mu_1} + F_M^{(1)} G_{M \times M}^{\mu_1} B_M^T(t)$$

$$D^{\nu_1} y(t) = D^{\nu_1} \left[L_M P_{M \times M}^\beta B_M^T(t) + F_M^{(2)} B_M^T(t) - L_M Q_{M \times M}^{\beta, \phi} B_M^T(t) \right]$$

$$= L_M \left(P_{M \times M}^\beta - Q_{M \times M}^{\beta, \phi} \right) G_{M \times M}^{\nu_1} + F_M^{(2)} G_{M \times M}^{\nu_1} B_M^T(t)$$

$$D^{\mu_2} x(t) = D^{\mu_2} \left[K_M P_{M \times M}^\alpha B_M^T(t) + F_M^{(1)} B_M^T(t) - K_M Q_{M \times M}^{\alpha, \phi} B_M^T(t) \right]$$

$$= K_M \left(P_{M \times M}^\alpha - Q_{M \times M}^{\alpha, \phi} \right) G_{M \times M}^{\mu_2} + F_M^{(1)} G_{M \times M}^{\mu_2} B_M^T(t)$$

and

$$D^{\nu_2} y(t) = D^{\nu_2} \left[K_M P_{M \times M}^\beta B_M^T(t) + F_M^{(2)} B_M^T(t) - K_M Q_{M \times M}^{\beta, \phi} B_M^T(t) \right]$$

$$= L_M \left(P_{M \times M}^\beta - Q_{M \times M}^{\beta, \phi} \right) G_{M \times M}^{\nu_2} + F_M^{(2)} G_{M \times M}^{\nu_2} B_M^T(t)$$

$$f(t) = F^{(3)} B_M^T(t), \quad g(t) = F^{(4)} B_M^T(t).$$

Thus system (44) implies that

$$\begin{aligned}
 &K_M B_M^T(t) + A_1 K_M \left(P_{M \times M}^\alpha - Q_{M \times M}^{\alpha, \phi} \right) G_{M \times M}^{\mu_1} + A_1 F_M^{(1)} G_{M \times M}^{\mu_1} B_M^T(t) \\
 &\quad + B_1 L_M \left(P_{M \times M}^\beta - Q_{M \times M}^{\beta, \phi} \right) G_{M \times M}^{\nu_1} + B_1 F_M^{(2)} G_{M \times M}^{\nu_1} B_M^T(t) + C_1 K_M P_{M \times M}^\alpha B_M^T(t) \\
 &\quad + C_1 F_M^{(1)} B_M^T(t) - C_1 K_M Q_{M \times M}^{\alpha, \phi} B_M^T(t) + D_1 L_M P_{M \times M}^\beta B_M^T(t) + D_1 F_M^{(2)} B_M^T(t) \\
 &\quad - D_1 L_M Q_{M \times M}^{\beta, \phi} B_M^T(t) = F^{(3)} B_M^T(t) \\
 &L_M B_M^T(t) + A_2 K_M \left(P_{M \times M}^\alpha - Q_{M \times M}^{\alpha, \phi} \right) G_{M \times M}^{\mu_2} + A_2 F_M^{(1)} G_{M \times M}^{\mu_2} B_M^T(t) \\
 &\quad + B_2 L_M \left(P_{M \times M}^\beta - Q_{M \times M}^{\beta, \phi} \right) G_{M \times M}^{\nu_2} + B_2 F_M^{(2)} G_{M \times M}^{\nu_2} B_M^T(t) + C_2 K_M P_{M \times M}^\alpha B_M^T(t) \\
 &\quad + C_2 F_M^{(1)} B_M^T(t) - C_2 K_M Q_{M \times M}^{\alpha, \phi} B_M^T(t) + D_2 L_M P_{M \times M}^\beta B_M^T(t) + D_2 F_M^{(2)} B_M^T(t) \\
 &\quad - D_2 L_M Q_{M \times M}^{\beta, \phi} B_M^T(t) = F^{(4)} B_M^T(t).
 \end{aligned} \tag{46}$$

Rearranging the terms in the above system and using the following notation for simplicity in Eq. (46)

$$\begin{aligned}
 \hat{Q}_{M \times M}^\alpha &= A_1 \left(P_{M \times M}^\alpha - Q_{M \times M}^{\alpha, \phi} \right) G_{M \times M}^{\mu_1} + C_1 \left(P_{M \times M}^\alpha - Q_{M \times M}^{\alpha, \phi} \right) \\
 \hat{Q}_{M \times M}^\beta &= B_1 \left(P_{M \times M}^\beta - Q_{M \times M}^{\beta, \phi} \right) G_{M \times M}^{\nu_1} + D_1 \left(P_{M \times M}^\beta - Q_{M \times M}^{\beta, \phi} \right) \\
 \hat{R}_{M \times M}^\alpha &= A_2 \left(P_{M \times M}^\alpha - Q_{M \times M}^{\alpha, \phi} \right) G_{M \times M}^{\mu_2} + C_2 \left(P_{M \times M}^\alpha - Q_{M \times M}^{\alpha, \phi} \right) \\
 \hat{R}_{M \times M}^\beta &= B_2 \left(P_{M \times M}^\beta - Q_{M \times M}^{\beta, \phi} \right) G_{M \times M}^{\nu_2} + D_2 \left(P_{M \times M}^\beta - Q_{M \times M}^{\beta, \phi} \right) \\
 F_M &= A_1 F_M^{(1)} G_{M \times M}^{\mu_1} + B_1 F_M^{(2)} G_{M \times M}^{\nu_1} + C_1 F_M^{(1)} + F_M^{(2)} - D_1 F_M^{(3)} \\
 G_M &= A_2 F_M^{(1)} G_{M \times M}^{\mu_2} + B_2 F_M^{(2)} G_{M \times M}^{\nu_2} + C_2 F_M^{(1)} + D_2 F_M^{(2)} - F_M^{(4)},
 \end{aligned}$$

the above system (46) becomes

$$\begin{aligned}
 &K_M B_M^T(t) + K_M \hat{Q}_{M \times M}^\alpha B_M^T(t) + L_M \hat{Q}_{M \times M}^\beta B_M^T(t) + F_M B_M^T(t) = 0 \\
 &L_M B_M^T(t) + K_M \hat{R}_{M \times M}^\alpha B_M^T(t) + L_M \hat{R}_{M \times M}^\beta B_M^T(t) + G_M B_M^T(t) = 0 \\
 &[K_M \ L_M] \begin{bmatrix} B_M^T(t) & 0 \\ 0 & B_M^T(t) \end{bmatrix} + [K_M \ L_M] \begin{bmatrix} \hat{Q}_{M \times M}^\alpha & 0 \\ 0 & \hat{R}_{M \times M}^\beta \end{bmatrix} \begin{bmatrix} B_M^T(t) & 0 \\ 0 & B_M^T(t) \end{bmatrix} \\
 &+ [K_M \ L_M] \begin{bmatrix} 0 & \hat{R}_{M \times M}^\alpha \\ \hat{Q}_{M \times M}^\beta & 0 \end{bmatrix} \begin{bmatrix} B_M^T(t) & 0 \\ 0 & B_M^T(t) \end{bmatrix} + [F_M \ G_M] \begin{bmatrix} B_M^T(t) & 0 \\ 0 & B_M^T(t) \end{bmatrix} = 0 \\
 &[K_M \ L_M] + [K_M \ L_M] \begin{bmatrix} \hat{Q}_{M \times M}^\alpha & \hat{R}_{M \times M}^\alpha \\ \hat{Q}_{M \times M}^\beta & \hat{R}_{M \times M}^\beta \end{bmatrix} + [F_M \ G_M] = 0,
 \end{aligned} \tag{47}$$

which is an algebraic equation that can be easily solved by using Matlab functional solver or Mathematica for unknown matrix $[K_M \ L_M]$. Calculating the coefficient matrix K_M, L_M and putting it in equations

$$x(t) = K_M P_{M \times M}^\alpha B_M^T(t) + F_M^{(1)} B_M^T(t) - K_M Q_{M \times M}^{\alpha, \phi} B_M^T(t)$$

$$y(t) = L_M P_{M \times M}^\beta B_M^T(t) + F_M^{(2)} B_M^T(t) - L_M Q_{M \times M}^{\beta, \phi} B_M^T(t),$$

we get the required approximate solution.

4. Applications of our method to some examples

Example 4.1. Consider

$$D^\alpha y(t) + c_1 D^\nu y(t) + c_2 y(t) = f(t), \quad 1 < \alpha < 2 \tag{48}$$

subject to the boundary conditions

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

Solution: We solve this problem under the following parameters sets defined as $S_1 = \{\alpha = 2, \nu = 1, c_1 = 1, c_2 = 1\}$, $S_2 = \{\alpha = 1.8, \nu = 0.8, c_1 = 10, c_2 = 100\}$, $S_3 = \{\alpha = 1.5, \nu = 0.5, c_1 = 1/10, c_2 = 1/100\}$, and select source term for S_1 as

$$f_1(t) = t^6(t-1)^3 + t^6((72t-168)+126) - 30t^4 + 3t^5(3t-2)(t-1)^2 \tag{49}$$

$$f_2(t) = \frac{11147682583723703125t^{\frac{21}{5}}(1750t^3 - 4200t^2 + 3255t - 806)}{406548945561989414912}$$

$$+ \frac{278692064593092578125t^{\frac{26}{5}}(5250t^3 - 14350t^2 + 12915t - 3813)}{25002760152062349017088}$$

$$+ 100t^6(t-1)^3, \tag{50}$$

$$f_3(t) = \frac{5081767996463981t^{\frac{9}{2}}(1344t^3 - 3360t^2 + 2730t - 715)}{264146673456906240}$$

$$+ \frac{5081767996463981t^{\frac{11}{2}}(1344t^3 - 3808t^2 + 3570t - 1105)}{22452467243837030400} + \frac{t^6(t-1)^3}{100}. \tag{51}$$

The exact solution of the above problem is

$$y(t) = t^6(t-1)^3.$$

We solve this problem with the proposed method under different sets of parameters as defined in S_1, S_2, S_3 . The observation and simulation demonstrate that the solution obtained with the proposed method is highly accurate. The comparison of exact solution with approximate solution obtained using the parameters set S_1 is displayed in **Figure 1** subplot (a), while in **Figure 1** subplot (b) we plot the absolute difference between the exact and approximate solutions using different scale levels. One can easily observe that the absolute error is much less than 10^{-12} . The order of derivatives in this set is an integer.

By solving the problem under parameters set S_2 and S_3 , we observe the same phenomena. The approximate solution matches very well with the exact solution. See **Figures 2 and 3** respectively.

Example 4.2. Consider

$$D^\alpha y(t) - 2D^{0.9}y(t) - 3y(t) = -4 \cos(2t) - 7 \sin(2t) \tag{52}$$

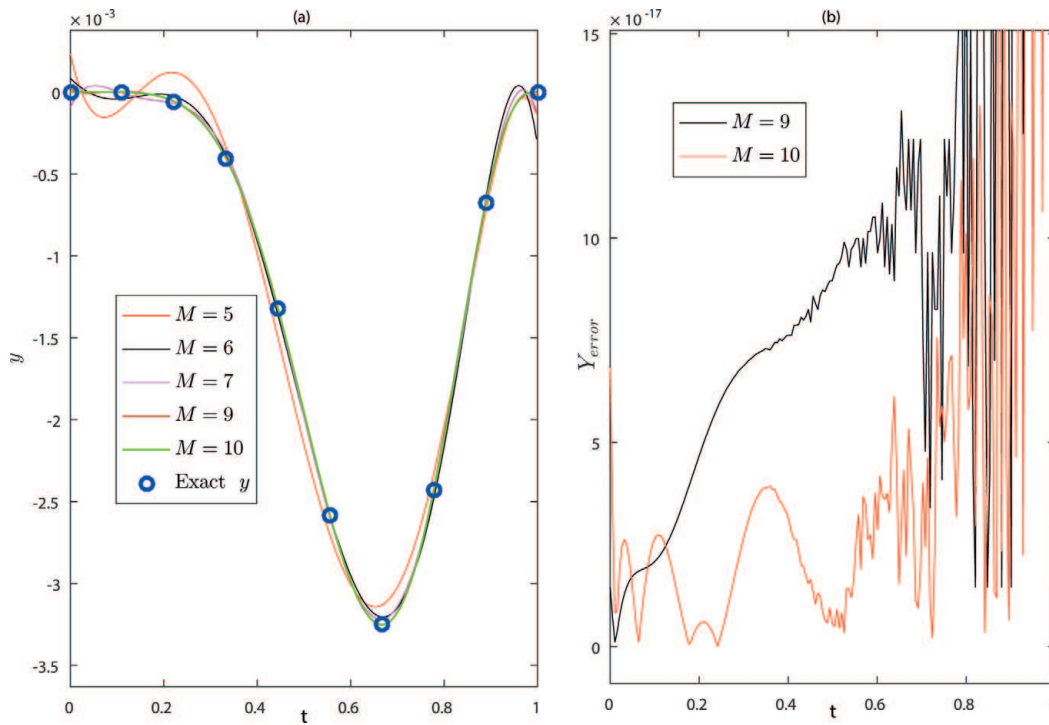


Figure 1. (a) Comparison of exact and approximate solution of Example 4.1, under parameters set S_1 . (b) Absolute error in the approximate solution of Example 4.1, under parameters set S_1 .

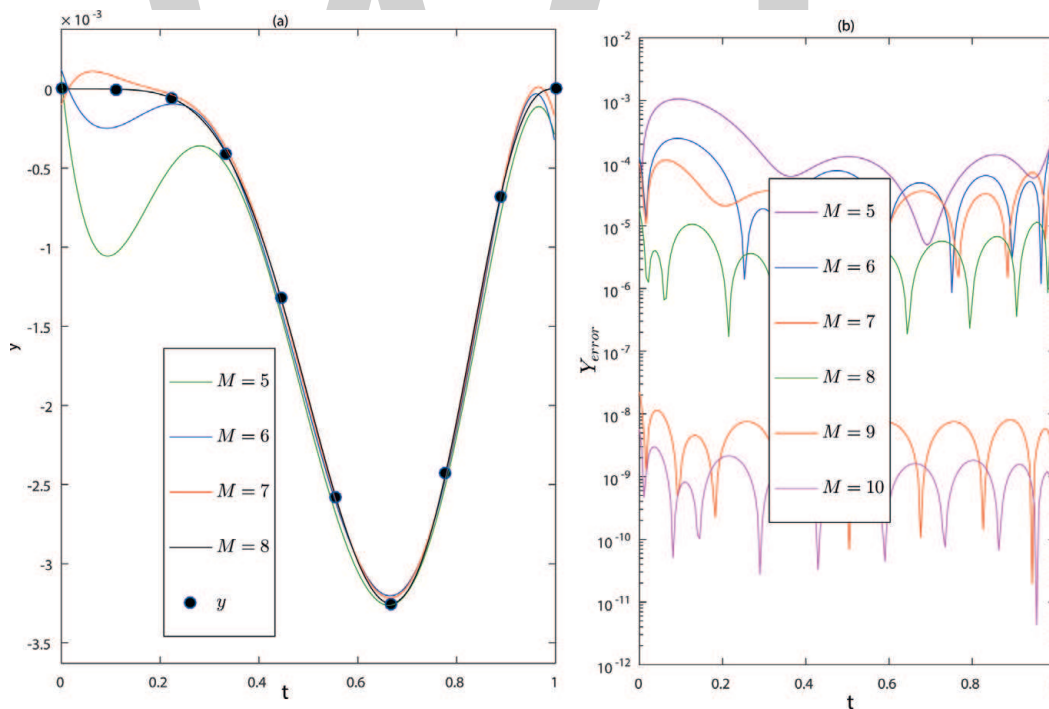


Figure 2. (a) Comparison of exact and approximate solution of Example 4.1, under parameters set S_2 . (b) Absolute error in the approximate solution of Example 4.1, under parameters set S_2 .

subject to the boundary conditions

$$y(0) = 0, \quad y(1) = \sin(2).$$

Solution: The exact solution of the above problem is $y(t) = \sin(2t)$, when $\alpha = 2$. However the exact solution at fractional order is not known. We use the well-known property of FDEs that when $\alpha \rightarrow 2$, the approximate solution approaches the exact solution for the evaluations of approximate solutions and check the accuracy by using different scale levels. By increasing the scale level M , the accuracy is also increased. By the

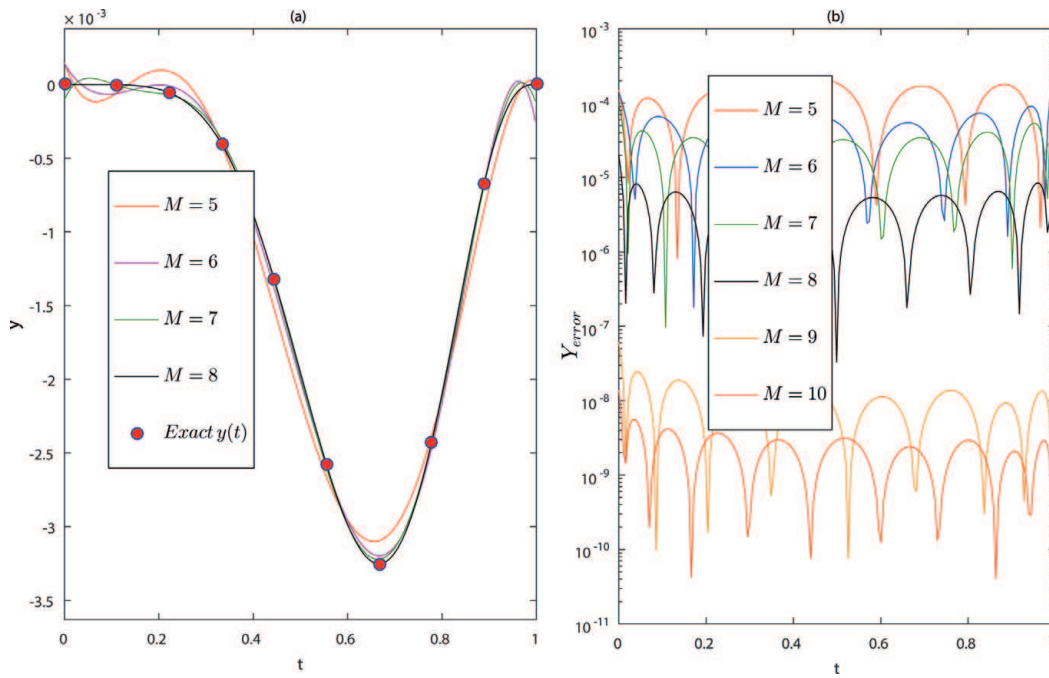


Figure 3. (a) Comparison of exact and approximate solution of Example 4.1, under parameters set S_3 . (b) Absolute error in the approximate solution of Example 4.1, under parameters set S_3 .

proposed method, the graph of exact and approximate solutions for different values of M and at $\alpha = 1.7$ is shown in **Figure 4**. From the plot, we observe that the approximate solution becomes equal to the exact solution at $\alpha = 2$. We approximate the error of the method at different scale levels and record that when scale level increases the absolute error decreases as shown in **Figure 4** subplot (b) and accuracy approaches 10^{-9} , which is a highly acceptable figure. For convergence of our proposed method, we examined the quantity $\int_0^1 |y_{exact} - y_{approx}| dt$ for different values of M and observed that the norm of error decreases with a high speed with the increase of scale level M as shown in **Figure 4b**.

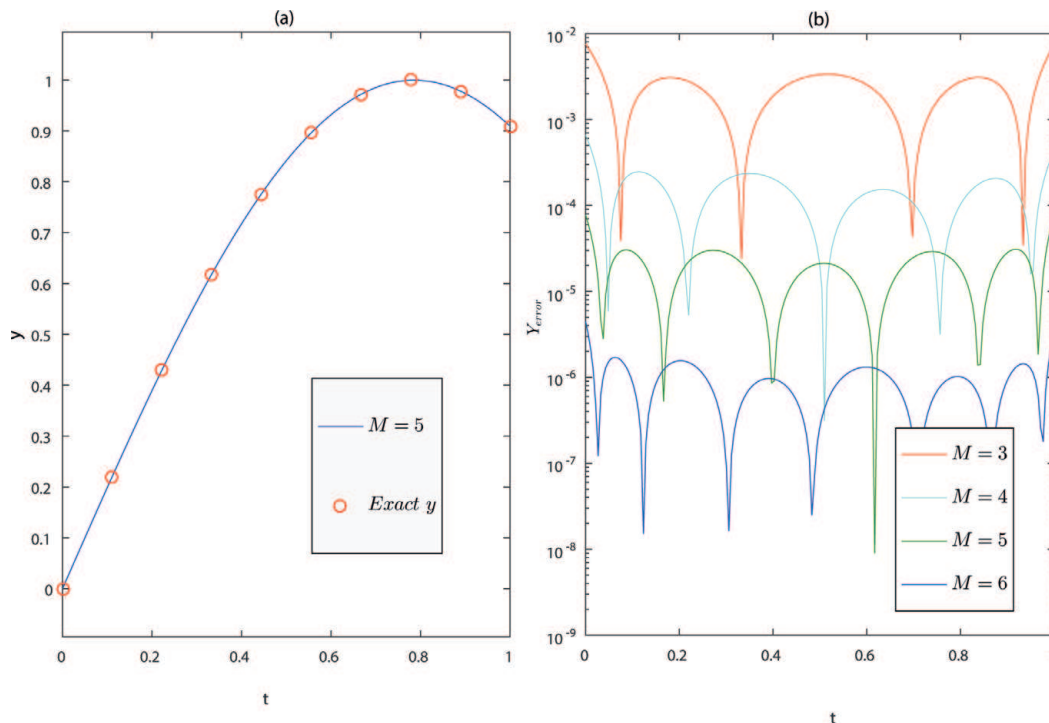


Figure 4. (a) Comparison of exact and approximate solution of Example 4.2. (b) Absolute error for different scale level M of Example 4.2.

Example 4.3. Consider the following coupled system of fractional differential equations

$$\begin{aligned} D^{1.8}x(t) + Dx(t) + 9D^{0.8}y(t) + 2x(t) - y(t) &= f(t) \\ D^{1.8}y(t) - 6D^{0.8}x(t) + Dy(t) - x(t) &= g(t) \end{aligned} \tag{53}$$

subject to the boundary conditions

$$x(0) = 1, \quad x(1) = 2 \quad \text{and} \quad y(0) = 2, \quad y(1) = 2.$$

Solution: The exact solution is

$$x(t) = t^5(1 - t), \quad y(t) = t^4(1 - t).$$

We approximate the solution of this problem with this new method. The source terms are given by

$$\begin{aligned} f(t) = 2t^5(t - 1) - t^4(t - 1) + t^4(6t - 5) - \frac{2229536516744740625t^{16}(10t - 7)}{1008806316530991104} \\ + \frac{1337721910046844375t^{16}(25t - 21)}{1008806316530991104} \end{aligned} \tag{54}$$

$$\begin{aligned} g(t) = t^3(5t - 4) - t^5(t - 1) - \frac{11147682583723703125t^{21}(15t - 13)}{6557241057451442176} \\ - \frac{89181460669789625t^{11}(25t - 16)}{144115188075855872}. \end{aligned} \tag{55}$$

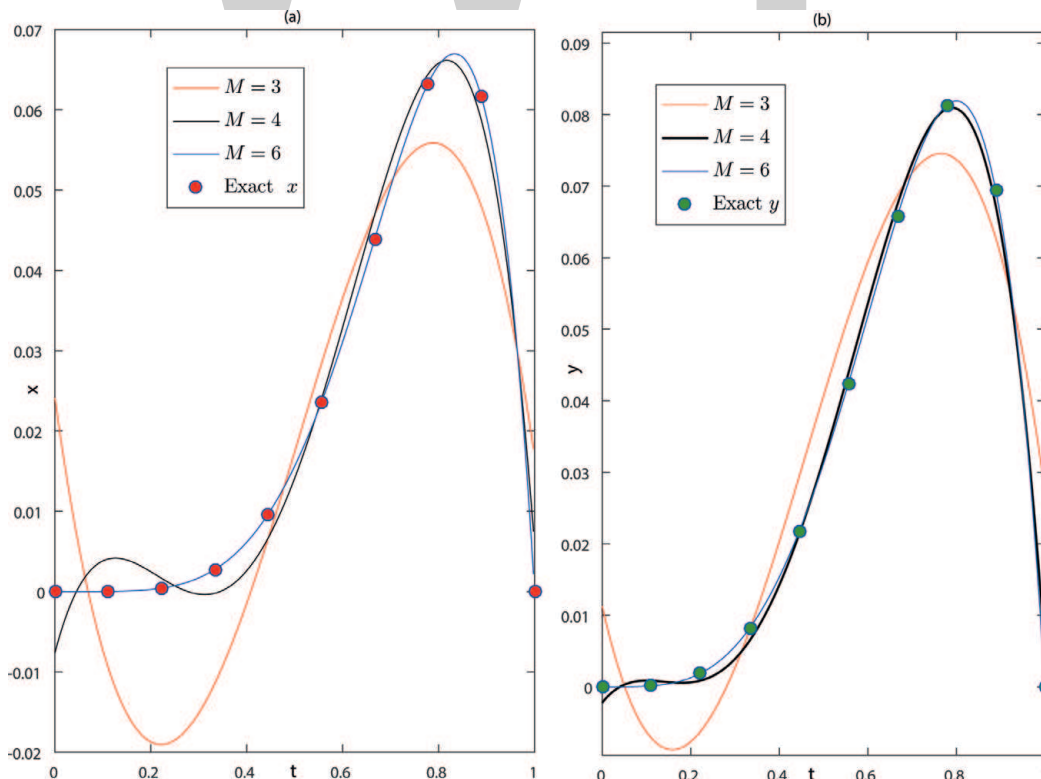


Figure 5. Comparison of exact and approximate solution of Example 4.3 for different scale level M.

In the given **Figure 5**, we have shown the comparison of exact $x(t), y(t)$ and approximate $x(t), y(t)$ in subplot (a) and (b) respectively.

As expected, the method provides a very good approximation to the solution of the problem. At first, we approximate the solutions of the problem at $\alpha = 2$ because the exact solution at $\alpha = 2$ is known. We observe that at very small scale levels, the method provides a very good approximation to the solution. We approximate the absolute error by the formula

$$X_{error} = |x_{exact} - x_{approx}|.$$

and

$$Y_{error} = |y_{exact} - y_{approx}|.$$

We approximate the absolute error at different scale level of M , and observe that the absolute error is much less than 10^{-10} at scale level $M = 7$, see **Figure 6**. We also approximate the solution at some fractional value of α and observe that as $\alpha \rightarrow 2$ the approximate solution approaches the exact solution, which guarantees the accuracy of the solution at fractional value of α . **Figure 6** shows this phenomenon. In **Figure 6**, the subplot (a) represents the absolute error of $x(t)$ and subplot (b) represents the absolute error of $y(t)$.

Example 4.4. Consider the following coupled system

$$\begin{aligned} D^{1.8}x(t) - x(t) + 3y(t) &= f(t) \\ D^{1.8}y(t) + 4x(t) - 2y(t) &= g(t), \end{aligned} \tag{56}$$

subject to the boundary conditions

$$x(0) = -1, \quad x(1) = -1 \quad \text{and} \quad y(0) = -1, \quad y(1) = -1.$$

Solution: The exact solution for $\alpha = \beta = 2$ is

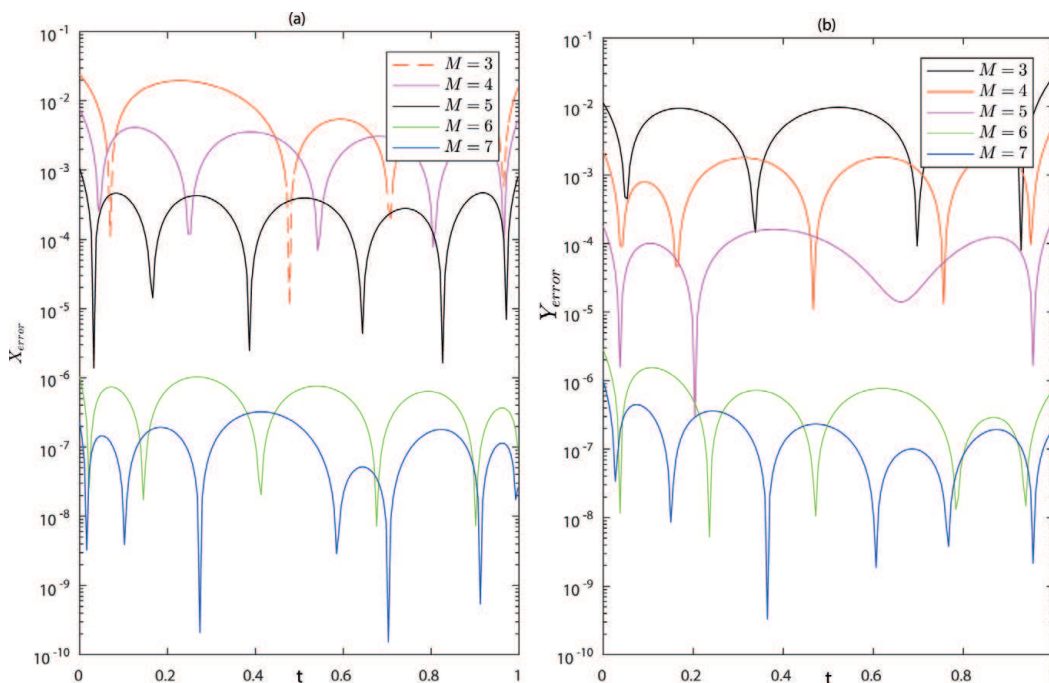


Figure 6. Absolute error in approximate solutions at different scale level $M = 3:7$ for Example 4.3.

$$x(t) = t^5 - t^4 - 1, \quad \text{and} \quad y(t) = t^4 - t^3 - 2. \quad (57)$$

The source terms are given by

$$f(t) = \frac{445907303348948125t^{3.2}(25t - 21)}{3026418949592973312} + t^3 - 4t^4 + 3t^5 - 2,$$

$$g(t) = \frac{89181460669789625t^{2.5}(5t - 4)}{14411518807585872} - 4t^3 + 6t^4 - 2t^5 - 2.$$

Approximating the solution with the proposed method, we observe that our scheme gives high accuracy of approximate solution. In **Figure 7**, we plot the exact solutions together with the approximate solutions in **Figure 7(a)** and **(b)** for $x(t)$ and $y(t)$, respectively. We see from the subplots (a) and (b) that our approximations have close agreement to that of exact solutions. This accuracy may be made better by increasing scale level. Further, one can observe that absolute error is below 10^{-10} in **Figure 8**, which indicates better accuracy of our proposed method for such types of practical problems of applied sciences.

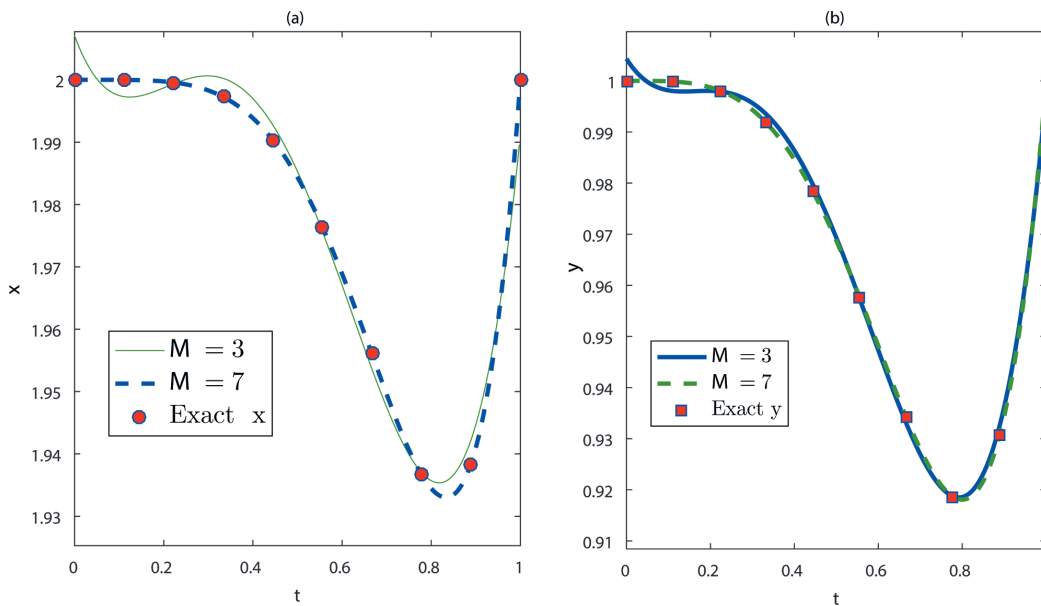


Figure 7. Comparison of exact and approximate solution at scale level $M = 3, 7$ for Example 4.4.

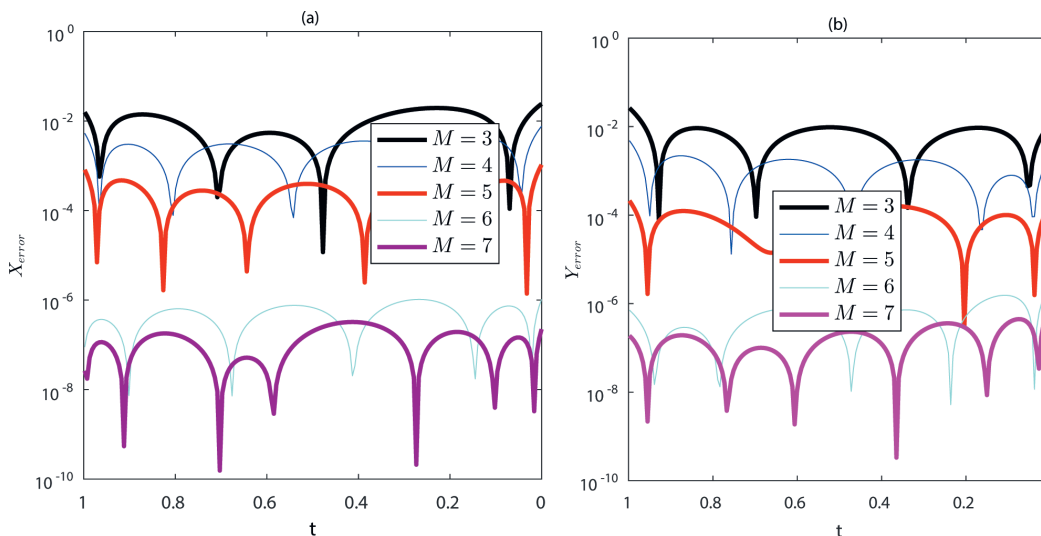


Figure 8. Absolute error for different scale level $M = 3:7$ for Example 4.4.

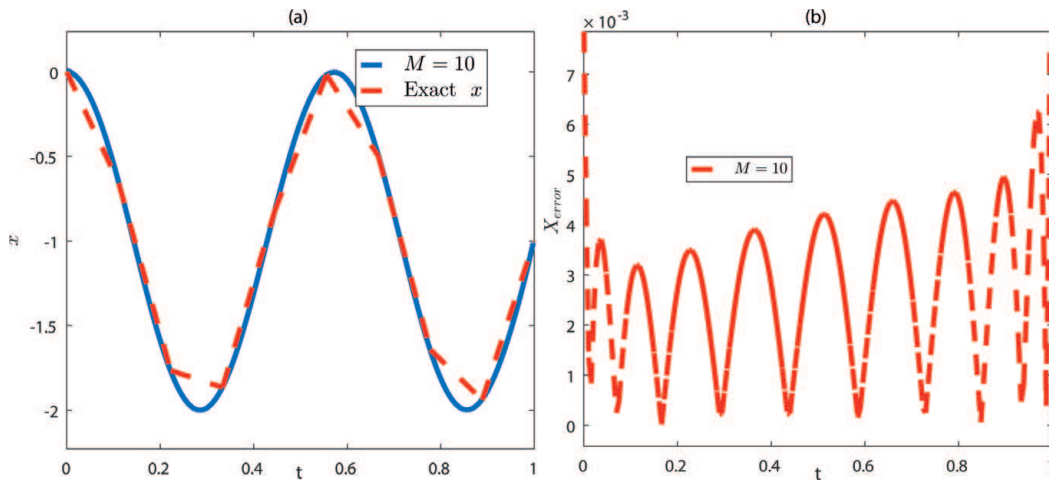


Figure 9. (a) Comparison of exact and approximate solution at scale level $M = 10, \omega = 3.5, \alpha = 2, \nu = 1$ for Example 4.5. (b) Absolute error at $M = 10$.

ω	M	α	ν	$\ x_{app} - x_{ex}\ $ at BM	$\ x_{app} - x_{ex}\ $ at WM	$\ x_{app} - x_{ex}\ $ at JM
0.5	10	2	1	7.000(-3)	2.966(-1)	1.500(-2)
1.5	15	1.6	0.9	6.091(-3)	4.918(-2)	1.623(-1)
2.0	20	1.8	0.8	1.237(-3)	2.108(-2)	2.723(-2)
3.5	25	1.9	0.7	1.008(-3)	5.795(-2)	1.813(-3)

Table 1. Comparison of solution between Legendre wavelet method (LWM) [47], Jacobi polynomial method (JM) and Bernstein polynomials method (BM) for Example 4.5.

In Figure 8, the subplot (a) represents absolute error for $x(t)$ while subplot (b) represents the same quantity for $y(t)$. From the subplots, we see that maximum absolute error for our proposed method for the given problem (4.4) is below 10^{-10} . This is very small and justifies the efficiency of our constructed method.

Example 4.5. Consider the boundary value problem

$$\begin{aligned}
 D^\alpha x(t) + (\omega\pi)^2 D^\nu x(t) + x(t) &= -\omega\pi(\sin(\omega\pi t) + \omega\pi) \\
 x(0) = 0, \quad x(1) &= -2.
 \end{aligned}
 \tag{58}$$

Taking $\alpha = 2, \nu = 1$ and $\omega = 1, 3, 5, \dots$, the exact solution is given by

$$x(t) = \cos(\omega\pi t) - 1.$$

We plot the comparison between exact and approximate solutions to the given example at $M = 10$ and corresponding to $\omega = 3.5, \alpha = 2, \beta = 1$. Further, we approximate the solution through Legendre wavelet method (LWM) [47], Jacobi polynomial method (JM) and Bernstein polynomials method (BM), as shown in Figure 9.

From Table 1, we see that Bernstein polynomials also provide excellent solutions to fractional differential equations [48].

5. Conclusion and future work

The above analysis and discussion take us to the conclusion that the new method is very efficient for the solution of boundary value problems as well as initial value

problems including coupled systems of fractional differential equations. One can easily extend the method for obtaining the solution of such types of problems with other kinds of boundary and initial conditions. Bernstein polynomials also give best approximate solutions to fractional order differential equations like Legendre wavelet method (LWM), approximation by Jacobi polynomial method (JPM), etc. The new operational matrices obtained in this method can easily be extended to two-dimensional and higher dimensional cases, which will help in the solution of fractional order partial differential equations. Also, we compare our result to that of approximate methods for different scale levels. We observed that the proposed method is also an accurate technique to handle numerical solutions.

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Spectral Observations of PM10 Fluctuations in the Hilbert Space

Thomas Plocoste and Rudy Calif

Abstract

During the last 20 years, many megacities have experienced air pollution leading to negative impacts on human health. In the Caribbean region, air quality is widely affected by African dust which causes several diseases, particularly, respiratory diseases. This is why it is crucial to improve the understanding of PM10 fluctuations in order to elaborate strategies and construct tools to predict dust events. A first step consists to characterize the dynamical properties of PM10 fluctuations, for instance, to highlight possible scaling in PM10 density power spectrum. For that, the scale-invariant properties of PM10 daily time series during 6 years are investigated through the theoretical Hilbert frame. Thereafter, the Hilbert spectrum in time-frequency domain is considered. The choice of theoretical frame must be relevant. A comparative analysis is also provided between the results achieved in the Hilbert and Fourier spaces.

Keywords: PM10 data, empirical mode decomposition, Hilbert spectral analysis, time-frequency representation, Fourier space

1. Introduction

Generally, the concentration of air pollutants varies and is impacted by the local pollutant emission levels and meteorological and topographical conditions [1, 2]. Particulate matter (PM) is a complex mixture of elemental and organic carbon, ammonium, nitrates, sulfates, mineral dust, trace elements, and water [3]. PM with an aerodynamic diameter of $<10 \mu\text{m}$, i.e., PM10, are well known for their impact on human health [4]. Many studies have highlighted that exposure to PM increases the number of hospital admissions for cardiovascular disease, acute bronchitis, asthma attacks, respiratory disease, and congestive heart failure [5–8]. In the Caribbean area, one of the main emitters of PM10 is from large-scale sources, i.e., African dust [9]. Knowledge of the dynamics of PM10 process is crucial to elaborate strategies and construct tools to predict dust events. The time-frequency distribution of a signal provides information about how the spectral content of a signal evolves with time, thus providing an ideal tool to dissect, analyze, and interpret nonstationary signals [10]. Contrary to classical methods, the need of a time-frequency representation (TFR) is stemmed from the inadequacy of either time domain or frequency domain analysis to fully describe the nature of nonstationary signals [10]. In literature, there are numerous methods to obtain energy density as a function of time and frequency simultaneously as the short-time Fourier transform (STFT), Hilbert-Huang transform (HHT), and wavelet transform (WT) [10–12].

In this study, the scaling properties of PM10 data are firstly analyzed, and then the TFR is investigated. In order to highlight the performance of the Hilbert space, an analysis of PM10 data was also performed in the Fourier space.

This chapter is organized as follows. Section 2 presents PM10 data analyzed in this study. Section 3 describes the methods applied in order to investigate PM10 dynamics. Section 4 comments on the results obtained and then discusses them.

2. Experimental data

Guadeloupe archipelago is a French West Indies island located in the middle of the Caribbean basin, i.e., 16.25°N latitude and 61.58°W longitude, which experiences a tropical and humid climate [13, 14]. The time series analyzed here belong to Guadeloupe air quality network which is managed by the Gwad'Air agency (<http://www.gwadair.fr/>). PM10 concentrations are measured at Pointe-à-Pitre (16.2422°N 61.5414°W) using the Thermo Scientific tapered element oscillating microbalance (TEOM) models 1400ab and 1400-FDMS. Hourly PM10 concentrations were sampled during the period from 1 January 2005 to 31 December 2010. We processed these data into daily average concentrations. In total, there are 2150 daily averaged data points available continuously for 6 years. **Figure 1** displays PM10 daily signal illustrating huge fluctuations and thus indicating a strong variability. These strong oscillations observed in the middle of each year are attributed to PM10 related to dust outbreaks coming from the African coast from May to September [9]. For the rest of the year, PM10 is mainly generated by anthropogenic pollution [15].

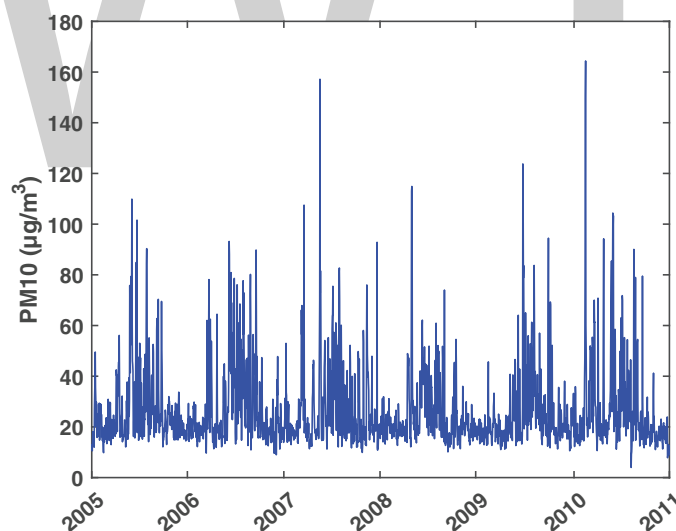


Figure 1.

Illustration of PM10 daily average concentrations between 2005 and 2010, highlighting intermittent burst events with huge fluctuations.

3. Methods

3.1 Scaling analysis (1D representation)

The description of natural phenomena by the study of statistical scale laws is not recent [16]. Self-similarity of complex systems has been widely observed in nature and is the simplest form of scale invariance. A scale invariance can be detected by

computing of power spectral density (PSD). The PSD separates and measures the amount of variability occurring in different frequency bands. In this study, PSD are estimated through the Fourier and Hilbert spaces.

3.1.1 Fourier analysis

In order to investigate the scaling properties of PM10 data, classically the discrete Fourier transform of the times series considered is computed. The expression of Fourier transform $X(f)$ for a process $x(t)$ is recalled here. An N point-long interval is used to construct the value at frequency domain point f , X_f [17]:

$$X(f) = \int_{-T}^{+T} x(t)e^{-2\pi ift} dt \quad (1)$$

Thus, the analytical expression of $X(f)$ is [18]

$$|X(f)| = \sqrt{\text{Re}^2(X(f)) + \text{Im}^2(X(f))} \quad (2)$$

Consequently the power spectral density $E(f)$ is estimated by computing the following expression:

$$E(f) = |X(f)|^2 \quad (3)$$

3.1.2 Hilbert analysis

To determine the scale invariance of a given time series in a joint amplitude-frequency space, the Hilbert-Huang transform [19, 20] is performed. HHT can be summarized in two steps: (i) empirical mode decomposition (EMD) and (ii) Hilbert spectral analysis (HSA). Empirical mode decomposition is a powerful tool to separate a nonlinear and nonstationary time series into a sum of intrinsic mode functions (IMF) without a priori basis as required by traditional Fourier-based method [19–21]. An IMF must satisfy the following two conditions: (i) the difference between the number of local extrema and the number of zero-crossings must be zero or one, and (ii) the local maxima and the envelope defined by the local minima are close to zero. Therefore, the original signal $x(t)$ is decomposed into a sum of $n-1$ IMF modes with the residual $r_n(t)$:

$$x(t) = \sum_{m=1}^{n-1} C_m(t) + r_n(t) \quad (4)$$

To obtain a physically significant IMF, this selection process must be stopped by a certain criterion. For more details, EMD decomposition is widely described in the literature [19–23].

To characterize the time-frequency energy distribution from the original signal $x(t)$, HSA is applied on each obtained IMF component $C_m(t)$ to extract the instantaneous amplitude and frequency [19, 24]. The Hilbert transform is defined by:

$$\tilde{C}_m(t) = \frac{1}{\pi} P \int_{-\infty}^{+\infty} \frac{C_m(t')}{t - t'} dt' \quad (5)$$

with P the Cauchy principal value [24, 25]. We can specify an analytical signal z for each IMF mode $C_m(t)$ with

$$Z_m(t) = C_m(t) + j\tilde{C}_m(t) = A_m(t)e^{j\varphi_m(t)} \quad (6)$$

where $A_m(t) = |Z_m(t)| = \sqrt{C_m(t)^2 + \tilde{C}_m(t)^2}$ describes an amplitude and $\varphi_m(t) = \arg(z) = \arctan\left[\frac{\tilde{C}_m(t)}{C_m(t)}\right]$ represents the phase function of IMF modes. Consequently, the instantaneous frequency $\omega_m(t)$ is defined from the phase $\varphi_m(t)$ by

$$\omega_m(t) = \frac{1}{2\pi} \frac{d\varphi_m(t)}{dt} \quad (7)$$

Thus, the original signal $x(t)$ can be expressed as

$$x(t) = \text{Re} \sum_{m=1}^N A_m(t)e^{j\varphi_m(t)} = \text{Re} \sum_{m=1}^N A_m(t)e^{j \int_{-\infty}^t \omega_m(t) dt} \quad (8)$$

where Re is a part real [19, 20, 26].

Due to the simultaneous representation of frequency modulation and amplitude modulation, the HHT can be considered as a generalization of the Fourier transform [19, 20]. The energy in a time-frequency space is designated as the Hilbert spectrum with $H(\omega, t) = \mathcal{A}^2(\omega, t)$. The Hilbert marginal spectrum $h(\omega)$ is defined by

$$h(\omega) = \frac{1}{T} \int_0^T H(\omega, t) dt \quad (9)$$

where T is the total data length. The Hilbert spectrum $H(\omega, t)$ gives a measure of amplitude from each frequency and time, while the marginal spectrum $h(\omega)$ gives a measure of the total amplitude from each frequency [27]. As a result, the marginal spectrum can be compared to the Fourier spectrum [19, 20].

In conclusion, for a scale-invariant process, the Fourier $E(f)$ and the Hilbert $h(\omega)$ spectral densities obtained follow a power law over a range of frequencies:

$$E(f) \sim f^{-\beta_f} \quad (10)$$

$$h(\omega) \sim \omega^{-\beta_h} \quad (11)$$

where f and ω are the frequencies and β_f and β_h are the spectral exponents, respectively, in the Fourier and Hilbert spaces. It reveals the scale-free memory effect as a power law dependence of the frequency distribution. Consequently, β_f and β_h contain information about the degree of stationarity of the studied parameter [16, 28, 29]:

- If β_f or $\beta_h < 1$, the process is stationary.
- If β_f or $\beta_h > 1$, the process is nonstationary.
- If $1 < \beta_f$ or $\beta_h < 3$, the process is nonstationary with increments stationary.
- Spectral analysis has been widely applied in various research fields [30–34].

3.2 Time-frequency representation (2D representation)

3.2.1 Spectrogram

The spectrogram (SPEC) of a signal $x(t)$ is defined as the squared magnitudes of the STFT as shown in Eq. (12) [12]:

$$\text{SPEC}_x(t, f) = |S_x(t, f)|^2 \quad (12)$$

where $S_x(t, f) = \int_{-\infty}^{+\infty} x(\tau)w(\tau - t)e^{-jf\tau}d\tau$ is the STFT of $x(t)$, $w(\tau)$ is a window (e.g., Hanning, rectangular, Hamming), t is time, and f is frequency.

As depicted in Eq. (13), SPEC roughly describes the energy density of the signal at point (t, f) [12]:

$$\int_{-\infty}^{+\infty} \int_{-\infty}^{+\infty} \text{SPEC}_x(t, f) dt df = \int_{-\infty}^{+\infty} |x(t)|^2 dt \quad (13)$$

The SPEC has been applied successfully in various research fields [12, 35–37]. The main advantages of SPEC are an easily understanding interpretation, and it allows a fast computation. However, the main drawback of SPEC is the same as that of the STFT [12]. Indeed, there is a trade-off between time and frequency resolution.

3.2.2 Hilbert spectrum

The Hilbert spectrum (HS) is a joint time-frequency representation introduced by [19]. It is important to notice that the two important tools (i.e., EMD and HS) for exploratory analysis of the data are provided by HSA method. This approach was applied successfully in various research fields as fault diagnosis for rolling bearing [11], turbulence [38], environment [34, 39], and geophysics [40], to cite a few.

4. Results

4.1 Scaling properties

In order to identify the presence of scaling in PM10 time series, the PSD is estimated in the Hilbert and Fourier spaces. **Figure 2** depicts the power spectral density provided by the Hilbert transform and the Fourier transform. On this figure, we try to detect a power law behavior of the form $h(\omega) \sim \omega^{-\beta_h}$ and $E(f) \sim f^{-\beta_f}$ where β_h and β_f are, respectively, the spectral exponents in the Hilbert and Fourier spaces. On the frequency range $2.09 \times 10^{-7} \leq f \leq 4.57 \times 10^{-5}$ Hz which corresponds to time scales $6.1 \text{ hours} \leq T \leq 55.4 \text{ days}$, a power law behavior is clearly

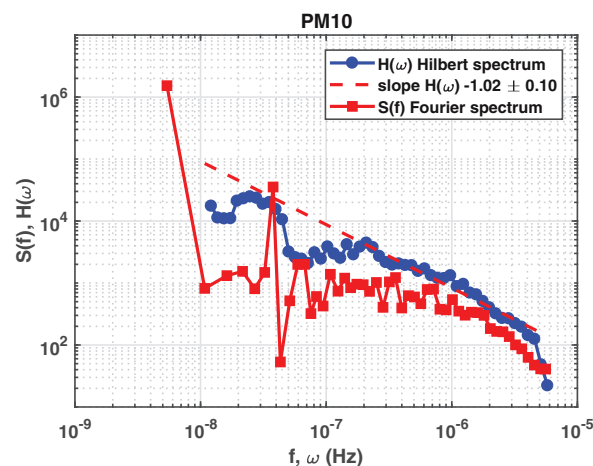


Figure 2.

The spectrum of PM10 time series in the Hilbert space and the Fourier space. A power law behavior is significant only in the Hilbert space.

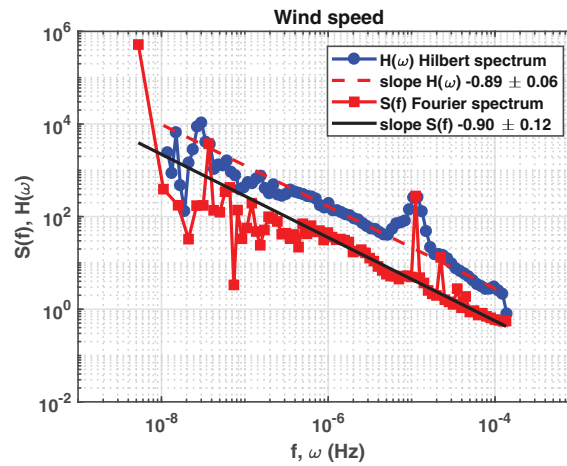


Figure 3.

The spectrum of wind speed in the Hilbert space and the Fourier space. A power law behavior is observed in both spaces.

noticed in the Hilbert space with an estimated spectral exponent $\beta_h = 1.02 \pm 0.10$. β_h is equal to 1 power law scaling observed in the mesoscale range [41]. In the Fourier space, this power law is not significant. This is due to the existence of intermittent dust events with huge fluctuations in PM10 data (see **Figure 1**). Indeed, the Fourier transform is a linear asymptotic approach which requires high-order harmonic components to mimic nonlinear and nonstationary process [42]. Thus, the high-order harmonics may lead an artificial energy transfer flux from a large scale (low frequency) to a small scale (high frequency) in the Fourier space. Consequently, the Fourier-based spectrum may be contaminated by this artificial energy flux [42]. The artificial energy transfer may give a less steep power spectrum as we observed in **Figure 2**. By contrast, combined with the EMD method, HSA has very local abilities both in physical and spectral domains and does not require any higher-order harmonic components to simulate the nonlinear and nonstationary events. As a consequence, HSA method may provide a more accurate scaling exponent and singularity spectrum [42].

According to [43], wind speed dominates the amount of pollutant dispersion in the atmospheric boundary layer. In addition, this meteorological parameter could also transport PM10 from large-scale sources, i.e., African dust [9]. To complete our results, we used hourly wind speed measurements provided by the French weather office (Météo France Guadeloupe) located at Abymes (16.2630°N 61.5147°W). PM10 and wind speed measurements are very close, i.e., ≈ 8.1 km of distance, and performed at the center of the island under the same atmospheric conditions [2]. **Figure 3** illustrates the PSD provided by the Hilbert transform and the Fourier transform for wind speed data. This time, a power law behavior is observed in both spaces on the same frequency range $3.54 \times 10^{-7} \leq f \leq 1.36 \times 10^{-4}$ Hz which corresponds to time scales $2.1 \text{ hours} \leq T \leq 32.7 \text{ days}$. Contrary to PM10 which is a passive scalar, wind speed is a vector quantity. The estimated spectral exponents are identical with, respectively, 0.89 ± 0.06 and 0.90 ± 0.12 in the Hilbert and Fourier spaces. As PM10, spectral exponent values are also close to -1 . For wind speed, at low frequencies, a spectrum close to the 1 power law is likely occurs close to a rough surface, due to a strong interaction between the mean flow vorticity and the fluctuating vorticity [44, 45].

4.2 Time-frequency domain

The TFR in the Fourier and Hilbert spaces are, respectively, illustrated in **Figures 4** and **5**. Both figures show a color gradient from strong energy (in red) to

weak energy (in blue). This highlights the energy activity related to PM10 concentrations during the study period. Such an approach gives the possibility of tracking the evolution of PM10 data spectral content in time, which is typically represented by variations of the amplitudes and frequencies of the components from which the signal is composed [46].

On **Figure 4**, strong energies are observed throughout the years with slight fluctuations on the frequency range $0 \leq f \leq 1 \times 10^{-6}$ Hz. For $f > 4 \times 10^{-6}$ Hz, strong energies are also noticed in the middle of each year and at the beginning of 2010 with more fluctuations. In **Figure 5**, energy distributions are more localized. On the frequency range $0 \leq f \leq 1 \times 10^{-6}$ Hz, we can observe the influence of small-scale event on energy behavior. As noticed, this energy may be weak or null. As an example, the impact of a general strike in early 2009 that paralyzed Guadeloupean archipelago at least 2 months is highlighted by zero energy due to the lack of PM10 sources, i.e., industrial activity and road traffic. For $f > 1 \times 10^{-6}$ Hz, one can see more precisely energy variation related to dust events from mesoscale to large scale. Contrary to SPEC, HS clearly illustrates localized energy fluctuations due to small-scale event. In fact, the STFT makes an assumption that any signal as piecewise stationary and uses suitable window function to produce the short-time spectral characteristics of the signal. However, in

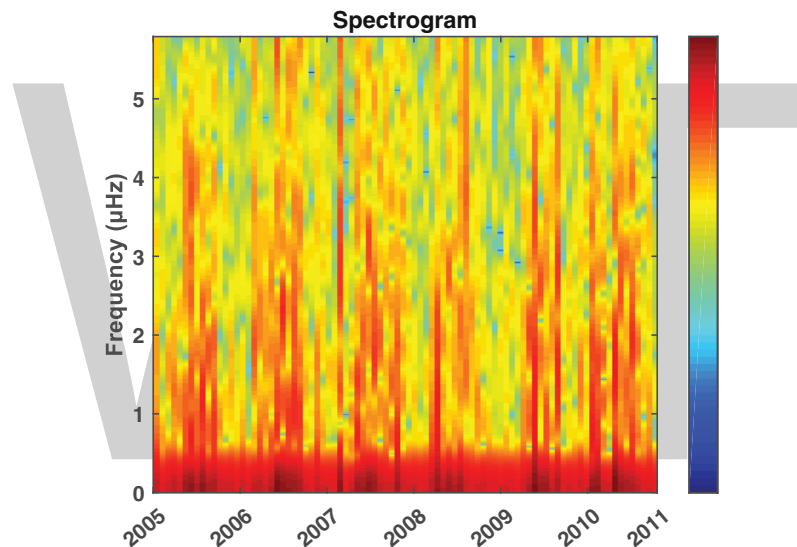


Figure 4.
Spectrogram of PM10 time series with a color gradient from strong energy (in red) to weak energy (in blue).

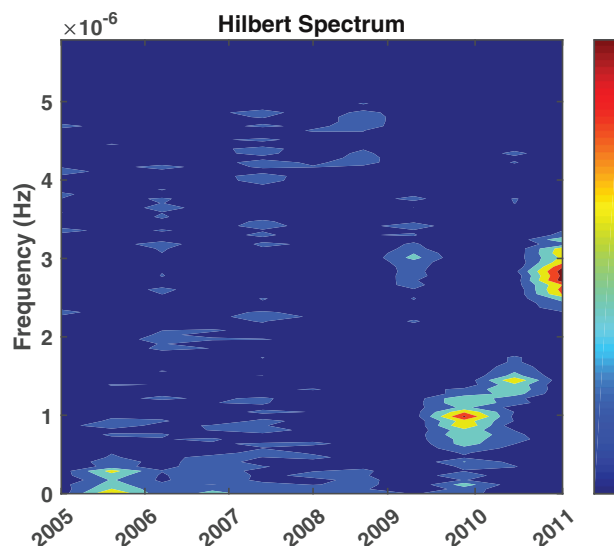


Figure 5.
Hilbert spectrum of PM10 time series with a color gradient from strong energy (in red) to weak energy (in blue).

reality, most of air pollution signals are usually nonstationary [9, 14, 47]. The Fourier transform-based technique treats the signal as a sum of predefined basis functions. If the analyzing signal is well matched with the bases, it performs better; otherwise the performance is degraded [10]. Here, the SPEC highlight energy fluctuations linked to PM10 coming from African dust between May and September (large-scale sources) [9] and from the eruption of Soufrière on Montserrat in February 2010 (mesoscale sources) [48]. However, the SPEC does not detect energy fluctuation related to anthropogenic pollution, i.e., local sources. This shows HS is a robust method in time-frequency domain. Indeed, based on the EMD method, this TFR is fully data adaptive, and the signal decomposition is performed without any predefined basis functions. These results confirm the superiority of HS over STFT in TFR.

5. Conclusion

In this paper, we investigated scaling and time-frequency properties of PM10 data in Hilbert frame. The performances obtained in the Hilbert space are compared with those achieved in the Fourier space. Firstly, with the Hilbert spectral analysis (HSA), a power law behavior is clearly observed on the frequency range $2.09 \times 10^{-7} \leq f \leq 4.57 \times 10^{-5}$ Hz which corresponds to time scales $6.1 \text{ hours} \leq T \leq 55.4 \text{ days}$ with an estimated spectral exponent $\beta_h = 1.02 \pm 0.10$. As HSA methodology has a very local ability in both physical and spectral spaces, the influence of intermittent dust events with huge fluctuations is included in the amplitude-frequency space which is not the case in Fourier spectrum. Thereafter, PM10 data are illustrated in time-frequency representations with the Hilbert spectrum and spectrogram. The results provide the evidence that HS-based TFR performs better than SPEC. The higher resolution in TFR offers better fluctuations of PM10 energy for $f < 1 \mu \text{ Hz}$. This is due to the fact that it is impossible to increase the TF resolution at the desired level in SPEC. The major asset of HS is that the time resolution can be as precise as the sampling period and the frequency resolution depends on the choice up to the Nyquist limit. In addition, contrary to SPEC which introduces a noticeable amount of cross-spectral energy terms during the use of window function with overlapping, HS is fully adaptive to datasets due to the decomposition of the signals. These first results suggest a substantial possibility to perform a profound dynamical analysis of PM10 concentrations for the Caribbean area in order to quantify the origin and the threshold pollution.

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Conflict of interest

The authors declare no conflict of interest.

Abbreviations

PM10	particulate matter with an aerodynamic diameter 10 μm or less
PSD	power spectral density
SPEC	spectrogram

EMD	empirical mode decomposition
IMF	intrinsic mode function
HSA	Hilbert spectral analysis
HHT	Hilbert-Huang transform
TFR	time-frequency representation
HS	Hilbert spectrum
STFT	short-time Fourier transform
WT	wavelet transform

Nomenclature

$E(f)$	Fourier spectral density
f	frequency (Hz)
β	spectral exponent
\mathcal{A}	instantaneous amplitude
$C(t)$	intrinsic mode function component
$h(\omega)$	Hilbert spectrum
ω	instantaneous frequency (Hz)
j	scale index
N	total length of a sequence
$x(t)$	particulate matter signal ($\mu\text{g}/\text{m}^3$)
$r(t)$	residual of the intrinsic mode function
φ	phase function of the intrinsic mode function

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WWT

Integral Inequalities and Differential Equations via Fractional Calculus

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Abstract

In this chapter, fractional calculus is used to develop some results on integral inequalities and differential equations. We develop some results related to the Hermite-Hadamard inequality. Then, we establish other integral results related to the Minkowski inequality. We continue to present our results by establishing new classes of fractional integral inequalities using a family of positive functions; these classes of inequalities can be considered as generalizations of order n for some other classical/fractional integral results published recently. As applications on inequalities, we generate new lower bounds estimating the fractional expectations and variances for the beta random variable. Some classical covariance identities, which correspond to the classical case, are generalised for any $\alpha \geq 1, \beta \geq 1$. For the part of differential equations, we present a contribution that allow us to develop a class of fractional chaotic electrical circuit. We prove recent results for the existence and uniqueness of solutions for a class of Langevin-type equation. Then, by establishing some sufficient conditions, another result for the existence of at least one solution is also discussed.

Keywords: fractional calculus, fixed point, Riemann-Liouville integral, Caputo derivative, integral inequality

1. Introduction

During the last few decades, fractional calculus has been extensively developed due to its important applications in many field of research [1–4]. On the other hand, the integral inequalities are very important in probability theory and in applied sciences. For more details, we refer the reader to [5–12] and the references therein. Moreover, the study of integral inequalities using fractional integration theory is also of great importance; we refer to [1, 13–17] for some applications.

Also, boundary value problems of fractional differential equations have occupied an important area in the fractional calculus domain, since these problems appear in several applications of sciences and engineering, like mechanics, chemistry, electricity, chemistry, biology, finance, and control theory. For more details, we refer the reader to [3, 18–20].

In this chapter, we use the Riemann-Liouville integrals to present some results related to Minkowski and Hermite-Hadamard inequalities [21]. We continue to present our results by establishing several classes of fractional integral inequalities

using a family of positive functions; these classes of inequalities can be considered as generalizations for some other fractional and classical integral results published recently [22]. Then, as applications, we generate new lower bounds estimating the fractional expectations and variances for the beta random variable. Some classical covariance identities, which correspond to $\alpha = 1$, are generalized for any $\alpha \geq 1$ and $\beta \geq 1$; see [23].

For the part of differential equations, with my coauthor, we present a contribution that allows us to develop a class of fractional differential equations generalizing the chaotic electrical circuit model. We prove recent results for the existence and uniqueness of solutions for a class of Langevin-type equations. Then, by establishing some sufficient conditions on the data of the problem, another result for the existence of at least one solution is also discussed. The considered class has some relationship with the good paper in [20].

The chapter is structured as follows: In Section 2, we recall some preliminaries on fractional calculus that will be used in the chapter. Section 3 is devoted to the main results on integral inequalities as well as to some estimates on continuous random variables. The Section 4 deals with the class of differential equations of Langevin type: we study the existence and uniqueness of solutions for the considered class by means of Banach contraction principle, and then using Schaefer fixed point theorem, an existence result is discussed. At the end, the Conclusion follows.

2. Preliminaries on fractional calculus

In this section, we present some definitions and lemmas that will be used in this chapter. For more details, we refer the reader to [2, 13, 15, 24].

Definition 1.1. The Riemann-Liouville fractional integral operator of order $\alpha \geq 0$, for a continuous function f on $[a, b]$ is defined as

$$J_a^\alpha[f(t)] = \frac{1}{\Gamma(\alpha)} \int_a^t (t - \tau)^{\alpha-1} f(\tau) d\tau, \alpha > 0, a < t \leq b, \quad (1)$$

$$J_a^0[f(t)] = f(t),$$

where $\Gamma(\alpha) := \int_0^\infty e^{-u} u^{\alpha-1} du$.

Note that for $\alpha > 0, \beta > 0$, we have

$$J^\alpha J^\beta f(t) = J^{\alpha+\beta} f(t), \quad (2)$$

and

$$J_a^\alpha J_a^\beta[f(t)] = J_a^\beta J_a^\alpha[f(t)]. \quad (3)$$

In the rest of this chapter, for short, we note a probability density function by $p.d.f$. So, let us consider a positive continuous function ω defined on $[a, b]$. We recall the ω -concepts:

Definition 1.2. The fractional ω -weighted expectation of order $\alpha > 0$, for a random variable X with a positive $p.d.f.f$ defined on $[a, b]$, is given by

$$E_{\alpha,\omega}(X) := J_a^\alpha[t\omega f](b) = \frac{1}{\Gamma(\alpha)} \int_a^b (b - \tau)^{\alpha-1} \tau \omega(\tau) f(\tau) d\tau, \alpha > 0, a < t \leq b, \quad (4)$$

Definition 1.3. The fractional ω -weighted variance of order $\alpha > 0$ for a random variable X having a $p.d.f.f$ on $[a, b]$ is given by

$$\sigma_{\alpha,\omega}^2(X) = V_{\alpha,\omega}(X) := \frac{1}{\Gamma(\alpha)} \int_a^b (b - \tau)^{\alpha-1} (\tau - E(X))^2 \omega(\tau) f(\tau) d\tau, \alpha > 0. \quad (5)$$

Definition 1.4. The fractional ω -weighted moment of orders $r > 0, \alpha > 0$ for a continuous random variable X having a $p.d.f.f$ defined on $[a, b]$ is defined by the quantity:

$$E_{\alpha,\omega}(X^r) := \frac{1}{\Gamma(\alpha)} \int_a^b (b - \tau)^{\alpha-1} \tau^r \omega(\tau) f(\tau) d\tau, \alpha > 0. \quad (6)$$

We introduce the covariance of fractional order as follows.

Definition 1.5. Let f_1 and f_2 be two continuous on $[a, b]$. We define the fractional ω -weighted covariance of order $\alpha > 0$ for $(f_1(X), f_2(X))$ by

$$Cov_{\alpha,\omega}(f_1(X), f_2(X)) := \frac{1}{\Gamma(\alpha)} \int_a^b (b - \tau)^{\alpha-1} (f_1(\tau) - f_1(\mu)) (f_2(\tau) - f_2(\mu)) \omega(\tau) f(\tau) d\tau, \alpha > 0, \quad (7)$$

where μ is the classical expectation of X .

It is to note that when $\omega(x) = 1, x \in [a, b]$, then we put

$$Var_{\alpha,\omega}(X) := Var_{\alpha}(X), Cov_{\alpha,\omega}(X) := Cov_{\alpha}(X), E_{\alpha,\omega}(X) := E_{\alpha}(X)$$

Definition 1.6. For a function $K \in C^n([a, b], \mathbb{R})$ and $n - 1 < \alpha \leq n$, the Caputo fractional derivative of order α is defined by

$$\begin{aligned} D^{\alpha}K(t) &= J^{n-\alpha} \frac{d^n}{dt^n} (K(t)) \\ &= \frac{1}{\Gamma(n - \alpha)} \int_a^t (t - s)^{n-\alpha-1} K^{(n)}(s) ds. \end{aligned}$$

We recall also the following properties.

Lemma 1.7. Let $n \in \mathbb{N}^*$, and $n - 1 < \alpha < n$. The general solution of $D^{\alpha}y(t) = 0, t \in [a, b]$ is given by

$$y(t) = \sum_{i=0}^{n-1} c_i (t - a)^i, \quad (8)$$

where $c_i \in \mathbb{R}, i = 0, 1, 2, \dots, n - 1$.

Lemma 1.8. Let $n \in \mathbb{N}^*$ and $n - 1 < \alpha < n$. Then

$$J^{\alpha} D^{\alpha}y(t) = y(t) + \sum_{i=0}^{n-1} c_i (t - a)^i, t \in [a, b], \quad (9)$$

for some $c_i \in \mathbb{R}, i = 0, 1, 2, \dots, n - 1$.

3. Some integral inequalities

3.1 On Minkowski and Hermite-Hadamard fractional inequalities

In this subsection, we present some fractional integral results related to Minkowski and Hermite-Hadamard integral inequalities. For more details, we refer the reader to [21].

Theorem 1.9. Let $\alpha > 0, p \geq 1$ and let f, g be two positive functions on $[0, \infty[$, such that for all $t > 0, J^\alpha f^p(t) < \infty, J^\alpha g^p(t) < \infty$. If $0 < m \leq \frac{f(\tau)}{g(\tau)} \leq M, \tau \in [0, t]$, then we have

$$[J^\alpha f^p(t)]^{\frac{1}{p}} + [J^\alpha g^p(t)]^{\frac{1}{p}} \leq \frac{1 + M(m + 2)}{(m + 1)(M + 1)} [J^\alpha (f + g)^p(t)]^{\frac{1}{p}}. \quad (10)$$

Proof: We use the hypothesis $\frac{f(\tau)}{g(\tau)} < M, \tau \in [0, t], t > 0$. We can write

$$\frac{(M + 1)^p}{\Gamma(\alpha)} \int_0^t (t - \tau)^{\alpha-1} f^p(\tau) d\tau \quad (11)$$

$$\leq \frac{M^p}{\Gamma(\alpha)} \int_0^t (t - \tau)^{\alpha-1} (f + g)^p(\tau) d\tau.$$

Hence, we have

$$J^\alpha f^p(t) \leq \frac{M^p}{(M + 1)^p} J^\alpha (f + g)^p(t). \quad (12)$$

Thus, it yields that

$$[J^\alpha f^p(t)]^{\frac{1}{p}} \leq \frac{M}{M + 1} [J^\alpha (f + g)^p(t)]^{\frac{1}{p}}. \quad (13)$$

In the same manner, we have

$$\left(1 + \frac{1}{m}\right) g(\tau) \leq \frac{1}{m} (f(\tau) + g(\tau)). \quad (14)$$

And then,

$$[J^\alpha g^p(t)]^{\frac{1}{p}} \leq \frac{1}{m + 1} [J^\alpha (f + g)^p(t)]^{\frac{1}{p}}. \quad (15)$$

Combining (13) and (15), we achieve the proof.

Remark 1.10. Applying the above theorem for $\alpha = 1$, we obtain Theorem 1.2 of [25] on $[0, t]$.

With the same arguments as before, we present the following theorem.

Theorem 1.11. Let $\alpha > 0, p \geq 1$ and let f, g be two positive functions on $[0, \infty[$, such that for all $t > 0, J^\alpha f^p(t) < \infty, J^\alpha g^p(t) < \infty$. If $0 < m \leq \frac{f(\tau)}{g(\tau)} \leq M, \tau \in [0, t]$, then we have

$$[J^\alpha f^p(t)]^{\frac{1}{p}} + [J^\alpha g^p(t)]^{\frac{1}{p}} \leq \frac{1 + M(m + 2)}{(m + 1)(M + 1)} [J^\alpha (f + g)^p(t)]^{\frac{1}{p}}. \tag{16}$$

Remark 1.12. Taking $\alpha = 1$ in this second theorem, we obtain Theorem 2.2 in [26] on $[0, t]$.

Using the notions of concave and L^p -functions, we present to the reader the following result.

Theorem 1.13. Suppose that $\alpha > 0, p > 1, q > 1$ and let f, g be two positive functions on $[0, \infty[$. If f^p, g^q are two concave functions on $[0, \infty[$, then we have

$$\begin{aligned} 2^{-p-q} (f(0) + f(t))^p (g(0) + g(t))^q (J^\alpha (t^{\alpha-1}))^2 \\ \leq J^\alpha (t^{\alpha-1} f^p(t)) J^\alpha (t^{\alpha-1} g^q(t)). \end{aligned} \tag{17}$$

The proof of this theorem is based on the following auxiliary result.

Lemma 1.14. Let h be a concave function on $[a, b]$. Then for any $x \in [a, b]$, we have

$$h(a) + h(b) \leq h(b + a - x) + h(x) \leq 2h\left(\frac{a + b}{2}\right). \tag{18}$$

3.2 A family of fractional integral inequalities

We present to the reader some integral results for a family of functions [22]. These results generalize some integral inequalities of [27]. We have

Theorem 1.15. Suppose that $(f_i)_{i=1, \dots, n}$ are n positive, continuous, and decreasing functions on $[a, b]$. Then, the following inequality

$$\frac{J^\alpha \left[\prod_{i \neq p}^n f_i^{\gamma_i} f_p^\beta(t) \right]}{J^\alpha \left[\prod_{i=1}^n f_i^{\gamma_i}(t) \right]} \geq \frac{J^\alpha \left[(t - a)^\delta \prod_{i \neq p}^n f_i^{\gamma_i} f_p^\beta(t) \right]}{J^\alpha \left[(t - a)^\delta \prod_{i=1}^n f_i^{\gamma_i}(t) \right]} \tag{19}$$

holds for any $a < t \leq b, \alpha > 0, \delta > 0, \beta \geq \gamma_p > 0$, where p is a fixed integer in $\{1, 2, \dots, n\}$.

Proof: It is clear that

$$\left((\rho - a)^\delta - (\tau - a)^\delta \right) \left(f_p^{\beta - \gamma_p}(\tau) - f_p^{\beta - \gamma_p}(\rho) \right) \geq 0, \tag{20}$$

for any fixed $p \in \{1, \dots, n\}$ and for any $\beta \geq \gamma_p > 0, \delta > 0, \tau, \rho \in [a, t]; a < t \leq b$.

Taking

$$K_p(\tau, \rho) := \frac{(t - \tau)^{\alpha-1}}{\Gamma(\alpha)} \prod_{i=1}^n f_i^{\gamma_i}(\tau) \left((\rho - a)^\delta - (\tau - a)^\delta \right) \left(f_p^{\beta - \gamma_p}(\tau) - f_p^{\beta - \gamma_p}(\rho) \right), \tag{21}$$

we observe that

$$K_p(\tau, \rho) \geq 0. \tag{22}$$

Also, we have

$$0 \leq \int_a^t K_p(\tau, \rho) d\tau = (\rho - a)^\delta J^\alpha \left[\prod_{i \neq p}^n f_i^{\gamma_i} f_p^\beta(t) \right] + f_p^{\beta - \gamma_p}(\rho) J^\alpha \left[(t - a)^\delta \prod_{i=1}^n f_i^{\gamma_i}(t) \right] - J^\alpha \left[(t - a)^\delta \prod_{i \neq p}^n f_i^{\gamma_i} f_p^\beta(t) \right] - (\rho - a)^\delta f_p^{\beta - \gamma_p}(\rho) J^\alpha \left[\prod_{i=1}^n f_i^{\gamma_i}(t) \right]. \tag{23}$$

Hence, we get

$$J^\alpha \left[(t - a)^\delta \prod_{i=1}^n f_i^{\gamma_i}(t) \right] J^\alpha \left[\prod_{i \neq p}^n f_i^{\gamma_i} f_p^\beta(t) \right] \geq J^\alpha \left[\prod_{i=1}^n f_i^{\gamma_i}(t) \right] J^\alpha \left[(t - a)^\delta \prod_{i \neq p}^n f_i^{\gamma_i} f_p^\beta(t) \right]. \tag{24}$$

The proof is thus achieved.

Remark 1.16. Applying Theorem 1.15 for $\alpha = 1, t = b, n = 1$, we obtain Theorem 3 in [27].

Using other sufficient conditions, we prove the following generalization.

Theorem 1.17. Suppose that $(f_i)_{i=1, \dots, n}$ are positive, continuous, and decreasing functions on $[a, b]$. Then for any fixed p in $\{1, 2, \dots, n\}$ and for any $a < t \leq b, \alpha > 0, \omega > 0, \delta > 0, \beta \geq \gamma_p > 0$, we have

$$\frac{J^\alpha \left[\prod_{i \neq p}^n f_i^{\gamma_i} f_p^\beta(t) \right] J^\omega \left[(t - a)^\delta \prod_{i=1}^n f_i^{\gamma_i}(t) \right] + J^\omega \left[\prod_{i \neq p}^n f_i^{\gamma_i} f_p^\beta(t) \right] J^\alpha \left[(t - a)^\delta \prod_{i=1}^n f_i^{\gamma_i}(t) \right]}{J^\alpha \left[(t - a)^\delta \prod_{i \neq p}^n f_i^{\gamma_i} f_p^\beta(t) \right] J^\omega \left[\prod_{i=1}^n f_i^{\gamma_i}(t) \right] + J^\omega \left[(t - a)^\delta \prod_{i \neq p}^n f_i^{\gamma_i} f_p^\beta(t) \right] J^\alpha \left[\prod_{i=1}^n f_i^{\gamma_i}(t) \right]} \geq 1. \tag{25}$$

Proof: Multiplying both sides of (23) by $\frac{(t-\rho)^{\omega-1}}{\Gamma(\omega)} \prod_{i=1}^n f_i^{\gamma_i}(\rho), \omega > 0$, then integrating the resulting inequality with respect to ρ over $(a, t), a < t \leq b$ and using Fubini's theorem, we obtain the desired inequality.

Remark 1.18.

- i. Applying Theorem 1.17 for $\alpha = \omega$, we obtain Theorem 1.15.
- ii. Applying Theorem 1.17 for $\alpha = \omega = 1, t = b, n = 1$, we obtain Theorem 3 of [27].

Introducing a positive increasing function g to the family $(f_i)_{i=1, \dots, n}$, we establish the following theorem.

Theorem 1.19. Let $(f_i)_{i=1, \dots, n}$ and g be positive continuous functions on $[a, b]$, such that g is increasing and $(f_i)_{i=1, \dots, n}$ are decreasing on $[a, b]$. Then, the following inequality

$$\frac{J^\alpha \left[\prod_{i \neq p}^n f_i^{\gamma_i} f_p^\beta(t) \right] J^\alpha \left[g^\delta(t) \prod_{i=1}^n f_i^{\gamma_i}(t) \right]}{J^\alpha \left[g^\delta(t) \prod_{i \neq p}^n f_i^{\gamma_i} f_p^\beta(t) \right] J^\alpha \left[\prod_{i=1}^n f_i^{\gamma_i}(t) \right]} \geq 1 \tag{26}$$

holds for any $a < t \leq b, \alpha > 0, \delta > 0, \beta \geq \gamma_p > 0$, where p is a fixed integer in $\{1, 2, \dots, n\}$.

Remark 1.20. Applying Theorem 1.19 for $\alpha = 1, t = b, n = 1$, we obtain Theorem 4 of [27].

3.3 Some estimations on random variables

3.3.1 Bounds for fractional moments of beta distribution

In what follows, we present some fractional results on the beta distribution [23]. So let us prove the following α -version.

Theorem 1.21. Let X, Y, U , and V be four random variables, such that $X \sim B(p, q), Y \sim B(m, n), U \sim B(p, n)$, and $V \sim B(m, q)$. If $(p - m)(q - n) \leq 0$, then

$$\frac{E_\alpha(X^r)E_\alpha(Y^r)}{E_\alpha(U^r)E_\alpha(V^r)} \geq \frac{B(p, n)B(m, q)}{B(p, q)B(m, n)}, \alpha \geq 1.$$

For the proof of this result, we can apply a weighted version of the fractional Chebyshev inequality as is mentioned in [1].

Remark 1.22. The above theorem generalizes Theorem 3.1 of [7].

We propose also the following (α, β) -version that generalizes the above result. We have

Theorem 1.23. Let X, Y, U , and V be four random variables, such that $X \sim B(p, q), Y \sim B(m, n), U \sim B(p, n)$, and $V \sim B(m, q)$. If $(p - m)(q - n) \leq 0$, then

$$\frac{E_\alpha(X^r)E_\beta(Y^r) + E_\beta(X^r)E_\alpha(Y^r)}{E_\alpha(U^r)E_\beta(V^r) + E_\beta(U^r)E_\alpha(V^r)} \geq \frac{B(p, n)B(m, q)}{B(p, q)B(m, n)}, \alpha, \beta \geq 1.$$

Remark 1.24. If $\alpha = \beta = 1$, then the above theorem reduces to Theorem 3.1 of [7].

3.3.2 Identities and lower bounds

In the following theorem, the fractional covariance of X and $g(X)$ is expressed with the derivative of $g(X)$. It can be considered as a generalization of a covariance identity established by the authors of [28]. So, we prove the result:

Theorem 1.25. Let X be a random variable having a $p.d.f$ defined on $[a, b]$; $\mu = E(X)$. Then, we have

$$Cov_\alpha(X, g(X)) = \frac{1}{\Gamma(\alpha)} \int_a^b g'(x) dx \int_a^x (b - t)^{\alpha-1} (\mu - t) f(t) dt, \alpha \geq 1. \quad (27)$$

We can prove this result by the application of the covariance definition in the case where $\omega(x) = 1$.

The following theorem establishes a lower bound for $Var_\alpha(g(X))$ of any function $g \in C^1([a, b])$. We have

Theorem 1.26. Let X be a random variable having a $p.d.f$ defined on $[a, b]$, such that $\mu = E(X)$. Then, we have

$$Var_\alpha(g(X)) \geq \frac{1}{Var_{X,\alpha}} \left(\frac{1}{\Gamma(\alpha)} \int_a^b g'(x) dx \int_a^x (b - t)^{\alpha-1} (\mu - t) f(t) dt \right)^2, \quad (28)$$

for any $g \in C^1([a, b])$.

To prove this result, we use fractional Cauchy-Schwarz inequality established in [29].

Remark 1.27. Let us consider $\Omega \in C([a, b])$ that satisfies $\int_a^x (b-t)^{\alpha-1}(\mu-t)f(t)dt = (b-x)^{\alpha-1}\sigma^2\Omega(x)f(x)$. Then, we present the following result.

Theorem 1.28. Let X be a random variable having a *p.d.f.* defined on $[a, b]$, such that $\mu = E(X)$, $\sigma^2 = Var(X)$ and $\Omega \in C([a, b])$; $\int_a^x (b-t)^{\alpha-1}(\mu-t)f(t)dt = (b-x)^{\alpha-1}\sigma^2\Omega(x)f(x)$. Then, we have

$$Var_\alpha(g(X)) \geq \frac{\sigma^4(X)}{Var_\alpha(X)} E_\alpha^2(g'(X)\Omega(X)). \tag{29}$$

Proof: We have

$$Cov_\alpha^2(X, g(X)) = \left[\frac{1}{\Gamma(\alpha)} \int_a^b g'(x)dx (b-x)^{\alpha-1} \sigma^2 \Omega(x) f(x) dx \right]^2. \tag{30}$$

On the other hand, we can see that

$$\left[\frac{1}{\Gamma(\alpha)} \int_a^b g'(x)dx (b-x)^{\alpha-1} \sigma^2 \Omega(x) f(x) dx \right]^2 = \sigma^4 E_\alpha^2(g'(X)\Omega(X)) \tag{31}$$

Thanks to the fractional version of Cauchy Schwarz inequality [29], and using the fact that

$$Cov_\alpha^2(X, g(X)) \leq Var_\alpha(X) Var_\alpha(g(X)), \tag{32}$$

we obtain

$$\sigma^4 E_\alpha^2(g'(X)\Omega(X)) \leq Var_\alpha(X) Var_\alpha(g(X)). \tag{33}$$

This ends the proof.

Remark 1.29. Thanks to (30) and (31), we obtain the following fractional covariance identity

$$\sigma^2 E_\alpha(g'(X)\Omega(X)) = Cov_\alpha(X, g(X)).$$

It generalizes the good standard identity obtained in [28] that corresponds to $\alpha = 1$ and it is given by

$$\sigma^2 E(g'(X)\Omega(X)) = Cov(X, g(X)).$$

We end this section by proving the following fractional integral identity between covariance and expectation in the fractional case.

Theorem 1.30. Let X be a continuous random variable with a *p.d.f.* having a support an interval $[a, b]$, $E(X) = \mu$. Then, for any $\alpha \geq 1$, the following general covariance identity holds

$$Cov_\alpha(h(X), g(X)) = E_\alpha(g'(X)Z(X)), \tag{34}$$

where $g \in C^1([a, b])$, with $E|Z(X)g'(X)| < \infty$, $h(x)$ is a given function and $Z(x)f(x) \frac{(b-x)^{\alpha-1}}{\Gamma(\alpha)} = \int_a^x (E(h(X)) - h(t)) \frac{(b-t)^{\alpha-1}}{\Gamma(\alpha)} f(t) dt$.

Proof: We have

$$Cov_\alpha(h(X), g(X)) = \frac{1}{\Gamma(\alpha)} \int_a^b (b-x)^{\alpha-1} (h(x) - h(\mu))(g(x) - g(\mu)) f(x) dx \quad (35)$$

and

$$E_\alpha(g'(X)Z(X)) = \frac{1}{\Gamma(\alpha)} \int_a^b (b-x)^{\alpha-1} g'(x) Z(x) f(x) dx. \quad (36)$$

The definition of $Z(X)$ implies that

$$E_\alpha(g'(X)Z(X)) = \frac{1}{\Gamma(\alpha)} \int_a^\mu (g(\mu) - g(t))(b-t)^{\alpha-1} (h(\mu) - h(t)) f(t) dt \quad (37)$$

$$+ \frac{1}{\Gamma(\alpha)} \int_\mu^b (g(t) - g(\mu))(b-t)^{\alpha-1} (h(t) - h(\mu)) f(t) dt.$$

Hence, we obtain

$$E_\alpha(Z(X)g'(X)) = Cov_\alpha(g(X), h(X)). \quad (38)$$

Remark 1.31. Taking $\alpha = 1$, in the above theorem, we obtain Theorem 2.2 of [10].

4. A class of differential equations of fractional order

Inspired by the work in [4, 20], in what follows we will be concerned with a more general class of Langevin equations of fractional order. The considered class will contain a nonlinearity that depends on a fractional derivative of order δ . So, let us consider the following problem:

$${}^c D^\alpha (D^2 + \lambda^2) u(t) = f(t, u(t), {}^c D^\delta u(t)),$$

$$t \in [0, 1], \quad \lambda \in \mathbb{R}_+^* \quad (39)$$

$$0 < \alpha \leq 1, \quad 0 \leq \delta < \alpha,$$

associated with the conditions

$$u(0) = 0, \quad u''(0) = 0, \quad u(1) = \beta u(\eta), \eta \in (0, 1), \quad (40)$$

where ${}^c D^\alpha$ denotes the Caputo fractional derivative of fractional order α , D^2 is the two-order classical derivative, $f: [0, 1] \times \mathbb{R} \times \mathbb{R} \rightarrow \mathbb{R}$ is a given function, and $\beta \in \mathbb{R}$, such that $\beta \sin(\lambda \eta) \neq \sin(\lambda)$.

4.1 Integral representation

We recall the following result [20]:

Lemma 1.32. Let θ be a continuous function on $[0, 1]$. The unique solution of the problem

$$\begin{aligned} {}^c D^\alpha (D^2 + \lambda^2) u(t) &= \theta(t), \\ t \in [0, 1], \quad \lambda \in \mathbb{R}_+^* & \\ n - 1 < \alpha \leq n, \quad n \in \mathbb{N}^*, & \end{aligned} \tag{41}$$

is given by

$$u(t) = \frac{1}{\lambda} \int_0^t \sin \lambda(t-s) \left(\int_0^s \frac{(s-\tau)^{\alpha-1}}{\Gamma(\alpha)} \theta(\tau) d\tau + \sum_{i=1}^{n-1} c_i s^i \right) ds + c_n \cos(\lambda t) + c_{n+1} \sin(\lambda t), \tag{42}$$

where $c_i \in \mathbb{R}, i = 1 \dots n + 1$.

Thanks to the above lemma, we can state that

The class of Langevin equations (39) and (40) has the following integral representation:

$$\begin{aligned} u(t) &= \frac{1}{\lambda} \int_0^t \sin \lambda(t-s) \left(\int_0^s \frac{(s-\tau)^{\alpha-1}}{\Gamma(\alpha)} f(\tau, u(\tau), D^\delta(\tau)) d\tau \right) ds \\ &+ \frac{\sin(\lambda t)}{\Delta} \left[\beta \int_0^\eta \sin \lambda(\eta-s) \left(\int_0^s \frac{(s-\tau)^{\alpha-1}}{\Gamma(\alpha)} f(\tau, u(\tau), D^\delta(\tau)) d\tau \right) ds \right. \\ &\left. - \int_0^1 \sin \lambda(1-s) \left(\int_0^s \frac{(s-\tau)^{\alpha-1}}{\Gamma(\alpha)} f(\tau, u(\tau), D^\delta(\tau)) d\tau \right) ds \right], \end{aligned} \tag{43}$$

where

$$\Delta := \lambda(\sin \lambda - \beta \sin \lambda \eta). \tag{44}$$

4.2 Existence and uniqueness of solutions

Using the above integral representation (43), we can prove the following existence and uniqueness theorem.

Theorem 1.33. Assume that the following hypotheses are valid:

(H1): The function $f : [0, 1] \times \mathbb{R} \times \mathbb{R} \rightarrow \mathbb{R}$ is continuous, and there exist two constants $\Lambda_1, \Lambda_2 > 0$, such that for all $t \in [0, 1]$ and $u_i, v_i \in \mathbb{R}, i = 1, 2$,

$$|f(t, u_1, u_2) - f(t, v_1, v_2)| \leq \Lambda_1 |u_1 - v_1| + \Lambda_2 |u_2 - v_2|. \tag{45}$$

(H2): Suppose that $\Lambda \leq \frac{1}{(\Phi + \Upsilon)}$,

where

$$\begin{aligned} \Phi &:= \frac{\Delta_1 + \lambda + \beta_1 \lambda \eta^{\alpha+1}}{\Gamma(\alpha + 2) \lambda \Delta_1}, \quad \Psi := \frac{\Delta_1 s^\alpha (\alpha + 1) + \lambda^2 + \beta_1 \lambda^2 \eta^{\alpha+1}}{\Gamma(\alpha + 2) \lambda \Delta_1}, \quad \Upsilon := \frac{\Psi}{\Gamma(2 - \delta)}, \\ \Lambda &:= \max(\Lambda_1, \Lambda_2), \quad \Delta_1 = |\Delta|, \quad \beta_1 = |\beta|. \end{aligned}$$

Then problem (39) and (40) has a unique solution on $[0, 1]$.

Proof: We introduce the space

$$E = \{u; u \in \mathcal{C}([0, 1]), D^\delta u \in \mathcal{C}([0, 1])\},$$

endowed with the norm $\|u\|_E := \|u\|_\infty + \|D^\delta u\|_\infty$.

Then, $(E, \|\cdot\|_E)$ is a Banach space.

Also, we consider the operator $T : E \rightarrow E$ defined by

$$\begin{aligned} (Tu)(t) := & \frac{1}{\lambda} \int_0^t \sin \lambda(t-s) J_0^\alpha f(s, u(s), D^\delta u(s)) ds \\ & + \frac{\sin(\lambda t)}{\Delta} \left[\beta \int_0^\eta \sin \lambda(\eta-s) J_0^\alpha f(s, u(s), D^\delta u(s)) ds \right. \\ & \left. - \int_0^1 \sin \lambda(1-s) J_0^\alpha f(s, u(s), D^\delta u(s)) ds \right] \end{aligned} \quad (46)$$

We shall prove that the above operator is contractive over the space E .

Let $u_1, u_2 \in E$. Then, for each $t \in [0, 1]$, we have

$$\begin{aligned} |Tu_1(t) - Tu_2(t)| \leq & \frac{1}{\lambda} \int_0^t |\sin \lambda(t-s)| |J_0^\alpha| |f(s, u_1(s), D^\delta u_1(s)) - f(s, u_2(s), D^\delta u_2(s))| ds \\ & + \frac{|\sin(\lambda t)|}{|\Delta|} \left[|\beta| \int_0^\eta |\sin \lambda(\eta-s)| |J_0^\alpha| |f(s, u_1(s), D^\delta u_1(s)) - f(s, u_2(s), D^\delta u_2(s))| ds \right. \\ & \left. + \int_0^1 |\sin \lambda(1-s)| |J_0^\alpha| |f(s, u_1(s), D^\delta u_1(s)) - f(s, u_2(s), D^\delta u_2(s))| ds \right] := \mathcal{A} \end{aligned}$$

By (H1), we have

$$\mathcal{A} \leq \frac{\Lambda}{\Gamma(\alpha+2)} \left(\frac{1}{\lambda} + \frac{1}{|\Delta|} + \frac{|\beta|}{|\Delta|} \eta^{\alpha+1} \right) (|u_1 - u_2| + |D^\delta u_1 - D^\delta u_2|).$$

Hence, it yields that

$$\|Tu_1 - Tu_2\|_\infty \leq \Lambda \Phi \|u_1 - u_2\|_E. \quad (47)$$

With the same arguments as before, we can write.

$$\begin{aligned} |T'u_1(t) - T'u_2(t)| \leq & \frac{1}{\lambda} |\sin \lambda(t-s)| |J_0^\alpha| |f(s, u_1(s), D^\delta u_1(s)) - f(s, u_2(s), D^\delta u_2(s))| ds \\ & + \frac{|\lambda \cos(\lambda t)|}{|\Delta|} \left[\beta \int_0^\eta |\sin \lambda(\eta-s)| |J_0^\alpha| |f(s, u_1(s), D^\delta u_1(s)) - f(s, u_2(s), D^\delta u_2(s))| ds \right. \\ & \left. + \int_0^1 |\sin \lambda(1-s)| |J_0^\alpha| |f(s, u_1(s), D^\delta u_1(s)) - f(s, u_2(s), D^\delta u_2(s))| ds \right] := \mathcal{B}. \end{aligned}$$

Again, by (H1), we obtain

$$\mathcal{B} \leq \Lambda \left(\frac{\Delta_1 s^\alpha (\alpha + 1) + \lambda^2 + \beta_1 \lambda^2 \eta^{\alpha+1}}{\Gamma(\alpha + 2) \lambda \Delta_1} \right) (|u_1 - u_2| + |D^\delta u_1 - D^\delta u_2|).$$

Consequently, we get

$$\|T'u_1 - T'u_2\|_\infty \leq \Lambda \Psi \|u_1 - u_2\|_E.$$

This implies that

$$\|D^\delta T u_1 - D^\delta T u_2\|_\infty \leq \Lambda \Upsilon \|u_1 - u_2\|_E. \quad (48)$$

Using (47) and (48), we can state that

$$\|T u_1 - T u_2\|_E \leq \Lambda (\Phi + \Upsilon) \|u_1 - u_2\|_E.$$

Thanks to (H2), we can say that the operator T is contractive.

Hence, by Banach fixed point theorem, the operator has a unique fixed point which corresponds to the unique solution of our Langevin problem.

4.3 Existence of solutions

We prove the following theorem.

Theorem 1.34. Assume that the following conditions are satisfied:

(H3): The function $f : [0, 1] \times \mathbb{R} \times \mathbb{R} \rightarrow \mathbb{R}$ is jointly continuous.

(H4): There exists a positive constant \mathcal{M} ; $|f(t, u, v)| \leq \mathcal{M}$ for any $t \in [0, 1], u, v \in \mathbb{R}$.

Then the problem (39), (40) has at least one solution on $[0, 1]$.

Proof: We use Schaefer fixed point theorem to prove this result. So we proceed into three steps.

Step 1: We prove that T is continuous and bounded.

Since the function f is continuous by (H3), then the operator is also continuous; this proof is trivial and hence it is omitted.

Let $\Omega \subset E$ be a bounded set. We need to prove that $T(\Omega)$ is a bounded set.

Let $u \in \Omega$. Then, for any $t \in [0, 1]$, we have

$$|(Tu)(t)| \leq \left(\frac{1}{\lambda} + \frac{1}{|\Delta|} \right) \int_0^1 J_0^\alpha |f(s, u(s), D^\delta u(s))| ds + \frac{|\beta|}{|\Delta|} \int_0^\eta J_0^\alpha |f(s, u(s), D^\delta(s))| ds := \mathcal{C}$$

Using (H4), we get

$$\|Tu\|_\infty \leq \Phi \mathcal{M}. \quad (49)$$

In the same manner, we find that

$$\|D^\delta Tu\|_\infty \leq \Upsilon \mathcal{M}. \quad (50)$$

From (49) and (50), we have

$$\|Tu\|_E \leq (\Phi + \Upsilon) \mathcal{M}.$$

The operator is thus bounded.

Step 2: Equicontinuity.

Let $u \in E$. Then, for each $t_1, t_2 \in [0, 1]$, we have

$$\begin{aligned}
 |Tu(t_2) - Tu(t_1)| &\leq \left| \frac{1}{\lambda} \left[\int_0^{t_2} \sin\lambda(t_2 - s) J_0^\alpha f(s, u(s), D^\delta u(s)) ds - \int_0^{t_1} \sin\lambda(t_1 - s) J_0^\alpha f(s, u(s), D^\delta u(s)) ds \right] \right. \\
 &\quad + \frac{\sin(\lambda t_2) - \sin(\lambda t_1)}{\Delta} \left[\beta \int_0^\eta \sin\lambda(\eta - s) J_0^\alpha f(s, u(s), D^\delta u(s)) ds \right. \\
 &\quad \left. \left. + \int_0^1 \sin\lambda(1 - s) J_0^\alpha f(s, u(s), D^\delta u(s)) ds \right] \right| \\
 &\leq \frac{|\Theta|}{|\Delta|} |\sin(\lambda t_2) - \sin(\lambda t_1)| + \frac{1}{\lambda} \int_{t_1}^{t_2} \sin\lambda(t_2 - s) J_0^\alpha |f(s, u(s), D^\delta u(s))| ds \\
 &\quad + \frac{1}{\lambda} \int_0^{t_2} |(\sin\lambda(t_2 - s) - \sin\lambda(t_1 - s)) J_0^\alpha f(s, u(s), D^\delta u(s))| ds,
 \end{aligned} \tag{51}$$

where

$$\Theta := \beta \int_0^\eta |\sin\lambda(\eta - s) J_0^\alpha |f(s, u(s), D^\delta u(s))| ds + \int_0^1 |\sin\lambda(1 - s) J_0^\alpha |f(s, u(s), D^\delta u(s))| ds.$$

Analogously, we can obtain

$$|T'u(t_2) - T'u(t_1)| \leq \lambda \frac{|\Theta|}{|\Delta|} |\cos(\lambda t_2) - \cos(\lambda t_1)| + \frac{1}{\lambda} |\sin\lambda(t_2 - s) - \sin\lambda(t_1 - s)| J_0^\alpha |f(s, u(s), D^\delta u(s))|.$$

Consequently, we can write

$$|D^\delta Tu(t_2) - D^\delta Tu(t_1)| \leq J^{1-\delta} |T'u(t_2) - T'u(t_1)| \tag{52}$$

As $t_1 \rightarrow t_2$, the right-hand sides of (51) and (52) tend to zero. Therefore,

$$\|Tu(t_2) - Tu(t_1)\|_E \rightarrow 0.$$

The operator T is thus equicontinuous.

As a consequence of Step 1 and Step 2 and thanks to Arzela-Ascoli theorem, we conclude that T is completely continuous.

Step 3: We prove that $\Sigma := \{u \in E; u = \lambda Tu, 0 < \lambda < 1\}$ is a bounded set.

Let $u \in \Sigma$. Then, for each $t \in [0, 1]$, the following two inequalities are valid:

$$|u(t)| = |\lambda Tu(t)| \leq |Tu(t)| \leq \mathcal{M}\Phi$$

and

$$|D^\delta u(t)| = |\lambda D^\delta Tu(t)| \leq |D^\delta Tu(t)| \leq \mathcal{M}\Upsilon.$$

Therefore,

$$\|u\|_E \leq \mathcal{M}(\Upsilon + \Phi).$$

Thanks to steps 1, 2, and 3 and by Schaefer fixed point theorem, the operator T has at least one fixed point. This ends the proof of the above theorem.

5. Conclusions

In this chapter, the fractional calculus has been applied for some classes of integral inequalities. In fact, using Riemann-Liouville integral, some Minkowski and Hermite-Hadamard-type inequalities have been established. Several other fractional integral results involving a family of positive functions have been also generated. The obtained results generalizes some classical integral inequalities in the literature. In this chapter, we have also presented some applications on continuous random variables; new identities have been established, and some estimates have been discussed.

The existence and the uniqueness of solutions for nonlocal boundary value problem including the Langevin equations with two fractional parameters have been studied. We have used Caputo approach together with Banach contraction principle to prove the existence and uniqueness result. Then, by application of Schaefer fixed point theorem, another existence result has been also proved. Our approach is simple to apply for a variety of real-world problems.

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Determinantal Representations of the Core Inverse and Its Generalizations

Ivan I. Kyrchei

Abstract

Generalized inverse matrices are important objects in matrix theory. In particular, they are useful tools in solving matrix equations. The most famous generalized inverses are the Moore-Penrose inverse and the Drazin inverse. Recently, it was introduced new generalized inverse matrix, namely the core inverse, which was later extended to the core-EP inverse, the BT, DMP, and CMP inverses. In contrast to the inverse matrix that has a definitely determinantal representation in terms of cofactors, even for basic generalized inverses, there exist different determinantal representations as a result of the search of their more applicable explicit expressions. In this chapter, we give new and exclusive determinantal representations of the core inverse and its generalizations by using determinantal representations of the Moore-Penrose and Drazin inverses previously obtained by the author.

Keywords: Moore-Penrose inverse, Drazin inverse, core inverse, core-EP inverse, **2000 AMS subject classifications:** 15A15, 16W10

1. Introduction

In the whole chapter, the notations \mathbb{R} and \mathbb{C} are reserved for fields of the real and complex numbers, respectively. $\mathbb{C}^{m \times n}$ stands for the set of all $m \times n$ matrices over \mathbb{C} . $\mathbb{C}_r^{m \times n}$ determines its subset of matrices with a rank r . For $\mathbf{A} \in \mathbb{C}^{m \times n}$, the symbols \mathbf{A}^* and $\text{rk}(\mathbf{A})$ specify the conjugate transpose and the rank of \mathbf{A} , respectively, $|\mathbf{A}|$ or $\det \mathbf{A}$ stands for its determinant. A matrix $\mathbf{A} \in \mathbb{C}^{n \times n}$ is Hermitian if $\mathbf{A}^* = \mathbf{A}$.

\mathbf{A}^\dagger means the Moore-Penrose inverse of $\mathbf{A} \in \mathbb{C}^{n \times m}$, i.e., the exclusive matrix \mathbf{X} satisfying the following four equations:

$$\mathbf{AXA} = \mathbf{A} \tag{1}$$

$$\mathbf{XAX} = \mathbf{X} \tag{2}$$

$$(\mathbf{AX})^* = \mathbf{AX} \tag{3}$$

$$(\mathbf{XA})^* = \mathbf{XA} \tag{4}$$

For $\mathbf{A} \in \mathbb{C}^{n \times n}$ with index $\text{Ind } \mathbf{A} = k$, i.e., the smallest positive number such that $\text{rk}(\mathbf{A}^{k+1}) = \text{rk}(\mathbf{A}^k)$, the Drazin inverse of \mathbf{A} , denoted by \mathbf{A}^d , is called the unique matrix \mathbf{X} that satisfies Eq. (2) and the following equations,

$$\mathbf{AX} = \mathbf{XA}; \quad (5)$$

$$\mathbf{XA}^{k+1} = \mathbf{A}^k \quad (6)$$

$$\mathbf{A}^{k+1}\mathbf{X} = \mathbf{A}^k. \quad (7)$$

In particular, if $\text{Ind } \mathbf{A} = 1$, then the matrix \mathbf{X} is called *the group inverse*, and it is denoted by $\mathbf{X} = \mathbf{A}^\#$. If $\text{Ind } \mathbf{A} = 0$, then \mathbf{A} is nonsingular and $\mathbf{A}^d = \mathbf{A}^\dagger = \mathbf{A}^{-1}$.

It is evident that if the condition (5) is fulfilled, then (6) and (7) are equivalent. We put both these conditions because they will be used below independently of each other and without the obligatory fulfillment of (5).

A matrix \mathbf{A} satisfying the conditions (i), (j), ... is called an $\{i, j, \dots\}$ -inverse of \mathbf{A} , and is denoted by $\mathbf{A}^{(i,j,\dots)}$. The set of matrices $\mathbf{A}^{(i,j,\dots)}$ is denoted $\mathbf{A}\{i, j, \dots\}$. In particular, $\mathbf{A}^{(1)}$ is called the inner inverse, $\mathbf{A}^{(2)}$ is called the outer inverse, $\mathbf{A}^{(1,2)}$ is called the reflexive inverse, $\mathbf{A}^{(1,2,3,4)}$ is the Moore-Penrose inverse, etc.

For an arbitrary matrix $\mathbf{A} \in \mathbb{C}^{m \times n}$, we denote by

- $\mathcal{N}(\mathbf{A}) = \{\mathbf{x} \in \mathbb{H}^{n \times 1} : \mathbf{Ax} = 0\}$, the kernel (or the null space) of \mathbf{A} ;
- $\mathcal{C}(\mathbf{A}) = \{\mathbf{y} \in \mathbb{H}^{m \times 1} : \mathbf{y} = \mathbf{Ax}, \mathbf{x} \in \mathbb{H}^{n \times 1}\}$, the column space (or the range space) of \mathbf{A} ; and
- $\mathcal{R}(\mathbf{A}) = \{\mathbf{y} \in \mathbb{H}^{1 \times n} : \mathbf{y} = \mathbf{xA}, \mathbf{x} \in \mathbb{H}^{1 \times m}\}$, the row space of \mathbf{A} .

$\mathbf{P}_A := \mathbf{AA}^\dagger$ and $\mathbf{Q}_A := \mathbf{A}^\dagger\mathbf{A}$ are the orthogonal projectors onto the range of \mathbf{A} and the range of \mathbf{A}^* , respectively.

The core inverse was introduced by Baksalary and Trenkler in [1]. Later, it was investigated by S. Malik in [2] and S.Z. Xu et al. in [3], among others.

Definition 1.1. [1] A matrix $\mathbf{X} \in \mathbb{C}^{n \times n}$ is called the core inverse of $\mathbf{A} \in \mathbb{C}^{n \times n}$ if it satisfies the conditions

$$\mathbf{AX} = \mathbf{P}_A, \text{ and } \mathcal{C}(\mathbf{X}) = \mathcal{C}(\mathbf{A}).$$

When such matrix \mathbf{X} exists, it is denoted as \mathbf{A}^\oplus .

In 2014, the core inverse was extended to the core-EP inverse defined by K. Manjunatha Prasad and K.S. Mohana [4]. Other generalizations of the core inverse were recently introduced for $n \times n$ complex matrices, namely BT inverses [5], DMP inverses [2], CMP inverses [6], etc. The characterizations, computing methods, and some applications of the core inverse and its generalizations were recently investigated in complex matrices and rings (see, e.g., [7–18]).

In contrast to the inverse matrix that has a definitely determinantal representation in terms of cofactors, for generalized inverse matrices, there exist different determinantal representations as a result of the search of their more applicable explicit expressions (see, e.g. [19–25]). In this chapter, we get new determinantal representations of the core inverse and its generalizations using recently obtained by the author determinantal representations of the Moore-Penrose inverse and the Drazin inverse over the quaternion skew field, and over the field of complex numbers as a special case [26–34]. Note that a determinantal representation of the core-EP generalized inverse in complex matrices has been derived in [4], based on the determinantal representation of an reflexive inverse obtained in [19, 20].

The chapter is organized as follows: in Section 2, we start with preliminary introduction of determinantal representations of the Moore-Penrose inverse and the

Drazin inverse. In Section 3, we give determinantal representations of the core inverse and its generalizations, namely the right and left core inverses are established in Section 3.1, the core-EP inverses in Section 3.2, the core DMP inverse and its dual in Section 3.3, and finally the CMP inverse in Section 3.4. A numerical example to illustrate the main results is considered in Section 4. Finally, in Section 5, the conclusions are drawn.

2. Preliminaries

Let $\alpha := \{\alpha_1, \dots, \alpha_k\} \subseteq \{1, \dots, m\}$ and $\beta := \{\beta_1, \dots, \beta_k\} \subseteq \{1, \dots, n\}$ be subsets with $1 \leq k \leq \min\{m, n\}$. By \mathbf{A}_β^α , we denote a submatrix of $\mathbf{A} \in \mathbb{H}^{m \times n}$ with rows and columns indexed by α and β , respectively. Then, \mathbf{A}_α^α is a principal submatrix of \mathbf{A} with rows and columns indexed by α , and $|\mathbf{A}|_\alpha^\alpha$ is the corresponding principal minor of the determinant $|\mathbf{A}|$. Suppose that

$$L_{k,n} := \{\alpha : \alpha = (\alpha_1, \dots, \alpha_k), 1 \leq \alpha_1 < \dots < \alpha_k \leq n\}$$

stands for the collection of strictly increasing sequences of $1 \leq k \leq n$ integers chosen from $\{1, \dots, n\}$. For fixed $i \in \alpha$ and $j \in \beta$, put $I_{r,m}\{i\} := \{\alpha : \alpha \in L_{r,m}, i \in \alpha\}$ and $J_{r,n}\{j\} := \{\beta : \beta \in L_{r,n}, j \in \beta\}$.

The j th columns and the i th rows of \mathbf{A} and \mathbf{A}^* denote $\mathbf{a}_{.j}$ and $\mathbf{a}_{.j}^*$ and \mathbf{a}_i and \mathbf{a}_i^* , respectively. By $\mathbf{A}_i(\mathbf{b})$ and $\mathbf{A}_j(\mathbf{c})$, we denote the matrices obtained from \mathbf{A} by replacing its i th row with the row \mathbf{b} , and its j th column with the column \mathbf{c} .

Theorem 2.1. [28] *If $\mathbf{A} \in \mathbb{H}_r^{m \times n}$, then the Moore-Penrose inverse $\mathbf{A}^\dagger = (a_{ij}^\dagger) \in \mathbb{C}^{n \times m}$ possesses the determinantal representations*

$$a_{ij}^\dagger = \frac{\sum_{\beta \in J_{r,n}\{i\}} |(\mathbf{A}^* \mathbf{A})_{.i}(\mathbf{a}_{.j}^*)|_\beta^\beta}{\sum_{\beta \in J_{r,n}} |\mathbf{A}^* \mathbf{A}|_\beta^\beta} = \tag{8}$$

$$= \frac{\sum_{\alpha \in I_{r,m}\{j\}} |(\mathbf{A} \mathbf{A}^*)_{.j}(\mathbf{a}_i^*)|_\alpha^\alpha}{\sum_{\alpha \in I_{r,m}} |\mathbf{A} \mathbf{A}^*|_\alpha^\alpha}. \tag{9}$$

Remark 2.2. For an arbitrary full-rank matrix $\mathbf{A} \in \mathbb{C}_r^{m \times n}$, a row vector $\mathbf{b} \in \mathbb{H}^{1 \times m}$, and a column-vector $\mathbf{c} \in \mathbb{H}^{n \times 1}$, we put, respectively,

$$|(\mathbf{A} \mathbf{A}^*)_{.i}(\mathbf{b})| = \sum_{\alpha \in I_{m,m}\{i\}} |(\mathbf{A} \mathbf{A}^*)_{.i}(\mathbf{b})|_\alpha^\alpha, \quad i = 1, \dots, m,$$

$$|\mathbf{A} \mathbf{A}^*| = \sum_{\alpha \in I_{m,m}} |\mathbf{A} \mathbf{A}^*|_\alpha^\alpha, \quad \text{when } r = m;$$

$$|(\mathbf{A}^* \mathbf{A})_{.j}(\mathbf{c})| = \sum_{\beta \in J_{n,n}\{j\}} |(\mathbf{A}^* \mathbf{A})_{.j}(\mathbf{c})|_\beta^\beta, \quad j = 1, \dots, n,$$

$$|\mathbf{A}^* \mathbf{A}| = \sum_{\beta \in J_{n,n}} |\mathbf{A}^* \mathbf{A}|_\beta^\beta, \quad \text{when } r = n.$$

Corollary 2.3. [21] *Let $\mathbf{A} \in \mathbb{C}_r^{m \times n}$. Then, the following determinantal representations can be obtained*

i. for the projector $\mathbf{Q}_A = (q_{ij})_{n \times n}$,

$$q_{ij} = \frac{\sum_{\beta \in J_{r,n}\{i\}} |(\mathbf{A}^* \mathbf{A})_{.i}(\dot{\mathbf{a}}_{.j})|_{\beta}^{\beta}}{\sum_{\beta \in J_{r,n}} |\mathbf{A}^* \mathbf{A}|_{\beta}^{\beta}} = \frac{\sum_{\alpha \in I_{r,n}\{j\}} |(\mathbf{A}^* \mathbf{A})_{.j}(\dot{\mathbf{a}}_{.i})|_{\alpha}^{\alpha}}{\sum_{\alpha \in I_{r,n}} |\mathbf{A}^* \mathbf{A}|_{\alpha}^{\alpha}}, \quad (10)$$

where $\dot{\mathbf{a}}_{.j}$ is the j th column and $\dot{\mathbf{a}}_{.i}$ is the i th row of $\mathbf{A}^* \mathbf{A}$; and

ii. for the projector $\mathbf{P}_A = (p_{ij})_{m \times m}$,

$$p_{ij} = \frac{\sum_{\alpha \in I_{r,m}\{j\}} |(\mathbf{A} \mathbf{A}^*)_{.j}(\ddot{\mathbf{a}}_{.i})|_{\alpha}^{\alpha}}{\sum_{\alpha \in I_{r,m}} |\mathbf{A} \mathbf{A}^*|_{\alpha}^{\alpha}} = \frac{\sum_{\beta \in J_{r,m}\{i\}} |(\mathbf{A} \mathbf{A}^*)_{.i}(\ddot{\mathbf{a}}_{.j})|_{\beta}^{\beta}}{\sum_{\beta \in J_{r,m}} |\mathbf{A} \mathbf{A}^*|_{\beta}^{\beta}}, \quad (11)$$

where $\ddot{\mathbf{a}}_{.i}$ is the i th row and $\ddot{\mathbf{a}}_{.j}$ is the j th column of $\mathbf{A} \mathbf{A}^*$.

The following lemma gives determinantal representations of the Drazin inverse in complex matrices.

Lemma 2.4. [21] *Let $\mathbf{A} \in \mathbb{C}^{n \times n}$ with $\text{Ind } \mathbf{A} = k$ and $\text{rk } \mathbf{A}^{k+1} = \text{rk } \mathbf{A}^k = r$. Then, the determinantal representations of the Drazin inverse $\mathbf{A}^d = (a_{ij}^d) \in \mathbb{C}^{n \times n}$ are*

$$a_{ij}^d = \frac{\sum_{\beta \in J_{r,n}\{i\}} |(\mathbf{A}^{k+1})_{.i}(\mathbf{a}_{.j}^{(k)})|_{\beta}^{\beta}}{\sum_{\beta \in J_{r,n}} |\mathbf{A}^{k+1}|_{\beta}^{\beta}} = \quad (12)$$

$$= \frac{\sum_{\alpha \in I_{r,n}\{j\}} |(\mathbf{A}^{k+1})_{.j}(\mathbf{a}_{.i}^{(k)})|_{\alpha}^{\alpha}}{\sum_{\alpha \in I_{r,n}} |\mathbf{A}^{k+1}|_{\alpha}^{\alpha}}, \quad (13)$$

where $\mathbf{a}_{.i}^{(k)}$ is the i th row and $\mathbf{a}_{.j}^{(k)}$ is the j th column of \mathbf{A}^k .

Corollary 2.5. [21] *Let $\mathbf{A} \in \mathbb{C}^{n \times n}$ with $\text{Ind } \mathbf{A} = 1$ and $\text{rk } \mathbf{A}^2 = \text{rk } \mathbf{A} = r$. Then, the determinantal representations of the group inverse $\mathbf{A}^{\#} = (a_{ij}^{\#}) \in \mathbb{C}^{n \times n}$ are*

$$a_{ij}^{\#} = \frac{\sum_{\beta \in J_{r,n}\{i\}} |(\mathbf{A}^2)_{.i}(\mathbf{a}_{.j})|_{\beta}^{\beta}}{\sum_{\beta \in J_{r,n}} |\mathbf{A}^2|_{\beta}^{\beta}} = \frac{\sum_{\alpha \in I_{r,n}\{j\}} |(\mathbf{A}^2)_{.j}(\mathbf{a}_{.i})|_{\alpha}^{\alpha}}{\sum_{\alpha \in I_{r,n}} |\mathbf{A}^2|_{\alpha}^{\alpha}}. \quad (14)$$

3. Determinantal representations of the core inverse and its generalizations

3.1 Determinantal representations of the core inverses

Together with the core inverse in [35], the dual core inverse was to be introduced. Since the both these core inverses are equipollent and they are different only in the position relative to the inducting matrix \mathbf{A} , we propose called them as the right and left core inverses regarding to their positions. So, from [1], we have the following definition that is equivalent to Definition 1.1.

Definition 3.1. A matrix $\mathbf{X} \in \mathbb{C}^{n \times n}$ is said to be the right core inverse of $\mathbf{A} \in \mathbb{C}^{n \times n}$ if it satisfies the conditions

$$\mathbf{A}\mathbf{X} = \mathbf{P}_A, \text{ and } \mathcal{C}(\mathbf{X}) = \mathcal{C}(\mathbf{A}).$$

When such matrix \mathbf{X} exists, it is denoted as A^\oplus .

The following definition of the left core inverse can be given that is equivalent to the introduced dual core inverse [35].

Definition 3.2 A matrix $\mathbf{X} \in \mathbb{C}^{n \times n}$ is said to be the left core inverse of $\mathbf{A} \in \mathbb{C}^{n \times n}$ if it satisfies the conditions

$$\mathbf{X}\mathbf{A} = \mathbf{Q}_A, \text{ and } \mathcal{R}(\mathbf{X}) = \mathcal{R}(\mathbf{A}). \quad (15)$$

When such matrix \mathbf{X} exists, it is denoted as A_\oplus .

Remark 3.3. In [35], the conditions of the dual core inverse are given as follows:

$$A_\oplus \mathbf{A} = \mathbf{P}_{A^*}, \text{ and } \mathcal{C}(A_\oplus) \subseteq \mathcal{C}(\mathbf{A}^*).$$

Since $\mathbf{P}_{A^*} = \mathbf{A}^* (\mathbf{A}^*)^\dagger = (\mathbf{A}^\dagger \mathbf{A})^* = \mathbf{A}^\dagger \mathbf{A} = \mathbf{Q}_A$, and $\mathcal{R}(\mathbf{A}) = \mathcal{C}(\mathbf{A}^*)$, then these conditions and (15) are analogous.

Due to [1], we introduce the following sets of quaternion matrices

$$\begin{aligned} \mathbb{C}_n^{\text{CM}} &= \{\mathbf{A} \in \mathbb{C}^{n \times n} : \text{rk } \mathbf{A}^2 = \text{rk } \mathbf{A}\}, \\ \mathbb{C}_n^{\text{EP}} &= \{\mathbf{A} \in \mathbb{C}^{n \times n} : \mathbf{A}^\dagger \mathbf{A} = \mathbf{A}\mathbf{A}^\dagger\} = \{\mathcal{C}(\mathbf{A}) = \mathcal{C}(\mathbf{A}^*)\}. \end{aligned}$$

The matrices from \mathbb{C}_n^{CM} are called group matrices or core matrices. If $\mathbf{A} \in \mathbb{C}_n^{\text{EP}}$, then clearly $\mathbf{A}^\dagger = \mathbf{A}^\#$. It is known that the core inverses of $\mathbf{A} \in \mathbb{C}^{n \times n}$ exist if and only if $\mathbf{A} \in \mathbb{C}_n^{\text{CM}}$ or $\text{Ind } \mathbf{A} = 1$. Moreover, if \mathbf{A} is nonsingular, $\text{Ind } \mathbf{A} = 0$, then its core inverses are the usual inverse. Due to [1], we have the following representations of the right and left core inverses.

Lemma 3.4. [1] Let $\mathbf{A} \in \mathbb{C}_n^{\text{CM}}$. Then,

$$\mathbf{A}^\oplus = \mathbf{A}^\# \mathbf{A}\mathbf{A}^\dagger, \quad (16)$$

$$\mathbf{A}_\oplus = \mathbf{A}^\dagger \mathbf{A}\mathbf{A}^\# \quad (17)$$

Remark 3.5. In Theorems 3.6 and 3.7, we will suppose that $\mathbf{A} \in \mathbb{C}_n^{\text{CM}}$ but $\mathbf{A} \notin \mathbb{C}_n^{\text{EP}}$. Because, if $\mathbf{A} \in \mathbb{C}_n^{\text{CM}}$ and $\mathbf{A} \in \mathbb{C}_n^{\text{EP}}$ (in particular, \mathbf{A} is Hermitian), then from Lemma 3.4 and the definitions of the Moore-Penrose and group inverses, it follows that $\mathbf{A}^\oplus = \mathbf{A}_\oplus = \mathbf{A}^\# = \mathbf{A}^\dagger$.

Theorem 3.6. Let $\mathbf{A} \in \mathbb{C}_n^{\text{CM}}$ and $\text{rk } \mathbf{A}^2 = \text{rk } \mathbf{A} = s$. Then, its right core inverse has the following determinantal representations

$$a_{ij}^{\oplus, r} = \frac{\sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}\mathbf{A}^*)_{j \cdot} (\mathbf{u}_i^{(1)})|_\alpha^\alpha}{\sum_{\beta \in J_{s,n}} |\mathbf{A}^2|_\beta^\beta \sum_{\alpha \in I_{s,n}} |\mathbf{A}\mathbf{A}^*|_\alpha^\alpha} = \quad (18)$$

$$= \frac{\sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A}^2)_{\cdot i} (\mathbf{u}_j^{(2)})|_\beta^\beta}{\sum_{\beta \in J_{s,n}} |\mathbf{A}^2|_\beta^\beta \sum_{\alpha \in I_{s,n}} |\mathbf{A}\mathbf{A}^*|_\alpha^\alpha}, \quad (19)$$

where

$$\mathbf{u}_i^{(1)} = \left[\sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A}^2)_{\cdot i}(\tilde{\mathbf{a}}_f)|_{\beta}^{\beta} \right] \in \mathbb{C}^{1 \times n}, \quad f = 1, \dots, n$$

$$\mathbf{u}_j^{(2)} = \left[\sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}\mathbf{A}^*)_{\cdot j}(\tilde{\mathbf{a}}_l)|_{\alpha}^{\alpha} \right] \in \mathbb{C}^{n \times 1}, \quad l = 1, \dots, n.$$

are the row and column vectors, respectively. Here $\tilde{\mathbf{a}}_f$ and $\tilde{\mathbf{a}}_l$ are the f th column and l th row of $\tilde{\mathbf{A}} := \mathbf{A}^2 \mathbf{A}^*$.

Proof. Taking into account (16), we have for $\#A$,

$$a_{ij}^{\#,r} = \sum_{l=1}^n \sum_{f=1}^n a_{il}^{\#} a_{lf} a_{jf}^{\dagger}. \quad (20)$$

By substituting (14) and (15) in (20), we obtain

$$a_{ij}^{\#,r} = \frac{\sum_{l=1}^n \sum_{f=1}^n \sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A}^2)_{\cdot i}(\mathbf{a}_f)|_{\beta}^{\beta} a_{fl} \sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}\mathbf{A}^*)_{\cdot j}(\mathbf{a}_l^*)|_{\alpha}^{\alpha}}{\sum_{\beta \in J_{s,n}} |\mathbf{A}^2|_{\beta}^{\beta} \sum_{\alpha \in I_{s,n}} |\mathbf{A}\mathbf{A}^*|_{\alpha}^{\alpha}} =$$

$$\frac{\sum_{f=1}^n \sum_{l=1}^n \sum_{\beta \in J_{s,n}\{j\}} |(\mathbf{A}^2)_{\cdot j}(\mathbf{e}_f)|_{\beta}^{\beta} \tilde{a}_{fl} \sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}\mathbf{A}^*)_{\cdot j}(\mathbf{e}_l)|_{\alpha}^{\alpha}}{\sum_{\beta \in J_{s,n}} |\mathbf{A}^2|_{\beta}^{\beta} \sum_{\alpha \in I_{s,n}} |\mathbf{A}\mathbf{A}^*|_{\alpha}^{\alpha}},$$

where \mathbf{e}_l and \mathbf{e}_l are the unit column and row vectors, respectively, such that all their components are 0, except the l th components which are 1; \tilde{a}_{fl} is the (lf) th element of the matrix $\tilde{\mathbf{A}} := \mathbf{A}^2 \mathbf{A}^*$.

Let

$$u_{il}^{(1)} := \sum_{f=1}^n \sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A}^2)_{\cdot i}(\mathbf{e}_f)|_{\beta}^{\beta} \tilde{a}_{fl} = \sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A}^2)_{\cdot i}(\tilde{\mathbf{a}}_l)|_{\beta}^{\beta}, \quad i, l = 1, \dots, n.$$

Construct the matrix $\mathbf{U}_1 = (u_{il}^{(1)}) \in \mathbb{H}^{n \times n}$. It follows that

$$\sum_l u_{il}^{(1)} \sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}\mathbf{A}^*)_{\cdot j}(\mathbf{e}_l)|_{\alpha}^{\alpha} = \sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}\mathbf{A}^*)_{\cdot j}(\mathbf{u}_i^{(1)})|_{\alpha}^{\alpha},$$

where $\mathbf{u}_i^{(1)}$ is the i th row of \mathbf{U}_1 . So, we get (18). If we first consider

$$u_{if}^{(2)} := \sum_l \tilde{a}_{fl} \sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}\mathbf{A}^*)_{\cdot j}(\mathbf{e}_l)|_{\alpha}^{\alpha} = \sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}\mathbf{A}^*)_{\cdot j}(\tilde{\mathbf{a}}_f)|_{\alpha}^{\alpha}, \quad f, j = 1, \dots, n.$$

and construct the matrix $\mathbf{U}_2 = (u_{if}^{(2)}) \in \mathbb{H}^{n \times n}$, then from

$$\sum_{f=1}^n \sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A}^2)_{\cdot i}(\mathbf{e}_f)|_{\beta}^{\beta} u_{if}^{(2)} = \sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A}^2)_{\cdot i}(\mathbf{u}_f^{(2)})|_{\beta}^{\beta},$$

it follows (19). □

Taking into account (17), the following theorem on the determinantal representation of the left core inverse can be proved similarly.

Theorem 3.7. Let $\mathbf{A} \in \mathbb{C}_n^{\text{CM}}$ and $\text{rk } \mathbf{A}^2 = \text{rk } \mathbf{A} = s$. Then for its left core inverse $(\#A) = (a_{ij}^{\#,l})$, we have

$$a_{ij}^{\#,l} = \frac{\sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}^2)_j(\mathbf{v}_i^{(1)})|_{\alpha}^{\alpha}}{\sum_{\beta \in J_{s,n}} |\mathbf{A}^* \mathbf{A}|_{\beta}^{\beta} \sum_{\alpha \in I_{s,n}} |\mathbf{A}^2|_{\alpha}^{\alpha}} = \frac{\sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A}^* \mathbf{A})_i(\mathbf{v}_j^{(2)})|_{\beta}^{\beta}}{\sum_{\beta \in J_{s,n}} |\mathbf{A}^* \mathbf{A}|_{\beta}^{\beta} \sum_{\alpha \in I_{s,n}} |\mathbf{A}^2|_{\alpha}^{\alpha}},$$

where

$$\mathbf{v}_i^{(1)} = \left[\sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A}^* \mathbf{A})_i(\bar{\mathbf{a}}_f)|_{\beta}^{\beta} \right] \in \mathbb{C}^{1 \times n}, \quad f = 1, \dots, n$$

$$\mathbf{v}_j^{(2)} = \left[\sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}^2)_j(\bar{\mathbf{a}}_l)|_{\alpha}^{\alpha} \right] \in \mathbb{C}^{n \times 1}, \quad l = 1, \dots, n.$$

Here $\bar{\mathbf{a}}_f$ and $\bar{\mathbf{a}}_l$ are the f th column and l th row of $\bar{\mathbf{A}} := \mathbf{A}^* \mathbf{A}^2$.

3.2 Determinantal representations of the core-EP inverses

Similar as in [4], we introduce two core-EP inverses.

Definition 3.8. A matrix $\mathbf{X} \in \mathbb{C}^{n \times n}$ is said to be the right core-EP inverse of $\mathbf{A} \in \mathbb{C}^{n \times n}$ if it satisfies the conditions

$$\mathbf{XAX} = \mathbf{A}, \text{ and } \mathcal{C}(\mathbf{X}) = \mathcal{C}(\mathbf{X}^*) = \mathcal{C}(\mathbf{A}^d).$$

It is denoted as \mathbf{A}^{\oplus} .

Definition 3.9. A matrix $\mathbf{X} \in \mathbb{C}^{n \times n}$ is said to be the left core-EP inverse of $\mathbf{A} \in \mathbb{C}^{n \times n}$ if it satisfies the conditions

$$\mathbf{XAX} = \mathbf{A}, \text{ and } \mathcal{R}(\mathbf{X}) = \mathcal{R}(\mathbf{X}^*) = \mathcal{R}(\mathbf{A}^d).$$

It is denoted as \mathbf{A}_{\oplus} .

Remark 3.10. Since $\mathcal{C}((\mathbf{A}^*)^d) = \mathcal{R}(\mathbf{A}^d)$, then the left core inverse \mathbf{A}_{\oplus} of $\mathbf{A} \in \mathbb{C}^{n \times n}$ is similar to the $*$ core inverse introduced in [4], and the dual core-EP inverse introduced in [35].

Due to [4], we have the following representations the core-EP inverses of $\mathbf{A} \in \mathbb{C}^{n \times n}$,

$$\mathbf{A}^{\oplus} = \mathbf{A}^{\{2,3,6a\}} \quad \text{and} \quad \mathcal{C}(\mathbf{A}^{\oplus}) \subseteq \mathcal{C}(\mathbf{A}^k),$$

$$\mathbf{A}_{\oplus} = \mathbf{A}^{\{2,4,6b\}} \quad \text{and} \quad \mathcal{R}(\mathbf{A}_{\oplus}) \subseteq \mathcal{R}(\mathbf{A}^k).$$

Thanks to [35], the following representations of the core-EP inverses will be used for their determinantal representations.

Lemma 3.11. Let $\mathbf{A} \in \mathbb{C}^{n \times n}$ and $\text{Ind } \mathbf{A} = k$. Then

$$\mathbf{A}^\oplus = \mathbf{A}^k \left(\mathbf{A}^{k+1} \right)^\dagger, \tag{21}$$

$$\mathbf{A}_\oplus = \left(\mathbf{A}^{k+1} \right)^\dagger \mathbf{A}^k. \tag{22}$$

Moreover, if $\text{Ind } \mathbf{A} = 1$, then we have the following representations of the right and left core inverses

$$A^\oplus = \mathbf{A} (\mathbf{A}^2)^\dagger, \tag{23}$$

$$A_\oplus = (\mathbf{A}^2)^\dagger \mathbf{A}. \tag{24}$$

Theorem 3.12. Suppose $\mathbf{A} \in \mathbb{C}^{n \times n}$, $\text{Ind } \mathbf{A} = k$, $\text{rk } \mathbf{A}^k = s$, and there exist \mathbf{A}^\oplus and \mathbf{A}_\oplus . Then $\mathbf{A}^\oplus = (a_{ij}^{\oplus,r})$ and $\mathbf{A}_\oplus = (a_{ij}^{\oplus,l})$ possess the determinantal representations, respectively,

$$a_{ij}^{\oplus,r} = \frac{\sum_{\alpha \in I_{s,n}\{j\}} \left| \left(\mathbf{A}^{k+1} \left(\mathbf{A}^{k+1} \right)^* \right)_j \left(\hat{\mathbf{a}}_i \right) \right|_\alpha}{\sum_{\alpha \in I_{s,n}} \left| \mathbf{A}^{k+1} \left(\mathbf{A}^{k+1} \right)^* \right|_\alpha}, \tag{25}$$

$$a_{ij}^{\oplus,l} = \frac{\sum_{\beta \in J_{s,n}\{i\}} \left| \left(\left(\mathbf{A}^{k+1} \right)^* \mathbf{A}^{k+1} \right)_i \left(\check{\mathbf{a}}_j \right) \right|_\beta}{\sum_{\beta \in J_{s,n}} \left| \left(\mathbf{A}^{k+1} \right)^* \mathbf{A}^{k+1} \right|_\beta}, \tag{26}$$

where $\hat{\mathbf{a}}_i$ is the i th row of $\hat{\mathbf{A}} = \mathbf{A}^k \left(\mathbf{A}^{k+1} \right)^*$ and $\check{\mathbf{a}}_j$ is the j th column of $\check{\mathbf{A}} = \left(\mathbf{A}^{k+1} \right)^* \mathbf{A}^k$.

Proof. Consider $\left(\mathbf{A}^{k+1} \right)^\dagger = (a_{ij}^{(k+1,\dagger)})$ and $\mathbf{A}^k = (a_{ij}^{(k)})$. By (21),

$$a_{ij}^{\oplus,r} = \sum_{t=1}^n a_{it}^{(k)} a_{tj}^{(k+1,\dagger)}.$$

Taking into account (9) for the determinantal representation of $\left(\mathbf{A}^{k+1} \right)^\dagger$, we get

$$a_{ij}^{\oplus,r} = \sum_{t=1}^n a_{it}^{(k)} \frac{\sum_{\alpha \in I_{s,n}\{j\}} \left| \left(\mathbf{A}^{k+1} \left(\mathbf{A}^{k+1} \right)^* \right)_j \left(\mathbf{a}_t^{(k+1,*)} \right) \right|_\alpha}{\sum_{\alpha \in I_{r,m}} \left| \mathbf{A}^{k+1} \left(\mathbf{A}^{k+1} \right)^* \right|_\alpha},$$

where $\mathbf{a}_t^{(k+1,*)}$ is the t th row of $\left(\mathbf{A}^{k+1} \right)^*$. Since $\sum_{t=1}^n a_{it}^{(k)} \mathbf{a}_t^{(k+1,*)} = \hat{\mathbf{a}}_i$, then it follows (25).

The determinantal representation (26) can be obtained similarly by integrating (8) for the determinantal representation of $\left(\mathbf{A}^{k+1} \right)^\dagger$ in (22). \square

Taking into account the representations (23)-(24), we obtain the determinantal representations of the right and left core inverses that have more simpler expressions than they are obtained in Theorems 3.6 and 3.7.

Corollary 3.13. Let $\mathbf{A} \in \mathbb{C}_s^{n \times n}$, $\text{Ind } \mathbf{A} = 1$, and there exist A^\oplus and A_\oplus . Then $A^\oplus = (a_{ij}^{\oplus,r})$ and $A_\oplus = (a_{ij}^{\oplus,l})$ can be expressed as follows

$$a_{ij}^{\oplus,r} = \frac{\sum_{\alpha \in I_{s,n}\{j\}} \left| \left(\mathbf{A}^2 (\mathbf{A}^2)^* \right)_{j \cdot} (\hat{\mathbf{a}}_i) \right|_{\alpha}^{\alpha}}{\sum_{\alpha \in I_{s,n}} \left| \mathbf{A}^2 (\mathbf{A}^2)^* \right|_{\alpha}^{\alpha}},$$

$$a_{ij}^{\oplus,l} = \frac{\sum_{\beta \in J_{s,n}\{i\}} \left| \left((\mathbf{A}^2)^* \mathbf{A}^2 \right)_{\cdot i} (\check{\mathbf{a}}_j) \right|_{\beta}^{\beta}}{\sum_{\beta \in J_{s,n}} \left| (\mathbf{A}^2)^* \mathbf{A}^2 \right|_{\beta}^{\beta}},$$

where $\hat{\mathbf{a}}_i$ is the i th row of $\hat{\mathbf{A}} = \mathbf{A}(\mathbf{A}^2)^*$ and $\check{\mathbf{a}}_j$ is the j th column of $\check{\mathbf{A}} = (\mathbf{A}^2)^* \mathbf{A}$.

3.3 Determinantal representations of the DMP and MPD inverses

The concept of the DMP inverse in complex matrices was introduced in [2] by S. Malik and N. Thome.

Definition 3.14. [2] Suppose $\mathbf{A} \in \mathbb{C}^{n \times n}$ and $\text{Ind } \mathbf{A} = k$. A matrix $\mathbf{X} \in \mathbb{C}^{n \times n}$ is said to be the DMP inverse of \mathbf{A} if it satisfies the conditions

$$\mathbf{XAX} = \mathbf{X}, \mathbf{XA} = \mathbf{A}^d \mathbf{A}, \text{ and } \mathbf{A}^k \mathbf{X} = \mathbf{A}^k \mathbf{A}^\dagger. \quad (27)$$

It is denoted as $\mathbf{A}^{d,\dagger}$.

Due to [2], if an arbitrary matrix satisfies the system of Eq. (27), then it is unique and has the following representation

$$\mathbf{A}^{d,\dagger} = \mathbf{A}^d \mathbf{A} \mathbf{A}^\dagger. \quad (28)$$

Theorem 3.15. Let $\mathbf{A} \in \mathbb{C}_s^{n \times n}$, $\text{Ind } \mathbf{A} = k$, and $\text{rk}(\mathbf{A}^k) = s_1$. Then, its DMP inverse $\mathbf{A}^{d,\dagger} = (a_{ij}^{d,\dagger})$ has the following determinantal representations.

$$a_{ij}^{d,\dagger} = \frac{\sum_{\alpha \in I_{s,n}\{j\}} \left| (\mathbf{A} \mathbf{A}^*)_{j \cdot} (\mathbf{u}_i^{(1)}) \right|_{\alpha}^{\alpha}}{\sum_{\beta \in J_{s_1,n}} \left| \mathbf{A}^{k+1} \right|_{\beta}^{\beta} \sum_{\alpha \in I_{s,n}} \left| \mathbf{A} \mathbf{A}^* \right|_{\alpha}^{\alpha}} = \quad (29)$$

$$= \frac{\sum_{\beta \in J_{s_1,n}\{i\}} \left| (\mathbf{A}^{k+1})_{\cdot i} (\mathbf{u}_j^{(2)}) \right|_{\beta}^{\beta}}{\sum_{\beta \in J_{s_1,n}} \left| \mathbf{A}^{k+1} \right|_{\beta}^{\beta} \sum_{\beta \in J_{s,n}} \left| \mathbf{A} \mathbf{A}^* \right|_{\beta}^{\beta}}, \quad (30)$$

where

$$\mathbf{u}_i^{(1)} = \left[\sum_{\beta \in J_{s_1,n}\{i\}} \left| (\mathbf{A}^{k+1})_{\cdot i} (\tilde{\mathbf{a}}_f) \right|_{\beta}^{\beta} \right] \in \mathbb{C}^{1 \times n}, \quad f = 1, \dots, n,$$

$$\mathbf{u}_j^{(2)} = \left[\sum_{\alpha \in I_{s,n}\{j\}} \left| (\mathbf{A} \mathbf{A}^*)_{j \cdot} (\hat{\mathbf{a}}_l) \right|_{\alpha}^{\alpha} \right] \in \mathbb{C}^{n \times 1}, \quad l = 1, \dots, n.$$

Here, $\tilde{\mathbf{a}}_f$ and $\hat{\mathbf{a}}_l$ are the f th column and the l th row of $\tilde{\mathbf{A}} := \mathbf{A}^{k+1} \mathbf{A}^*$.

Proof. Taking into account (28) for $\mathbf{A}^{d,\dagger}$, we get

$$a_{ij}^{d,\dagger} = \sum_{l=1}^n \sum_{f=1}^n a_{il}^d a_{lf} a_{fj}^\dagger. \quad (31)$$

By substituting (12) and (9) for the determinantal representations of \mathbf{A}^d and \mathbf{A}^\dagger in (31), we get

$$\begin{aligned} a_{ij}^{d,\dagger} &= \\ & \sum_{l=1}^n \sum_{f=1}^n \frac{\sum_{\beta \in J_{s_1,n}\{i\}} \left| (\mathbf{A}^{k+1})_{.i} (\mathbf{a}_{.l}^{(k)}) \right|_\beta^\beta}{\sum_{\beta \in J_{s_1,n}} |\mathbf{A}^{k+1}|_\beta^\beta} a_{lf} \frac{\sum_{\alpha \in I_{s,n}\{j\}} \left| (\mathbf{A}\mathbf{A}^*)_{.j} (\mathbf{a}_{.f}^*) \right|_\alpha^\alpha}{\sum_{\alpha \in I_{s,n}} |\mathbf{A}\mathbf{A}^*|_\alpha^\alpha} = \\ & \sum_{l=1}^n \sum_{f=1}^n \frac{\sum_{\beta \in J_{s_1,n}\{i\}} \left| (\mathbf{A}^{k+1})_{.i} (\mathbf{e}_{.l}) \right|_\beta^\beta}{\sum_{\beta \in J_{s_1,n}} |\mathbf{A}^{k+1}|_\beta^\beta} \tilde{a}_{lf} \frac{\sum_{\alpha \in I_{s,n}\{j\}} \left| (\mathbf{A}\mathbf{A}^*)_{.j} (\mathbf{e}_{.f}) \right|_\alpha^\alpha}{\sum_{\alpha \in I_{s,n}} |\mathbf{A}\mathbf{A}^*|_\alpha^\alpha}, \end{aligned} \quad (32)$$

where $\mathbf{e}_{.l}$ and $\mathbf{e}_{.f}$ are the l th unit column and row vectors, and \tilde{a}_{lf} is the (lf) th element of the matrix $\tilde{\mathbf{A}} = \mathbf{A}^{k+1} \mathbf{A}^*$. If we put

$$u_{if}^{(1)} := \sum_{l=1}^n \sum_{\beta \in J_{s_1,n}\{i\}} \left| (\mathbf{A}^{k+1})_{.i} (\mathbf{e}_{.l}) \right|_\beta^\beta \tilde{a}_{lf} = \sum_{\beta \in J_{s_1,n}\{i\}} \left| (\mathbf{A}^{k+1})_{.i} (\tilde{\mathbf{a}}_{.f}) \right|_\beta^\beta,$$

as the f th component of the row vector $\mathbf{u}_{i.}^{(1)} = [u_{i1}^{(1)}, \dots, u_{in}^{(1)}]$, then from

$$\sum_{f=1}^n u_{if}^{(1)} \sum_{\alpha \in I_{s,n}\{j\}} \left| (\mathbf{A}\mathbf{A}^*)_{.j} (\mathbf{e}_{.f}) \right|_\alpha^\alpha = \sum_{\alpha \in I_{s,n}\{j\}} \left| (\mathbf{A}\mathbf{A}^*)_{.j} (\mathbf{u}_{i.}^{(1)}) \right|_\alpha^\alpha,$$

it follows (29). If we initially obtain

$$u_{lj}^{(2)} := \sum_{f=1}^n \tilde{a}_{lf} \sum_{\alpha \in I_{s,n}\{j\}} \left| (\mathbf{A}\mathbf{A}^*)_{.j} (\mathbf{e}_{.f}) \right|_\alpha^\alpha = \sum_{\alpha \in I_{s,n}\{j\}} \left| (\mathbf{A}\mathbf{A}^*)_{.j} (\tilde{\mathbf{a}}_{.l}) \right|_\alpha^\alpha,$$

as the l th component of the column vector $\mathbf{u}_{.j}^{(2)} = [u_{1j}^{(2)}, \dots, u_{nj}^{(2)}]$, then from

$$\sum_{l=1}^n \sum_{\beta \in J_{s_1,n}\{i\}} \left| (\mathbf{A}^{k+1})_{.i} (\mathbf{e}_{.l}) \right|_\beta^\beta u_{lj}^{(2)} = \sum_{\beta \in J_{s_1,n}\{i\}} \left| (\mathbf{A}^{k+1})_{.i} (\mathbf{u}_{.j}^{(2)}) \right|_\beta^\beta,$$

it follows (30). \square

The name of the DMP inverse is in accordance with the order of using the Drazin inverse (D) and the Moore-Penrose (MP) inverse. In that connection, it would be logical to consider the following definition.

Definition 3.16. Suppose $\mathbf{A} \in \mathbb{C}^{n \times n}$ and $\text{Ind } \mathbf{A} = k$. A matrix $\mathbf{X} \in \mathbb{C}^{n \times n}$ is said to be the MPD inverse of \mathbf{A} if it satisfies the conditions

$$\mathbf{XAX} = \mathbf{X}, \mathbf{AX} = \mathbf{AA}^d, \text{ and } \mathbf{XA}^k = \mathbf{A}^\dagger \mathbf{A}^k.$$

It is denoted as $\mathbf{A}^{\dagger,d}$.

The matrix $\mathbf{A}^{\dagger,d}$ is unique, and it can be represented as

$$\mathbf{A}^{\dagger,d} = \mathbf{A}^\dagger \mathbf{A} \mathbf{A}^d. \tag{33}$$

Theorem 3.17. Let $\mathbf{A} \in \mathbb{C}_s^{n \times n}$, $\text{Ind } \mathbf{A} = k$, and $\text{rk } \mathbf{A}^k = s_1$. Then, its MPD inverse $\mathbf{A}^{\dagger,d} = (a_{ij}^{\dagger,d})$ has the following determinantal representations

$$a_{ij}^{\dagger,d} = \frac{\sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A}^* \mathbf{A})_{.i}(\mathbf{v}_{.j}^{(1)})|_\beta^\beta}{\sum_{\beta \in J_{s,n}} |\mathbf{A}^* \mathbf{A}|_\beta^\beta \sum_{\beta \in I_{s_1,n}} |\mathbf{A}^{k+1}|_\alpha^\alpha} = \frac{\sum_{\alpha \in I_{s_1,n}\{j\}} |(\mathbf{A}^{k+1})_{.j}(\mathbf{v}_{.i}^{(2)})|_\alpha^\alpha}{\sum_{\alpha \in I_{s_1,n}} |\mathbf{A}^* \mathbf{A}|_\beta^\beta \sum_{\alpha \in I_{s,n}} |\mathbf{A}^{k+1}|_\alpha^\alpha},$$

where

$$\mathbf{v}_{.j}^{(1)} = \left[\sum_{\alpha \in I_{s_1,n}\{j\}} |(\mathbf{A}^{k+1})_{.j}(\hat{\mathbf{a}}_{.l})|_\alpha^\alpha \right] \in \mathbb{C}^{n \times 1}, \quad l = 1, \dots, n$$

$$\mathbf{v}_{.i}^{(2)} = \left[\sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A}^* \mathbf{A})_{.i}(\hat{\mathbf{a}}_{.f})|_\beta^\beta \right] \in \mathbb{C}^{1 \times n}, \quad l = 1, \dots, n.$$

Here, $\hat{\mathbf{a}}_{.l}$ and $\hat{\mathbf{a}}_{.f}$ are the l th row and the f th column of $\hat{\mathbf{A}} := \mathbf{A}^* \mathbf{A}^{k+1}$.

Proof. The proof is similar to the proof of Theorem 3.15. □

3.4 Determinantal representations of the CMP inverse

Definition 3.18. [6] Suppose $\mathbf{A} \in \mathbb{C}^{n \times n}$ has the core-nilpotent decomposition $\mathbf{A} = \mathbf{A}_1 + \mathbf{A}_2$, where $\text{Ind } \mathbf{A}_1 = \text{Ind } \mathbf{A}$, \mathbf{A}_2 is nilpotent, and $\mathbf{A}_1 \mathbf{A}_2 = \mathbf{A}_2 \mathbf{A}_1 = 0$. The CMP inverse of \mathbf{A} is called the matrix $\mathbf{A}^{c,\dagger} := \mathbf{A}^\dagger \mathbf{A}_1 \mathbf{A}^\dagger$.

Lemma 3.19. [6] Let $\mathbf{A} \in \mathbb{C}^{n \times n}$. The matrix $\mathbf{X} = \mathbf{A}^{c,\dagger}$ is the unique matrix that satisfies the following system of equations:

$$\mathbf{X} \mathbf{A} \mathbf{X} = \mathbf{X}, \mathbf{A} \mathbf{X} \mathbf{A} = \mathbf{A}_1, \mathbf{A} \mathbf{X} = \mathbf{A}_1 \mathbf{A}^\dagger, \text{ and } \mathbf{X} \mathbf{A} = \mathbf{A}^\dagger \mathbf{A}_1.$$

Moreover,

$$\mathbf{A}^{c,\dagger} = \mathbf{Q}_A \mathbf{A}^d \mathbf{P}_A. \tag{34}$$

Taking into account (34), it follows the next theorem about determinantal representations of the quaternion CMP inverse.

Theorem 3.20. Let $\mathbf{A} \in \mathbb{C}_s^{n \times n}$, $\text{Ind } \mathbf{A} = m$, and $\text{rk}(\mathbf{A}^m) = s_1$. Then, the determinantal representations of its CMP inverse $\mathbf{A}^{c,\dagger} = (a_{ij}^{c,\dagger})$ can be expressed as

$$a_{ij}^{c,\dagger} = \frac{\sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A}^* \mathbf{A})_{.i}(\mathbf{v}_{.j}^{(l)})|_\beta^\beta}{\left(\sum_{\beta \in J_{s,n}} |\mathbf{A}^* \mathbf{A}|_\beta^\beta \right)^2 \sum_{\beta \in J_{s_1,n}} |\mathbf{A}^{m+1}|_\beta^\beta} \tag{35}$$

$$a_{ij}^{c,\dagger} = \frac{\sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{AA}^*)_j(\mathbf{w}_i^{(l)})|_\alpha^\alpha}{\left(\sum_{\alpha \in I_{s,n}} |\mathbf{AA}^*|_\alpha^\alpha\right)^2 \sum_{\beta \in J_{s_1,n}} |\mathbf{A}^{m+1}|_\beta^\beta} \quad (36)$$

for all $l = 1, 2$, where

$$\mathbf{v}_j^{(1)} = \left[\sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{AA}^*)_j(\hat{\mathbf{u}}_t)|_\alpha^\alpha \right] \in \mathbb{C}^{n \times 1}, t = 1, \dots, n, \quad (37)$$

$$\mathbf{w}_i^{(1)} = \left[\sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A}^* \mathbf{A})_i(\hat{\mathbf{u}}_k)|_\beta^\beta \right] \in \mathbb{C}^{1 \times n}, k = 1, \dots, n, \quad (38)$$

$$\mathbf{v}_j^{(2)} = \left[\sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}^* \mathbf{A})_j(\tilde{\mathbf{g}}_t)|_\alpha^\alpha \right] \in \mathbb{C}^{n \times 1}, t = 1, \dots, n, \quad (39)$$

$$\mathbf{w}_i^{(2)} = \left[\sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A}^* \mathbf{A})_i(\tilde{\mathbf{g}}_k)|_\beta^\beta \right] \in \mathbb{C}^{1 \times n}, k = 1, \dots, n. \quad (40)$$

Here, $\hat{\mathbf{u}}_t$ is the t th row and $\hat{\mathbf{u}}_k$ is the k th column of $\hat{\mathbf{U}} := \mathbf{UAA}^*$, $\tilde{\mathbf{g}}_t$ is the t th row and $\tilde{\mathbf{g}}_k$ is the k th column of $\tilde{\mathbf{G}} := \mathbf{A}^* \mathbf{AG}$, and the matrices $\mathbf{U} = (u_{ij}) \in \mathbb{H}^{n \times n}$ and $\mathbf{G} = (g_{ij}) \in \mathbb{H}^{n \times n}$ are such that

$$u_{ij} = \sum_{\alpha \in I_{s_1,n}\{j\}} |(\mathbf{A}^{m+1})_j(\hat{\mathbf{a}}_i)|_\alpha^\alpha, \quad g_{ij} = \sum_{\beta \in J_{s_1,n}\{i\}} |(\mathbf{A}^{m+1})_i(\tilde{\mathbf{a}}_j)|_\beta^\beta,$$

where $\hat{\mathbf{a}}_i$ is the i th row of $\hat{\mathbf{A}} := \mathbf{A}^* \mathbf{A}^{m+1}$ and $\tilde{\mathbf{a}}_j$ is the j th column of $\tilde{\mathbf{A}} := \mathbf{A}^{m+1} \mathbf{A}^*$.
Proof. Taking into account (34), we get

$$a_{ij}^{c,\dagger} = \sum_{l=1}^n \sum_{k=1}^n q_{il}^A a_{lk}^d p_{kj}^A, \quad (41)$$

where $\mathbf{Q}_A = (q_{il}^A)$, $\mathbf{A}^d = (a_{il}^d)$, and $\mathbf{P}_A = (p_{il}^A)$.

a. Taking into account the expressions (13), (10), and (11) for the determinantal representations of \mathbf{A}^d , \mathbf{Q}_A , and \mathbf{P}_A , respectively, we have

$$a_{ij}^{c,\dagger} = \sum_l \sum_t \frac{\sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A}^* \mathbf{A})_i(\hat{\mathbf{a}}_t)|_\beta^\beta \sum_{\alpha \in I_{s_1,n}\{l\}} |(\mathbf{A}^{m+1})_l(\mathbf{a}_t^{(m)})|_\alpha^\alpha \sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{AA}^*)_j(\hat{\mathbf{a}}_l)|_\alpha^\alpha}{\sum_{\beta \in J_{s,n}} |\mathbf{A}^* \mathbf{A}|_\beta^\beta \sum_{\alpha \in I_{s_1,n}} |\mathbf{A}^{m+1}|_\alpha^\alpha \sum_{\alpha \in I_{s,n}} |\mathbf{AA}^*|_\alpha^\alpha},$$

where $\hat{\mathbf{a}}_t$ is the t th column of $\mathbf{A}^* \mathbf{A}$, $\hat{\mathbf{a}}_l$ is the l th row of \mathbf{AA}^* , and $\mathbf{a}_t^{(m)}$ is the t th row of \mathbf{A}^m . So, it is clear that

$$a_{ij}^{c,\dagger} = \sum_l \sum_t \sum_k \frac{\sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A}^* \mathbf{A})_i(\mathbf{e}_t)|_\beta^\beta \hat{a}_{tk} \sum_{\alpha \in I_{s_1,n}\{l\}} |(\mathbf{A}^{m+1})_l(\mathbf{e}_k)|_\alpha^\alpha \sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{AA}^*)_j(\hat{\mathbf{a}}_l)|_\alpha^\alpha}{\sum_{\beta \in J_{s,n}} |\mathbf{A}^* \mathbf{A}|_\beta^\beta \sum_{\alpha \in I_{s_1,n}} |\mathbf{A}^{m+1}|_\alpha^\alpha \sum_{\alpha \in I_{s,n}} |\mathbf{AA}^*|_\alpha^\alpha},$$

where \mathbf{e}_t is the t th unit column vector, \mathbf{e}_k is the k th row vector, and \hat{a}_{tk} is the (tk) th element of $\hat{\mathbf{A}} = \mathbf{A} * \mathbf{A}^{m+1}$.

Denote

$$u_{tl} := \sum_k \hat{a}_{tk} \sum_{\alpha \in I_{s_1,n}\{j\}} |(\mathbf{A}^{m+1})_l(\mathbf{e}_k)|_\alpha^\alpha = \sum_{\alpha \in I_{s_1,n}\{j\}} |(\mathbf{A}^{m+1})_l(\hat{\mathbf{a}}_t)|_\alpha^\alpha \quad (42)$$

as the t th component of a column vector $\mathbf{u}_l = [u_{1l}, \dots, u_{nl}]$. Then from

$$\sum_t \sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A} * \mathbf{A})_i(\mathbf{e}_t)|_\beta^\beta u_{tl} = \sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A} * \mathbf{A})_i(\mathbf{u}_l)|_\beta^\beta,$$

we have

$$a_{ij}^{c,\dagger} = \sum_l \frac{\sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A} * \mathbf{A})_i(\mathbf{u}_l)|_\beta^\beta \sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}\mathbf{A}^*)_j(\hat{\mathbf{a}}_l)|_\alpha^\alpha}{\sum_{\beta \in J_{r,n}} |\mathbf{A} * \mathbf{A}|_\beta^\beta \sum_{\alpha \in I_{s_1,n}} |\mathbf{A}^{m+1}|_\alpha^\alpha \sum_{\alpha \in I_{s,n}} |\mathbf{A}\mathbf{A}^*|_\alpha^\alpha}.$$

Construct the matrix $\mathbf{U} = (u_{tl}) \in \mathbb{H}^{n \times n}$, where u_{tl} is given by (42), and denote $\hat{\mathbf{U}} := \mathbf{U}\mathbf{A}\mathbf{A}^*$. Then, taking into account that $|\mathbf{A} * \mathbf{A}|_\beta^\beta = |\mathbf{A}\mathbf{A}^*|_\alpha^\alpha$, we have

$$a_{ij}^{c,\dagger} = \frac{\sum_t \sum_k \sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A} * \mathbf{A})_i(\mathbf{e}_t)|_\beta^\beta \hat{u}_{tk} \sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}\mathbf{A}^*)_j(\mathbf{e}_k)|_\alpha^\alpha}{\left(\sum_{\beta \in J_{s,n}} |\mathbf{A} * \mathbf{A}|_\beta^\beta\right)^2 \sum_{\alpha \in I_{s_1,n}} |\mathbf{A}^{m+1}|_\alpha^\alpha}.$$

If we put that

$$v_{tj}^{(1)} := \sum_k \hat{u}_{tk} \sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}\mathbf{A}^*)_j(\mathbf{e}_k)|_\alpha^\alpha = \sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}\mathbf{A}^*)_j(\hat{\mathbf{u}}_t)|_\alpha^\alpha$$

is the t th component of a column vector $\mathbf{v}_j^{(1)} = [v_{1j}^{(1)}, \dots, v_{nj}^{(1)}]$, then from

$$\sum_t \sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A} * \mathbf{A})_i(\mathbf{e}_t)|_\beta^\beta v_{tj}^{(1)} = \sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A} * \mathbf{A})_i(\mathbf{v}_j^{(1)})|_\beta^\beta,$$

it follows (35) with $\mathbf{v}_j^{(1)}$ given by (37). If we initially put

$$w_{ik}^{(1)} := \sum_t \sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A} * \mathbf{A})_i(\mathbf{e}_t)|_\beta^\beta \hat{u}_{tk} = \sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A} * \mathbf{A})_i(\hat{\mathbf{u}}_k)|_\beta^\beta$$

as the k th component of the row vector $\mathbf{w}_i^{(1)} = [w_{i1}^{(1)}, \dots, w_{in}^{(1)}]$, then from

$$\sum_k w_{ik}^{(1)} \sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}^2)_j(\mathbf{e}_k)|_\alpha^\alpha = \sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}^2)_j(\mathbf{w}_i^{(1)})|_\alpha^\alpha,$$

it follows (36) with $\mathbf{w}_i^{(1)}$ given by (38).

b. By using the determinantal representation (12) for \mathbf{A}^d in (41), we have

$$a_{ij}^{c,\dagger} = \sum_k \sum_t \frac{\sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A}^* \mathbf{A})_{.i}(\dot{\mathbf{a}}_t)|_\beta^\beta \sum_{\beta \in J_{s_1,n}\{t\}} |(\mathbf{A}^{m+1})_{.t}(\mathbf{a}_{.k}^{(m)})|_\beta^\beta \sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}\mathbf{A}^*)_{.j}(\dot{\mathbf{a}}_{k.})|_\alpha^\alpha}{\sum_{\beta \in J_{s,n}} |\mathbf{A}^* \mathbf{A}|_\beta^\beta \sum_{\beta \in J_{s,n}} |\mathbf{A}^{m+1}|_\beta^\beta \sum_{\alpha \in I_{s,n}} |\mathbf{A}\mathbf{A}^*|_\alpha^\alpha}.$$

Therefore,

$$a_{ij}^{c,\dagger} = \sum_l \sum_k \sum_t \frac{\sum_{\beta \in J_{s,n}\{i\}} ((\mathbf{A}^* \mathbf{A})_{.i}(\dot{\mathbf{a}}_t))_\beta^\beta}{\sum_{\beta \in J_{s,n}} |\mathbf{A}^* \mathbf{A}|_\beta^\beta} \times \frac{\sum_{\beta \in J_{s_1,n}\{t\}} |(\mathbf{A}^{m+1})_{.t}(\mathbf{e}_{.k})|_\beta^\beta}{\sum_{\beta \in J_{s_1,n}} |\mathbf{A}^{m+1}|_\beta^\beta} \tilde{a}_{kl} \frac{\sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}\mathbf{A}^*)_{.j}(\mathbf{e}_{l.})|_\alpha^\alpha}{\sum_{\alpha \in I_{s,n}} |\mathbf{A}\mathbf{A}^*|_\alpha^\alpha}.$$

where $\mathbf{e}_{.k}$ is the k th unit column vector, $\mathbf{e}_{l.}$ is the l th unit row vector, and \tilde{a}_{kl} is the (kl) th element of $\tilde{\mathbf{A}} = \mathbf{A}^{m+1} \mathbf{A}^*$.

If we denote

$$g_{tl} := \sum_{\beta \in J_{s_1,n}\{t\}} |(\mathbf{A}^{m+1})_{.t}(\mathbf{e}_{.k})|_\beta^\beta \tilde{a}_{kl} = \sum_{\beta \in J_{s_1,n}\{t\}} |(\mathbf{A}^{m+1})_{.t}(\tilde{\mathbf{a}}_{l.})|_\beta^\beta \quad (43)$$

as the l th component of a row vector $\mathbf{g}_t = [g_{t1}, \dots, g_{tn}]$, then

$$\sum_l g_{tl} \sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}\mathbf{A}^*)_{.j}(\mathbf{e}_{l.})|_\alpha^\alpha = \sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}\mathbf{A}^*)_{.j}(\mathbf{g}_t.)|_\alpha^\alpha.$$

From this, it follows that

$$a_{ij}^{c,\dagger} = \sum_t \frac{\sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A}^* \mathbf{A})_{.i}(\dot{\mathbf{a}}_t)|_\beta^\beta \sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}\mathbf{A}^*)_{.j}(\mathbf{g}_t.)|_\alpha^\alpha}{\sum_{\beta \in J_{s,n}} |\mathbf{A}^* \mathbf{A}|_\beta^\beta \sum_{\alpha \in I_{s_1,n}} |\mathbf{A}^{m+1}|_\alpha^\alpha \sum_{\alpha \in I_{s,n}} |\mathbf{A}\mathbf{A}^*|_\alpha^\alpha}.$$

Construct the matrix $\mathbf{G} = (g_{tl}) \in \mathbb{H}^{n \times n}$, where g_{tl} is given by (43). Denote $\tilde{\mathbf{G}} := \mathbf{A}^* \mathbf{A} \mathbf{G}$. Then,

$$a_{ij}^{c,\dagger} = \frac{\sum_t \sum_k \sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A}^* \mathbf{A})_{.i}(\mathbf{e}_{.t})|_\beta^\beta \tilde{g}_{tk} \sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}\mathbf{A}^*)_{.j}(\mathbf{e}_{k.})|_\alpha^\alpha}{\sum_{\beta \in J_{s,n}} (|\mathbf{A}^* \mathbf{A}|_\beta^\beta)^2 \sum_{\alpha \in I_{s_1,n}} |\mathbf{A}^{m+1}|_\alpha^\alpha}.$$

If we denote

$$v_{tj}^{(2)} := \sum_k \tilde{g}_{tk} \sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}\mathbf{A}^*)_{.j}(\mathbf{e}_{k.})|_\alpha^\alpha = \sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}\mathbf{A}^*)_{.j}(\tilde{\mathbf{g}}_t.)|_\alpha^\alpha$$

as the t th component of a column vector $\mathbf{v}_{\cdot j}^{(2)} = [v_{1j}^{(2)}, \dots, v_{nj}^{(2)}]$, then

$$\sum_t \sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A}^* \mathbf{A})_{\cdot i}(\mathbf{e}_t)|_{\beta}^{\beta} v_{tj}^{(2)} = \sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A}^* \mathbf{A})_{\cdot i}(\mathbf{v}_{\cdot j}^{(2)})|_{\beta}^{\beta}.$$

Thus, we have (35) with $\mathbf{v}_{\cdot j}^{(2)}$ given by (39).

If, now, we denote

$$w_{ik}^{(2)} := \sum_t \sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A}^* \mathbf{A})_{\cdot i}(\mathbf{e}_t)|_{\beta}^{\beta} \tilde{g}_{tk} = \sum_{\beta \in J_{s,n}\{i\}} |(\mathbf{A}^* \mathbf{A})_{\cdot i}(\tilde{\mathbf{g}}_{\cdot k})|_{\beta}^{\beta}$$

as the k th component of a row vector $\mathbf{w}_i^{(2)} = [w_{i1}^{(2)}, \dots, w_{in}^{(2)}]$, then

$$\sum_k w_{ik}^{(2)} \sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}\mathbf{A}^*)_{\cdot j}(\mathbf{e}_k)|_{\alpha}^{\alpha} = \sum_{\alpha \in I_{s,n}\{j\}} |(\mathbf{A}\mathbf{A}^*)_{\cdot j}(\mathbf{w}_i^{(2)})|_{\alpha}^{\alpha}.$$

So, finally, we have (36) with $\mathbf{w}_i^{(2)}$ given by (40).

4. An example

Given the matrix

$$\mathbf{A} = \begin{bmatrix} 2 & 0 & 0 \\ -i & i & i \\ -i & -i & -i \end{bmatrix}.$$

Since

$$\mathbf{A}\mathbf{A}^* = \begin{bmatrix} 4 & 2i & 2i \\ -2i & 3 & -1 \\ -2i & -1 & 3 \end{bmatrix}, \quad \mathbf{A}^2 = \begin{bmatrix} 4 & 0 & 0 \\ 2-2i & 0 & 0 \\ -2-2i & 0 & 0 \end{bmatrix}, \quad \mathbf{A}^3 = \begin{bmatrix} 8 & 0 & 0 \\ 4-4i & 0 & 0 \\ -4-4i & 0 & 0 \end{bmatrix},$$

then $\text{rk } \mathbf{A} = 2$ and $\text{rk } \mathbf{A}^2 = \text{rk } \mathbf{A}^3 = 1$, and $k = \text{Ind } \mathbf{A} = 2$ and $r_1 = 1$. So, we shall find \mathbf{A}^{\oplus} and \mathbf{A}_{\oplus} by (25) and (26), respectively.

Since

$$\hat{\mathbf{A}} = \mathbf{A}^2(\mathbf{A}^3)^* = 16 \begin{bmatrix} 2 & 1+i & -1+i \\ 1-i & 1 & i \\ -1-i & i & 1 \end{bmatrix},$$

then by (25),

$$a_{11}^{\oplus,r} = \frac{\sum_{\alpha \in I_{1,3}\{1\}} |(\mathbf{A}^3(\mathbf{A}^3)^*)_{\cdot 1}(\hat{\mathbf{a}}_1)|_{\alpha}^{\alpha}}{\sum_{\alpha \in I_{1,3}} |\mathbf{A}^3(\mathbf{A}^3)^*|_{\alpha}^{\alpha}} = \frac{1}{4}.$$

By similarly continuing, we get

$$\mathbf{A}^{\oplus} = \frac{1}{8} \begin{bmatrix} 2 & 1+i & -1+i \\ 1-i & 1 & i \\ -1-i & i & 1 \end{bmatrix}.$$

By analogy, due to (26), we have

$$\mathbf{A}_{\oplus} = \frac{1}{2} \begin{bmatrix} 1 & 0 & 0 \\ 0 & 0 & 0 \\ 0 & 0 & 0 \end{bmatrix}.$$

The DMP inverse $\mathbf{A}^{d,\dagger}$ can be found by Theorem 3.15. Since

$$\tilde{\mathbf{A}} = \mathbf{A}^3 \mathbf{A}^* = 4 \begin{bmatrix} 4 & 2i & 2i \\ 2-2i & 1+i & 1+i \\ -2-2i & 1-i & 1-i \end{bmatrix}.$$

and $\text{rk}(\mathbf{A}^3) = 1$, then

$$\mathbf{u}_1^{(1)} = \tilde{\mathbf{a}}_1, \quad \mathbf{u}_2^{(1)} = \tilde{\mathbf{a}}_2, \quad \mathbf{u}_3^{(1)} = \tilde{\mathbf{a}}_3.$$

Furthermore, by (29),

$$a_{11}^{d,\dagger} = \frac{\sum_{\alpha \in I_{2,3}\{1\}} |(\mathbf{A}\mathbf{A}^*)_1(\mathbf{u}_1^{(1)})|_{\alpha}^{\alpha}}{\sum_{\beta \in J_{1,3}} |\mathbf{A}^3|_{\beta}^{\beta} \sum_{\alpha \in I_{2,3}} |\mathbf{A}\mathbf{A}^*|_{\alpha}^{\alpha}} = \frac{1}{192} \left(\det \begin{bmatrix} 16 & 8i \\ -2i & 3 \end{bmatrix} + \det \begin{bmatrix} 16 & 8i \\ -2i & 3 \end{bmatrix} \right) = \frac{1}{3}.$$

By similarly continuing, we get

$$\mathbf{A}^{d,\dagger} = \frac{1}{12} \begin{bmatrix} 4 & 2i & 2i \\ 2-2i & 1+i & 1+i \\ -2-2i & 1-i & 1-i \end{bmatrix}.$$

Similarly by Theorem 3.17, we get

$$\mathbf{A}^{\dagger,d} = \frac{1}{4} \begin{bmatrix} 2 & 0 & 0 \\ -i & 0 & 0 \\ -i & 0 & 0 \end{bmatrix}.$$

Finally, by theorem, we find the CMP inverse $\mathbf{A}^{c,\dagger} = (a_{ij}^{c,\dagger})$. Since $\text{rk} \mathbf{A}^3 = 1$, then $\mathbf{G} = \tilde{\mathbf{A}}$ and

$$\tilde{\mathbf{G}} = \mathbf{A}^* \mathbf{A} \tilde{\mathbf{A}} = 16 \begin{bmatrix} 6 & 3\mathbf{i} & 3\mathbf{i} \\ -2\mathbf{i} & 1 & 1 \\ -2\mathbf{i} & 1 & 1 \end{bmatrix}.$$

Furthermore, by (40),

$$w_{11}^{(2)} = \sum_{\beta \in J_{2,3}\{1\}} |(\mathbf{A}^* \mathbf{A})_{.1}(\tilde{\mathbf{g}}_{.1})|_{\beta}^{\beta} = \left(\det \begin{bmatrix} 6 & 0 \\ -2\mathbf{i} & 2 \end{bmatrix} + \det \begin{bmatrix} 6 & 0 \\ -2\mathbf{i} & 2 \end{bmatrix} \right) = 24.$$

By similar calculations, we get

$$\mathbf{w}_1^{(2)} = [384, 96\mathbf{i}, 96\mathbf{i}], \quad \mathbf{w}_2^{(2)} = [-192\mathbf{i}, 96, 96], \quad \mathbf{w}_3^{(2)} = [-192\mathbf{i}, 96, 06].$$

So, by (36), we get

$$\begin{aligned} a_{11}^{c,\dagger} &= \frac{\sum_{\alpha \in I_{2,3}\{1\}} |(\mathbf{A}\mathbf{A}^*)_{.1}(\mathbf{w}_1^{(2)})|_{\alpha}^{\alpha}}{\left(\sum_{\alpha \in I_{2,3}} |\mathbf{A}\mathbf{A}^*|_{\alpha}^{\alpha} \right)^2 \sum_{\beta \in J_{1,3}} |\mathbf{A}^3|_{\beta}^{\beta}} \\ &= \frac{1}{4608} \left(\det \begin{bmatrix} 384 & 192\mathbf{i} \\ -2\mathbf{i} & 3 \end{bmatrix} + \det \begin{bmatrix} 384 & 192\mathbf{i} \\ -2\mathbf{i} & 3 \end{bmatrix} \right) = \frac{1}{3}. \end{aligned}$$

By similarly continuing, we derive

$$\mathbf{A}^{c,\dagger} = \frac{1}{12} \begin{bmatrix} 4 & 2\mathbf{i} & 2\mathbf{i} \\ -2\mathbf{i} & 1 & 1 \\ -2\mathbf{i} & 1 & 1 \end{bmatrix}.$$

5. Conclusions

In this chapter, we get the direct method to find the core inverse and its generalizations that are based on their determinantal representations. New determinantal representations of the right and left core inverses, the right and left core-EP inverses, the DMP, MPD, and CMP inverses are derived.



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Folding on the Chaotic Graph Operations and Their Fundamental Group

Mohammed Abu Saleem

Abstract

Our aim in the present chapter is to introduce a new type of operations on the chaotic graph, namely, chaotic connected edge graphs under the identification topology. The concept of chaotic foldings on the chaotic edge graph will be discussed from the viewpoint of algebra and geometry. The relation between the chaotic homeomorphisms and chaotic foldings on the chaotic connected edge graphs and their fundamental group is deduced. The fundamental group of the limit chaotic chain of foldings on chaotic. Many types of chaotic foldings are achieved. Theorems governing these relations are achieved. We also discuss some applications in chemistry and biology.

Keywords: chaotic graph, edge graph, chaotic folding, limit folding fundamental group

2010 Mathematics Subject Classification: 51H20, 57N10, 57M05, 14F35, 20F34

1. Introduction and definitions

During the past few decades, examinations of social, biological, and communication networks have taken on enhanced attention throughout these examinations; graphical representations of those networks and systems have been evident to be terribly helpful. Such representations are accustomed to confirm or demonstrate the interconnections or relationships between parts of those networks [1, 2].

A graph is an ordered $G = (V(G), E(G))$ where $V(G) \neq \varnothing$, $E(G)$ is a set disjoint from $V(G)$, elements of $V(G)$ are called the vertices of G , and elements of $E(G)$ are called the edges. The foundation stone of graph theory was laid by Euler in 1736 by solving a puzzle called Königsberg seven-bridge problem as in **Figure 1** [1, 3].

There are many graphs with which one can construct a new graph from a given graph or set of graphs, such as the Cartesian product and the line graph. A graph G is a finite non-empty set V of objects called vertices (the singular is vertex) together with a set E of two-element subsets of V called edges. The number of vertices in a graph G is the order of G , and the number of edges is the size of G . To indicate that a graph G has vertex set V and edge set E , we sometimes write $G = (V, E)$. To emphasize that V is the vertex set of a graph G , we often write V as $V(G)$. For the same reason, we also write E as $E(G)$. A graph H is said to be a subgraph of a graph G if $V(H) \subseteq V(G)$ and $E(H) \subseteq E(G)$. The complete graph with n -vertices will be denoted by K_n . A null graph is a graph containing no edges; the null graph with

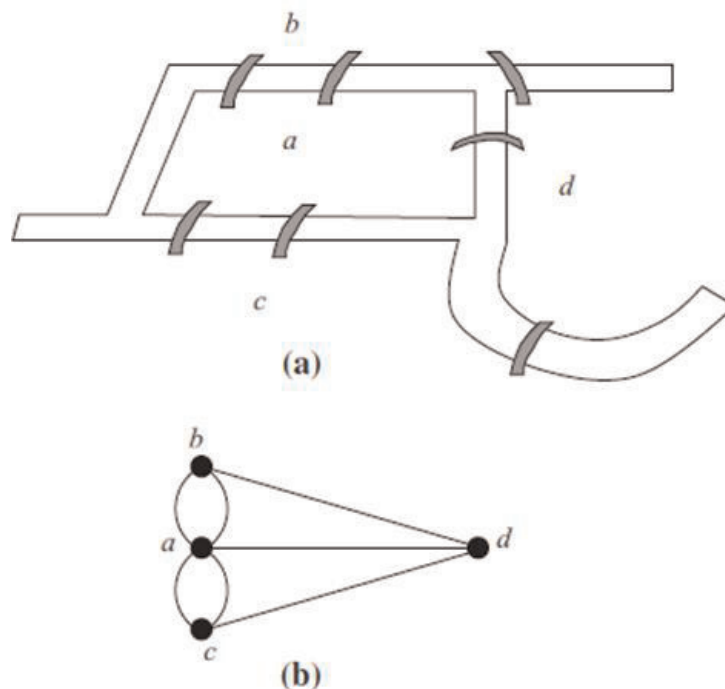


Figure 1.
Königsberg seven-bridge problem.

n -vertices is denoted by N_n . A cycle graph is a graph consisting of a single cycle, the cycle graph with n -vertices is denoted by C_n . The path graph is a graph consisting of a single path; the path graph with n -vertices is denoted by P_n [1–11]. Let G and H be two graphs. A function $\varphi : V(G) \rightarrow V(H)$ is a homomorphism from G to H if it preserves edges, that is, if for every edge $e \in E(G)$, $f(e) \in E(H)$ [12, 13]. A core is a graph which does not retract to a proper subgraph. Any graph is homomorphically equivalent to a unique core [7].

The folding is a continuous function $f : G \rightarrow H$ such that for each $v \in V(G)$, $f(v) \in V(H)$, and for each $e \in E(G)$, $f(e) \in E(H)$ [14]. Let X be a space, and let I be the unit interval $[0,1]$ in \mathbb{R} , a homotopy of paths in X is a family $g_t : I \rightarrow X$, $0 \leq t \leq 1$ such that (i) the endpoints $g_t(0) = x_0$ and $g_t(1) = x_1$ are independent of t and (ii) the associated map $G : I \times I \rightarrow X$ defined by $G(s,t) = g_t(s)$ is continuous [15]. Given spaces X and Y with chosen points $x_0 \in X$, and $y_0 \in Y$, the wedge sum $X \vee Y$ is the quotient of the disjoint union $X \cup Y$ obtained identifying x_0 and y_0 to a single point [15]. Two spaces X and Y are of the “same homotopy type” if there exist continuous maps $f : X \rightarrow Y$ and $g : Y \rightarrow X$ such that $g \circ f \cong I_X : X \rightarrow X$ and $f \circ g \cong I_Y : Y \rightarrow Y$ [16]. The fundamental group briefly consists of equivalence classes of homotopic closed paths with the law of composition following one path to another. However, the set of homotopy classes of loops based at the point x_0 with the product operation $[f][g] = [f \cdot g]$ is called the fundamental group and denoted by $\pi_1(X, x_0)$ [4, 17–24]. Over many years, chaos has been shown to be an interesting and even common phenomenon in nature. Chaos has been shown to exist in a wide variety of settings: in fluid dynamics such as Raleigh-Bernard convection, in chemistry such as the Belousov-Zhabotinsky reaction, in nonlinear optics in certain lasers, in celestial mechanics, in electronics in the flutter of an overdriven airplane wing, some models of population dynamics, and likely in meteorology, physiological oscillations such as certain heart rhythms, as well as brain patterns [17, 24–30]. AI algorithms related to adjacency matrices on the operations of the graph are discussed in [31, 32].

2. The main results

First, we will introduce the following:

Definition 1. The chaotic edge \bar{e} is a geometric edge e_1 that carries many other edges (e_2, e_3, \dots), each one of them homotopic to the original one as in **Figure 2**. Also the chaotic vertices of \bar{e} are $\bar{v} = (v_1, v_2, \dots)$ and $\bar{u} = (u_1, u_2, \dots)$. For chaotic edge \bar{e} , we have two cases:

Case 1 (1) e_1, e_2, e_3, \dots are of the same physical properties.

Case 2 (2) e_1, e_2, e_3, \dots represent different physical properties; for example, e_1 represents density, e_2 represents hardness, e_3 represents magnetic fields, and so on.

Definition 2. A chaotic graph \bar{G} is a collection of finite non-empty set \bar{V} of objects called chaotic vertices together with a set \bar{E} of two-element subsets of \bar{V} called chaotic edges. The number of chaotic edges is the size of \bar{G} .

Definition 3. Given chaotic connected graphs \bar{G}_1 and \bar{G}_2 with given edges $\bar{e}_1 \in \bar{G}_1$ and $\bar{e}_2 \in \bar{G}_2$, then the chaotic connected edge graph $\bar{G}_1 \bar{G}_2$ is the quotient of disjoint union $\bar{G}_1 \cup \bar{G}_2$ acquired by identifying two chaotic edges \bar{e}_1 and \bar{e}_2 to a single chaotic edge (up to chaotic isomorphism) as in **Figure 3**.

Definition 4. A chaotic graph \bar{H} is called a chaotic subgraph of a chaotic graph \bar{G} if $\bar{V}(\bar{H}) \subseteq \bar{V}(\bar{G})$ and $\bar{E}(\bar{H}) \subseteq \bar{E}(\bar{G})$.

Definition 5. Let \bar{G} and \bar{H} be two chaotic graphs. A function $\bar{\varphi} : \bar{V}(\bar{G}) \rightarrow \bar{V}(\bar{H})$ is chaotic homomorphism from \bar{G} to \bar{H} if it preserves chaotic edges, that is, if for any chaotic edge $[\bar{u}, \bar{v}]$ of \bar{G} , $[\bar{\varphi}(\bar{u}), \bar{\varphi}(\bar{v})]$ is a chaotic edge of \bar{H} .

Definition 6. A chaotic folding of a graph \bar{G} is a chaotic subgraph \bar{H} of \bar{G} such that there exists a chaotic homomorphism $\bar{f} : \bar{G} \rightarrow \bar{H}$, called chaotic folding with $\bar{f}(\bar{x}) = \bar{x}$ for every chaotic vertex \bar{x} of \bar{H} .

Definition 7. A chaotic core is a chaotic graph which does not chaotic retract to chaotic proper subgraph.

Theorem 1. Let \bar{G}_1 and \bar{G}_2 be two chaotic connected graphs. Then $\bar{\pi}_1(\bar{G}_1 \bar{G}_2) = \bar{\pi}_1(\bar{G}_1) * \bar{\pi}_1(\bar{G}_2)$.

Proof. Let \bar{G}_1 and \bar{G}_2 be two chaotic connected graphs. Since $\bar{G}_1 \bar{G}_2$ and $\bar{G}_1 \vee \bar{G}_2$ are of same chaotic homotopy type, it follows that $\bar{\pi}_1(\bar{G}_1 \bar{G}_2) \approx \bar{\pi}_1(\bar{G}_1) * \bar{\pi}_1(\bar{G}_2)$. Hence, $\bar{\pi}_1(\bar{G}_1 \bar{G}_2) = \bar{\pi}_1(\bar{G}_1) * \bar{\pi}_1(\bar{G}_2)$.

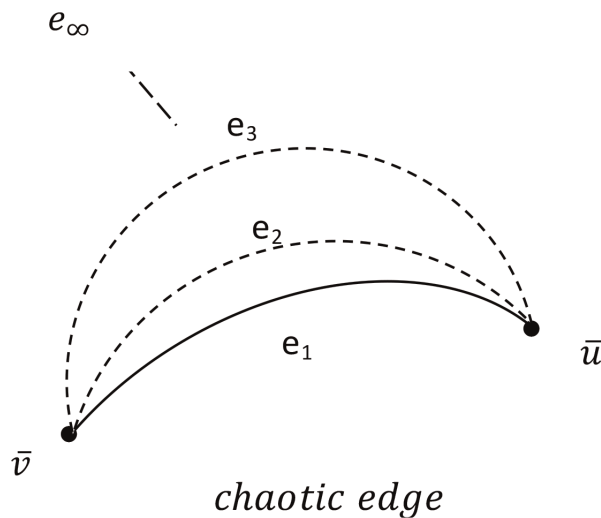


Figure 2.
Chaotic edge.

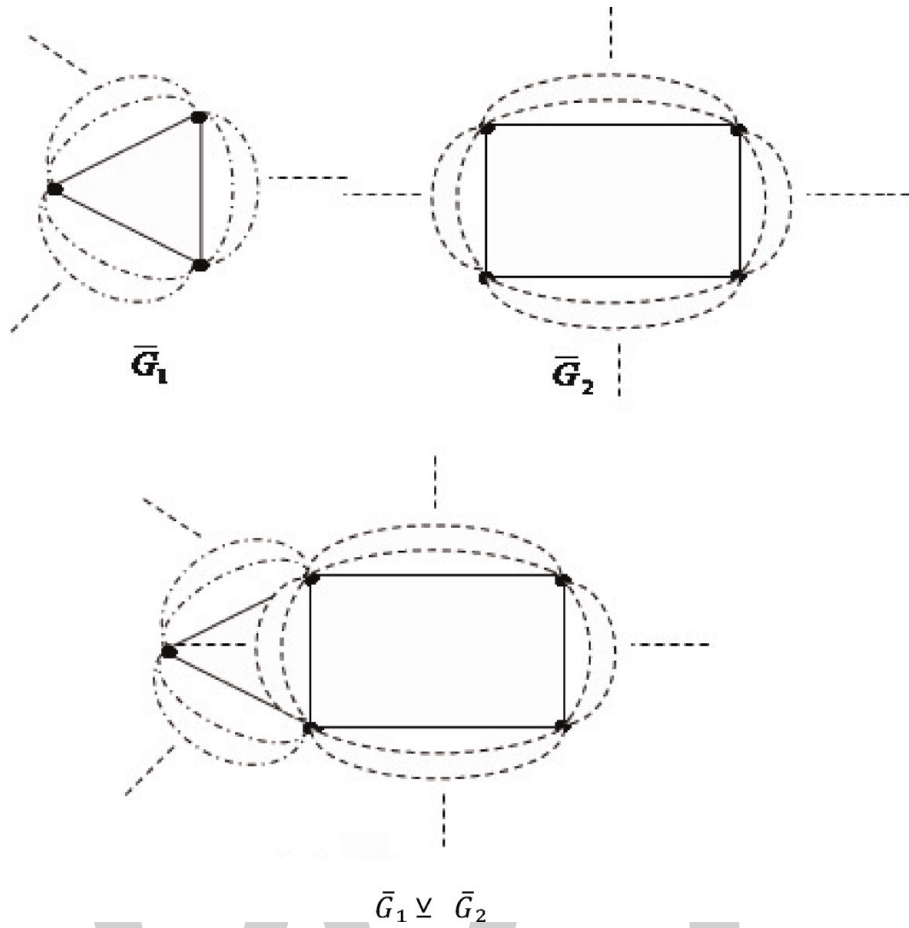


Figure 3.
Chaotic connected edge.

Theorem 2. The chaotic graphs \bar{G}_1 and \bar{G}_2 are chaotic subgraphs of $\bar{G}_1 \vee \bar{G}_2$. Also, for any chaotic tree \bar{G}_1 and \bar{G}_2 , $\bar{G}_1 \vee \bar{G}_2$ is also chaotic tree and $\bar{\pi}_1(\bar{G}_1 \vee \bar{G}_2) = \bar{0}$.

Proof. The proof of this theorem is clear.

Theorem 3. If $\bar{G}_1, \bar{G}_2, \dots, \bar{G}_n$ are connected graphs, and $\langle \bar{f}_1, \bar{f}_2, \dots, \bar{f}_n \rangle$ is a sequence of chaotic topological foldings of $\vee_{i=1}^n \bar{G}_i$ into itself, then there is an induced sequence $\langle \bar{f}_1, \bar{f}_2, \dots, \bar{f}_n \rangle$ of non-trivial chaotic topological folding $\bar{f}_j : *_{i=1}^n \bar{\pi}_1(\bar{G}_{ii}) \rightarrow *_{i=1}^n \bar{\pi}_1(\bar{G}_{ii})$, $j = 1, 2, \dots, n$ such that $\bar{f}_j(*_{i=1}^n \bar{\pi}_1(\bar{G}_{ii}))$ reduces the rank of $*_{i=1}^n \bar{\pi}_1(\bar{G}_{ii})$.

Proof. Consider the following sequence of topological foldings $\langle \bar{f}_1, \bar{f}_2, \dots, \bar{f}_n \rangle$, where $\bar{f}_1 : \vee_{i=1}^n \bar{G}_i \rightarrow \vee_{i=1}^n \bar{G}_i$, is a topological folding from $\vee_{i=1}^n \bar{G}_i$ into itself such that $\bar{f}_1(\vee_{i=1}^n \bar{G}_i) = \bar{G}_1 \vee \bar{G}_2 \vee \dots \vee \bar{f}_1(\bar{G}_s) \vee \dots \vee \bar{G}_n$ for $s = 1, 2, \dots, n$.

Since $size(\bar{f}_1(\bar{G}_s)) \leq size(\bar{G}_s)$ and $\bar{f}_1(\bar{\pi}_1(\bar{G}_i)) = \bar{\pi}_1(\bar{f}_1(\bar{G}_i))$, it follows that $rank(\bar{f}_1(\bar{\pi}_1(\bar{G}_s))) = rank(\bar{\pi}_1(\bar{f}_1(\bar{G}_s))) \leq rank(\bar{\pi}_1(\bar{G}_s))$, and so \bar{f}_1 reduces the rank of $*_{i=1}^n \bar{\pi}_1(\bar{G}_{ii})$. Also, if $\bar{f}_2(\vee_{i=1}^n \bar{G}_i) = \bar{G}_1 \vee \bar{G}_2 \vee \dots \vee \bar{f}_2(\bar{G}_s) \vee \dots \vee \bar{f}_2(\bar{G}_k) \vee \dots \vee \bar{G}_n$ for $k = 1, 2, \dots, n$ and $s < k$ and $size(\bar{f}_2(\bar{G}_s)) \leq size(\bar{G}_s)$ and $size(\bar{f}_2(\bar{G}_k)) \leq size(\bar{G}_k)$, we have $rank(\bar{f}_2(\bar{\pi}_1(\bar{G}_s))) = rank(\bar{\pi}_1(\bar{f}_2(\bar{G}_s))) \leq rank(\bar{\pi}_1(\bar{G}_s))$, $rank(\bar{f}_2(\bar{\pi}_1(\bar{G}_k))) = rank(\bar{\pi}_1(\bar{f}_2(\bar{G}_k))) \leq rank(\bar{\pi}_1(\bar{G}_k))$; thus \bar{f}_2 reduces the rank of $*_{i=1}^n \bar{\pi}_1(\bar{G}_{ii})$. Moreover, by continuing with this procedure if $\bar{f}_n(\vee_{i=1}^n \bar{G}_i) = \vee_{i=1}^n (\bar{f}_n(\bar{G}_i))$, then $\bar{f}_n(*_{i=1}^n \bar{\pi}_1(\bar{G}_{ii})) = \bar{\pi}_1(\bar{f}_n(\vee_{i=1}^n \bar{G}_i)) = \bar{\pi}_1(\vee_{i=1}^n \bar{f}_n(\bar{G}_i)) \approx *_{i=1}^n \bar{\pi}_1(\bar{f}_n(\bar{G}_{ii}))$. Hence, \bar{f}_n reduces the rank of $*_{i=1}^n \bar{\pi}_1(\bar{G}_{ii})$.

Theorem 4. Let \overline{G}_1 and \overline{G}_2 be two chaotic graphs; then there is a chaotic homomorphism $\overline{f} : \overline{G}_1 \rightarrow \overline{G}_2$ which induces $\overline{\overline{f}} : \overline{\pi}_1(\overline{G}_1) \rightarrow \overline{\pi}_1(\overline{G}_2)$ if $\overline{\pi}_1(\overline{G}_2)$ is a chaotic folding of $\overline{\pi}_1(\overline{G}_1 \vee \overline{G}_2)$.

Proof. Let $\overline{f} : \overline{G}_1 \rightarrow \overline{G}_2$ be a chaotic homomorphism. Since \overline{G}_2 is chaotic subgraph of $\overline{G}_1 \vee \overline{G}_2$, then there exists a chaotic homomorphism $\overline{f} : \overline{G}_1 \vee \overline{G}_2 \rightarrow \overline{G}_2$ with $\overline{f}(\overline{v}) = \overline{v}$ for any chaotic vertex \overline{v} of \overline{G}_2 which induces $\overline{\overline{f}} : \overline{\pi}_1(\overline{G}_1) \rightarrow \overline{\pi}_1(\overline{G}_2)$. What follows from \overline{G}_2 is a chaotic folding of $\overline{G}_1 \vee \overline{G}_2$ in that $\overline{\pi}_1(\overline{G}_2)$ is a chaotic folding of $\overline{\pi}_1(\overline{G}_1 \vee \overline{G}_2)$. Conversely, assume that \overline{G}_2 is a chaotic folding of $\overline{G}_1 \vee \overline{G}_2$; thus $\overline{f} : \overline{G}_1 \vee \overline{G}_2 \rightarrow \overline{G}_2$ is a chaotic homomorphism with $\overline{f}(\overline{v}) = \overline{v}$ for any chaotic vertex \overline{v} of \overline{G}_2 , and so there is a chaotic homomorphism $\overline{f} : \overline{G}_1 \rightarrow \overline{G}_2$ which induce $\overline{\overline{f}} : \overline{\pi}_1(\overline{G}_1) \rightarrow \overline{\pi}_1(\overline{G}_2)$.

Theorem 5. For any chaotic path graphs $\overline{P}_n, \overline{P}_m, n, m \geq 2$, there is a sequence of topological foldings with variation curvature $\{\overline{f}_i : i = 1, 2, \dots, k\}$ on $\overline{P}_n \vee \overline{P}_m$ which induce a sequence of topological foldings $\{\overline{\overline{f}}_i : i = 1, 2, \dots, k\}$ such that $\overline{\overline{f}}_k(\overline{\overline{f}}_{k-1}(\dots(\overline{\overline{f}}_1(\pi_1(\overline{P}_n \vee \overline{P}_m))\dots))) = \overline{Z}$ and $\lim_{k \rightarrow \infty} (\overline{\overline{f}}_k(\overline{\overline{f}}_{k-1}(\dots(\overline{\overline{f}}_1(\pi_1(\overline{P}_n \vee \overline{P}_m))\dots)))) = \overline{0}$.

Proof. Consider the following sequence of chaotic topological foldings with variation curvature, $\overline{f}_1 : \overline{P}_n \vee \overline{P}_m \rightarrow (\overline{P}_n \vee \overline{P}_m)_1$, where $(\overline{P}_n \vee \overline{P}_m)_1$ is a chaotic subgraph with decreasing inner curvature between every two adjacent chaotic edges in $\overline{P}_n \vee \overline{P}_m$ and $\overline{f}_2 : \overline{f}_1(\overline{P}_n \vee \overline{P}_m) \rightarrow \overline{f}_1((\overline{P}_n \vee \overline{P}_m)_1)$ where $\overline{f}_2(\overline{f}_1((\overline{P}_n \vee \overline{P}_m)_1))$ is a chaotic subgraph with decreasing inner curvature between every two adjacent chaotic edges in $\overline{f}_1((\overline{P}_n \vee \overline{P}_m)_1)$, and so on, such that $\overline{f}_k(\overline{f}_{k-1}(\overline{f}_{k-2}(\dots(\overline{f}_1(\overline{P}_n \vee \overline{P}_m))\dots))) = \overline{C}_{n+m-2}$ and $\lim_{k \rightarrow \infty} (\overline{f}_k(\overline{f}_{k-1}(\overline{f}_{k-2}(\dots(\overline{f}_1(\overline{P}_n \vee \overline{P}_m))\dots)))) = \overline{N}_1$, thus $\overline{\overline{f}}_k(\overline{\overline{f}}_{k-1}(\overline{\overline{f}}_{k-2}(\dots(\overline{\overline{f}}_1(\pi_1(\overline{P}_n \vee \overline{P}_m))\dots)))) = \overline{\pi}_{11}(\overline{C}_{n+m-2}) = \overline{Z}$. Also, $\lim_{k \rightarrow \infty} (\overline{\overline{f}}_k(\overline{\overline{f}}_{k-1}(\overline{\overline{f}}_{k-2}(\dots(\overline{\overline{f}}_1(\pi_1(\overline{P}_n \vee \overline{P}_m))\dots)))) = \overline{\pi}_1(\overline{N}_1) = \overline{0}$.

Theorem 6. For every two chaotic connected graphs \overline{G}_1 and \overline{G}_2 , the fundamental group of the limit of chaotic topological folding of $\overline{G}_1 \vee \overline{G}_2 = \overline{0}$.

Proof. Let \overline{G}_1 and \overline{G}_2 be two chaotic connected graphs; then we have two cases:

Case (1): If $\overline{f}_1 : \overline{G}_1 \vee \overline{G}_2 \rightarrow \overline{G}_1 \vee \overline{G}_2$ is a chaotic topological folding such that $\overline{f}_1(\overline{G}_1 \vee \overline{G}_2)$ consists of chaotic cycles, so we can define a sequence of chaotic topological folding $\overline{f}_2 : \overline{f}_1(\overline{G}_1 \vee \overline{G}_2) \rightarrow \overline{f}_1(\overline{G}_1 \vee \overline{G}_2)$ where $\overline{f}_2(\overline{f}_1(\overline{G}_1 \vee \overline{G}_2))$ is a chaotic tree with \overline{n} chaotic edges, $\overline{f}_3 : \overline{f}_2(\overline{f}_1(\overline{G}_1 \vee \overline{G}_2)) \rightarrow \overline{f}_2(\overline{f}_1(\overline{G}_1 \vee \overline{G}_2))$, such that $\overline{f}_3(\overline{f}_2(\overline{f}_1(\overline{G}_1 \vee \overline{G}_2)))$ is a chaotic tree with $\overline{k} < \overline{n}$ chaotic edges, chaotic edges by continuing this process we get $\overline{f}_k : \overline{f}_{k-1}(\overline{f}_{k-2}(\dots(\overline{f}_1(\overline{G}_1 \vee \overline{G}_2))\dots))) \rightarrow \overline{f}_{k-1}(\overline{f}_{k-2}(\dots(\overline{f}_1(\overline{G}_1 \vee \overline{G}_2))\dots)))$ such that $\lim_{k \rightarrow \infty} (\overline{f}_k(\overline{f}_{k-1}(\overline{f}_{k-2}(\dots(\overline{f}_1(\overline{G}_1 \vee \overline{G}_2))\dots))))$ is a chaotic edge, and so $\overline{\pi}_1(\lim_{k \rightarrow \infty} (\overline{f}_k(\overline{f}_{k-1}(\overline{f}_{k-2}(\dots(\overline{f}_1(\overline{G}_1 \vee \overline{G}_2))\dots)))) = \overline{0}$.

Case (2): If $\overline{g}_1 : \overline{G}_1 \vee \overline{G}_2 \rightarrow \overline{G}_1 \vee \overline{G}_2$ is a chaotic topological folding such that $\overline{g}_1(\overline{G}_1 \vee \overline{G}_2)$ has no chaotic cycles, then clearly $\lim_{k \rightarrow \infty} (\overline{g}_k(\overline{g}_{k-1}(\overline{g}_{k-2}(\dots(\overline{g}_1(\overline{G}_1 \vee \overline{G}_2))\dots))))$ is a chaotic edge and $\overline{\pi}_1(\lim_{k \rightarrow \infty} (\overline{g}_k(\overline{g}_{k-1}(\overline{g}_{k-2}(\dots(\overline{g}_1(\overline{G}_1 \vee \overline{G}_2))\dots)))) = \overline{0}$.

Theorem 7. If \overline{G}_1 and \overline{G}_2 are chaotic connected and not chaotic cores graphs, then $\overline{\pi}_1(\lim_{n \rightarrow \infty} \overline{f}_n(\overline{G}_1 \vee \overline{G}_2)) = \overline{\pi}_1(\lim_{n \rightarrow \infty} \overline{f}_n(\overline{G}_1)) * \overline{\pi}_1(\lim_{n \rightarrow \infty} \overline{f}_n(\overline{G}_2))$.

Proof. If \bar{G}_1 and \bar{G}_2 are chaotic connected and not chaotic cores graphs, then we get the following chaotic induced graphs $\lim_{n \rightarrow \infty} \bar{f}_n(\bar{G}_1 \vee \bar{G}_2)$, $\lim_{n \rightarrow \infty} \bar{f}_n(\bar{G}_1)$, $\lim_{n \rightarrow \infty} \bar{f}_n(\bar{G}_2)$, and each of them are isomorphic to \bar{k}_2 . Since $\bar{k}_2 \approx \bar{k}_2 \vee \bar{k}_2$ it follows that $\lim_{n \rightarrow \infty} \bar{f}_n(\bar{G}_1 \vee \bar{G}_2) = \lim_{n \rightarrow \infty} \bar{f}_n(\bar{G}_1) \vee \lim_{n \rightarrow \infty} \bar{f}_n(\bar{G}_2)$ and $\bar{\pi}_1(\lim_{n \rightarrow \infty} \bar{f}_n(\bar{G}_1 \vee \bar{G}_2)) = \bar{\pi}_1(\lim_{n \rightarrow \infty} \bar{f}_n(\bar{G}_1)) * \bar{\pi}_1(\lim_{n \rightarrow \infty} \bar{f}_n(\bar{G}_2))$.

3. Some applications

- i. A polymer is composed of many repeating units called monomers. Starch, cellulose, and proteins are natural polymers. Nylon and polyethylene are synthetic polymers. Polymerization is the process of joining monomers. Polymers may be formed by addition polymerization; furthermore, one essential advance likewise polymerization is mix as in **Figure 4**, which happens when the polymer's development is halted by free electrons from two developing chains that join and frame a solitary chain. The accompanying chart portrays mix, with the image (R) speaking to whatever remains of the chain.
- ii. Chemical nature of enzymes, all known catalysts are proteins. They are high atomic weight mixes made up primarily of chains of amino acids connected together by peptide bonds as in **Figure 5**.

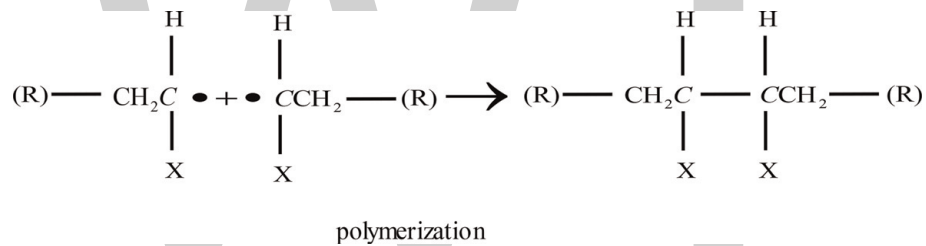


Figure 4.
Polymerization.

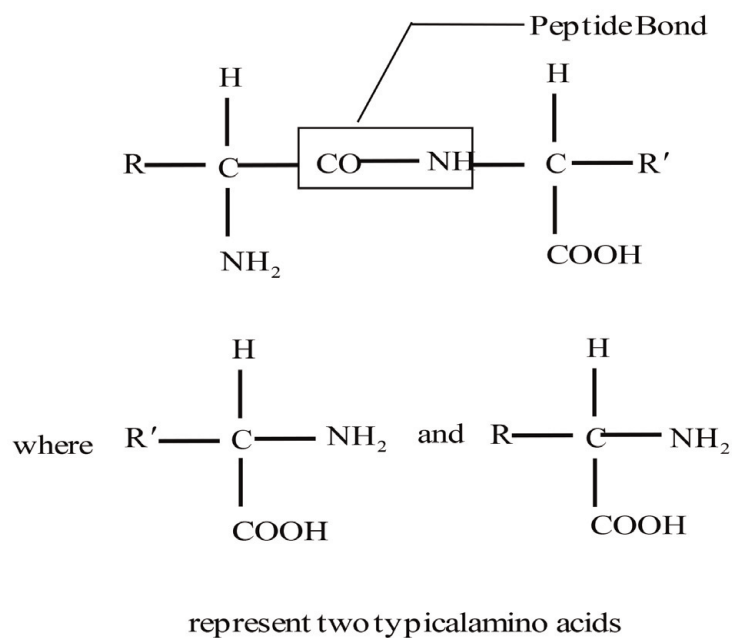
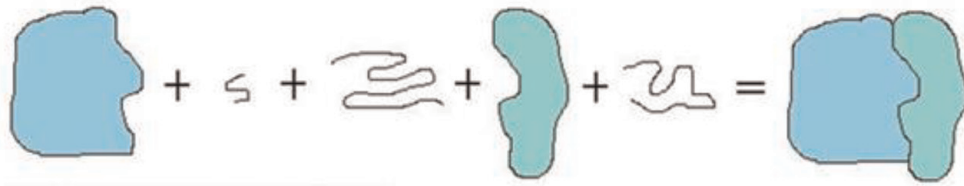


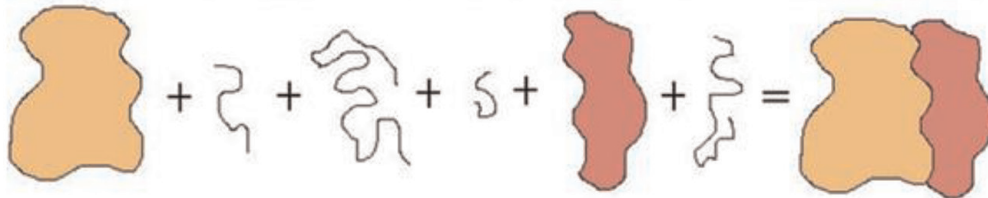
Figure 5.
Typical amino acids.

50S Protein Sununit+5S rRNA+23SrRNA+30S protein subunit+16S rRNA =70S ribosome



Eukaryotic Ribosome Components

60S Protein Subunit+5.6S rRNA++28srRNA+5SrRNA+40S Protein Sununit+18SrRNA=80S



Prokaryotic Ribosome Components

Figure 6.

Prokaryotic ribosome components.

- iii. There are two types of the subunit structure of ribosomes as in **Figure 6** which is represented by the different connected types of protein subunit and rRNA to form a new type of ribosomes.

4. Conclusion

In this chapter, the fundamental group of the limit chaotic foldings on chaotic connected edge graphs is deduced. Also, we can deduce some algorithms from a new operation of a graph by using the adjacency matrices.

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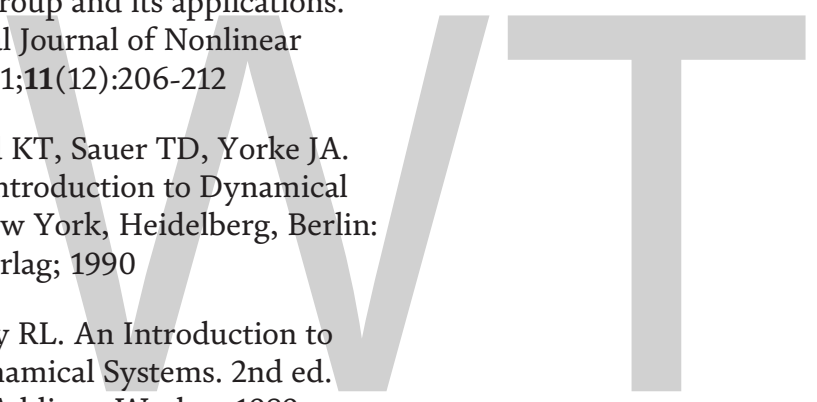
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New Matrix Series Formulae for Matrix Exponentials and for the Solution of Linear Systems of Algebraic Equations

Ioan R. Ciric

Abstract

The solution of certain differential equations is expressed using a special type of matrix series and is directly related to the solution of general systems of algebraic equations. Efficient formulae for matrix exponentials are derived in terms of rapidly convergent series of the same type. They are essential for two new solution methods, especially beneficial for large linear systems, namely an iterative method and a method based on an exact matrix product formula. The computational complexity of these two methods is analysed, and for both of them, the number of matrix exponential-vector multiplications required for an imposed accuracy can be predetermined in terms of the system condition. The total number of arithmetic operations involved is roughly proportional to n^2 , where n is the matrix dimension. The common feature of all the series in the results presented is that starting with a first term that is already well-conditioned, each subsequent term is computed by multiplication with an even better conditioned matrix, tending quickly to the identity matrix. This contributes substantially to the stability of the numerical computation. A very efficient method based on the numerical integration of a special kind of differential equations, applicable to even ill-conditioned systems, is also presented.

Keywords: matrix equations, matrix exponentials, numerical solutions

1. Introduction

New matrix series expressions were recently derived by the author [1] for the solution of simple first order differential equations associated with general systems of linear algebraic equations. These differential equations describe the orthogonal trajectories of a family of hypersurfaces that represent a quadratic functional related to the linear algebraic system. The solution of the latter can be obtained by minimizing the functional along an orthogonal trajectory instead of applying various techniques based on minimization along conjugate gradient directions or based on minimized iterations [2]. Since the solutions of the differential equations considered are simply related to the solutions of the corresponding algebraic systems through matrix exponentials, there is the possibility to develop efficient solution methods if only the matrix exponentials could be used in numerical calculations

accurately and with a small computational effort. A survey of various existent algorithms for computing matrix exponentials and a useful discussion of the difficulties involved are presented in [3].

In the present work, we use new formulae for arbitrary matrix exponentials that contain highly convergent infinite series which allow accurate and stable numerical computations. Employing these formulae, two new solution methods are proposed which are particularly efficient for large-scale general linear algebraic systems.

2. Differential equations associated with linear systems of algebraic equations

We start with simple vector differential equations whose solutions are related to the solution of general systems of linear algebraic equations. Later, in Section 5 we construct differential equations that allow an efficient numerical integration in order to obtain the solution of these systems.

2.1 Matrix series solution of some vector differential equations

Consider a first order vector differential equation of the form

$$\frac{dx}{dv} = f(v)(Ax - b) \tag{1}$$

where $A \in R^{n \times n}$ is a general nonsingular matrix, $b \in R^n$ is a given n -dimensional vector, $x = (x_1, x_2, \dots, x_n)^T$ is the unknown n -dimensional vector, T indicates the transpose, and $f(v)$ is a continuous function over some interval of the real variable v . With the condition

$$x(v_o) = x_o \tag{2}$$

x_o being a given vector, (1) has a unique solution over the interval considered [4]. Integrating both sides of (1) from v_o to v yields

$$Ax(v) - b = e^{A[g(v)-g(v_o)]}(Ax_o - b) \tag{3}$$

where

$$g(v) \equiv \int f(v) dv \tag{4}$$

is a primitive for $f(v)$, i.e., $f(v) = dg/dv$. Thus, (1) can be written in the form

$$\frac{dx}{dv} = f(v)e^{A[g(v)-g(v_o)]}(Ax_o - b) \tag{5}$$

Integrating again both sides from v_o to v gives

$$x(v) = x_o + e^{-Ag(v_o)} \sum_{k=0}^{\infty} \frac{A^k}{(k+1)!} [(g(v))^{k+1} - (g(v_o))^{k+1}] (Ax_o - b) \tag{6}$$

Two particular cases are presented.

If $f(v)$ is chosen to be $f(v) = -1$ then $g(v) = -v$. Choosing $v_o = 0$ gives $g(v_o) = 0$ and (6) becomes

$$x(v) = e^{-Av}x_o + v \sum_{k=0}^{\infty} (-1)^k \frac{(Av)^k}{(k+1)!} b \quad (7)$$

If $f(v) = 1/v$ then $g(v) = \ln v$. With $v_o = 1$ and $g(v_o) = 0$ (6) becomes now

$$x(v) = x_o + (\ln v) \sum_{k=0}^{\infty} \frac{(A \ln v)^k}{(k+1)!} (Ax_o - b) \quad (8)$$

The solution of (1) with $f(v) = 1/v$ can also be obtained by employing a Taylor series expansion about $x_o = 1$. Indeed,

$$x(v) = x(1) + \frac{(v-1)}{1!} x^{(1)}(1) + \frac{(v-1)^2}{2!} x^{(2)}(1) + \dots + \frac{(v-1)^k}{k!} x^{(k)}(1) + \dots \quad (9)$$

and evaluating the derivatives yields

$$x(v) = x_o - (1-v) \left[I + \sum_{k=1}^{\infty} \frac{(1-v)^k}{k+1} (I-A) \left(I - \frac{A}{2} \right) \dots \left(I - \frac{A}{k} \right) \right] (Ax_o - b) \quad (10)$$

where I is the identity matrix. The series in (10) is convergent for $v \in (0, 2)$ but its rate of convergence is, in general, very small for v very close to zero.

2.2 Relationship with linear systems of algebraic equations

Consider now a system of equations written in matrix form as

$$Ax = b \quad (11)$$

and assume that x is a continuous function of the real variable v over a certain interval. The solution of (11) can be obtained from (3) for a $v \equiv v_s$ for which

$$Ax(v_s) - b = 0 \quad (12)$$

To satisfy this condition $g(v)$ in (3) must be chosen such that

$$e^{A[g(v_s) - g(v_o)]} = 0 \quad (13)$$

For positive definite matrices A and a finite $g(v_o)$ this is achieved if $g(v) \rightarrow -\infty$ for $v \rightarrow v_s$. Thus, the solution of (11) cannot be computed directly from (6). On the other hand, in the particular case of $f(v) = 1/v, g(v) = \ln v$ the solution of (11) can be obtained from (10) with $v = v_s = 0$,

$$x_s = x_o - \left[I + \sum_{k=1}^{\infty} \frac{1}{k+1} (I-A) \left(I - \frac{A}{2} \right) \dots \left(I - \frac{A}{k} \right) \right] (Ax_o - b) \quad (14)$$

The expression in the brackets is just the inverse of A ,

$$A^{-1} = I + \sum_{k=1}^{\infty} \frac{1}{k+1} (I - A) \left(I - \frac{A}{2} \right) \cdots \left(I - \frac{A}{k} \right) \quad (15)$$

The rate of convergence of the series in (14) and (15) is very small and they cannot be practically used in numerical computation for arbitrary matrices.

We note that the solution of (11) can be formally expressed from (6) as

$$x_s = x_o + e^{Ag(v_o)} \left(e^{Ag(v_o)} - e^{Ag(v)} \right)^{-1} (x(v) - x_o) \quad (16)$$

which is valid for any $v \neq v_o$ in the interval considered. To compute x_s with (16) would require the computation of $(e^{Ag(v_o)} - e^{Ag(v)})^{-1}$. Equation (16) will be used in Subsection 4.1.

Note. The matrix product in the terms of the series in (10), (14) and (15) can be expressed as a matrix polynomial using the relationship with the Stirling numbers of the first kind $S_{k+1}^{(m)}$ [5]

$$(I - A) \left(I - \frac{A}{2} \right) \cdots \left(I - \frac{A}{k} \right) = (-1)^k \frac{1}{k!} \sum_{m=1}^{k+1} S_{k+1}^{(m)} A^{m-1} \quad (17)$$

While each new term in such series as those in (10), (14) and (15) is calculated through a multiplication with a matrix that becomes more and more well-conditioned as k increases, the computation with the expression in (17) would require successive multiplications with the same original matrix and, for each k , a new polynomial is to be constructed and new Stirling numbers have to be generated. The formulae derived in the next two sections contain the same type of series and, therefore, are simpler and more efficient to be used for numerical computations.

3. New formulae for computing matrix exponentials

In this section, we derive matrix exponential expressions which contain highly convergent infinite series that allow accurate and stable numerical computations in numerous applications. They shall also be used in the next section.

3.1 Series expressions for matrix exponentials

Consider the matrix function $v^A \equiv e^{A \ln v}$ where v is a positive real variable and A a general square matrix with real number entries. By integration,

$$\int_1^v v^{A-I} dv = \frac{v^A}{A} \Big|_{v=1}^v = \frac{1}{A} (e^{A \ln v} - I) \quad (18)$$

For $v \in (0, 2)$ the integrand can be expanded in a power series as

$$\begin{aligned} v^{A-I} &= [1 - (1-v)]^{A-I} = I - \frac{(1-v)}{1!} (A-I) + \frac{(1-v)^2}{2!} (A-I)(A-2I) \\ &\quad - \frac{(1-v)^3}{3!} (A-I)(A-2I)(A-3I) + \cdots \end{aligned} \quad (19)$$

that can be integrated term by term. From (18) and (19)

$$e^{A \ln v} = I - (1-v)A \left[I + \sum_{k=1}^{\infty} \frac{(1-v)^k}{k+1} (I-A) \left(I - \frac{A}{2} \right) \cdots \left(I - \frac{A}{k} \right) \right] \quad (20)$$

which is valid for any A , positive definite or not, of arbitrary condition. One can see that, for a positive definite A and for $v \rightarrow 0$, the expression in the brackets of (20) gives a series expansion for the inverse A^{-1} (see (14) and (15)). Various expressions for the matrix exponential are obtained by giving particular values to v in (20). For example, for any $v = e^{-q}$, $q > -\ln 2$, we have

$$e^{-qA} = I - (1 - e^{-q})A \left[I + \sum_{k=1}^{\infty} \frac{(1 - e^{-q})^k}{k+1} (I - A) \left(I - \frac{A}{2} \right) \cdots \left(I - \frac{A}{k} \right) \right] \quad (21)$$

the series becoming less convergent as q increases above $q = 0$. On the other hand, with $v = 1/e$ in (20) and then A replaced with qA we obtain for any real number q

$$e^{-qA} = I - (1 - e^{-1})qA \left[I + \sum_{k=1}^{\infty} \frac{(1 - e^{-1})^k}{k+1} (I - qA) \left(I - \frac{qA}{2} \right) \cdots \left(I - \frac{qA}{k} \right) \right] \quad (22)$$

this series being more convergent than that in (21) for greater values of q .

For any $v \in (1, 2)$, the terms in the series expressions derived from (20) have coefficients that are alternately positive and negative. With $v = e^{1/2} = 1.64872127$, e.g., and then replacing A with $2A$ we have

$$\left| e^A = I + 2(e^{1/2} - 1)A \left[I + \sum_{k=1}^{\infty} \frac{(-1)^k (e^{1/2} - 1)^k}{k+1} (I - 2A) \left(I - \frac{2A}{2} \right) \cdots \left(I - \frac{2A}{k} \right) \right] \right| \quad (23)$$

As well, by replacing A by $-qA$ in (23) one obtains instead of (22) an expression with alternating in sign series coefficients.

3.2 Rapidly convergent series formulae

From the basic Eq. (20) we derive now formulae which contain series that have a higher rate of convergence than those presented in the previous subsection.

Firstly, it is obvious that for values of v close to 1 the series in (20) has a high rate of convergence. For instance, with $v = 1 + 10^{-q}$, $10^{-q} \ll 1$, and replacing $A \ln(1 + 10^{-q})$ by A we obtain

$$e^A = I + 10^{-q} c_q A \left[I + \sum_{k=1}^{\infty} \frac{(-1)^k 10^{-qk}}{k+1} (I - c_q A) \left(I - \frac{c_q A}{2} \right) \cdots \left(I - \frac{c_q A}{k} \right) \right] \quad (24)$$

where $c_q \equiv 1/\ln(1 + 10^{-q})$. This series is rapidly convergent.

Secondly, the convergence of the series in the expressions derived from (20) can further be improved by successive integrations. Indeed, integrating both sides of (20) from 1 to v , we have

$$ve^{A \ln v} = I + (1-v)(I+A) \left[-I + \frac{1-v}{2} A + A \sum_{k=1}^{\infty} \frac{(1-v)^{k+1}}{(k+1)(k+2)} (I-A) \left(I - \frac{A}{2} \right) \cdots \left(I - \frac{A}{k} \right) \right] \quad (25)$$

Integrating repeatedly we obtain the identity

$$\begin{aligned}
 v^p e^{A \ln v} &= I + p! \sum_{k=1}^p \frac{(-1)^k (1-v)^k}{k!(p-k)!} \left(I + \frac{A}{p} \right) \left(I + \frac{A}{p-1} \right) \cdots \left(I + \frac{A}{p-k+1} \right) \\
 &+ (-1)^{p+1} (1-v)^{p+1} A(I+A) \left(I + \frac{A}{2} \right) \cdots \left(I + \frac{A}{p} \right) \left[\frac{1}{p+1} I \right. \\
 &\left. + p! \sum_{k=1}^{\infty} \frac{(1-v)^k}{(k+1)(k+2) \cdots (k+p+1)} (I-A) \left(I - \frac{A}{2} \right) \cdots \left(I - \frac{A}{k} \right) \right], \\
 & \qquad \qquad \qquad p = 1, 2, \dots; \quad v \in (0, 2)
 \end{aligned} \tag{26}$$

Same result is obtained by replacing A with $pI + A$ in (20). This identity contains an infinite series whose coefficients decrease rapidly as p increases. Obviously, for a given A and v , (26) generates more efficient computational formulae than those in the previous subsection. As before, for $v \in (1, 2)$ the infinite series have coefficients that alternate in sign. For example, with $v = e^{1/2}$ in (26) and then replacing A by $2A$ and p by $2p$ we have instead of (23)

$$\begin{aligned}
 e^A &= e^{-p} \left\{ I + (2p)! \sum_{k=1}^{2p} \frac{(e^{1/2} - 1)^k}{k!(2p-k)!} \left(I + \frac{2A}{2p} \right) \left(I + \frac{2A}{2p-1} \right) \cdots \left(I + \frac{2A}{2p-k+1} \right) \right. \\
 &+ 2(e^{1/2} - 1)^{2p+1} A(I+2A) \left(I + \frac{2A}{2} \right) \cdots \left(I + \frac{2A}{2p} \right) \left[\frac{1}{2p+1} I \right. \\
 &\left. + (2p)! \sum_{k=1}^{\infty} \frac{(-1)^k (e^{1/2} - 1)^k}{(k+1)(k+2) \cdots (k+2p+1)} (I-2A) \left(I - \frac{2A}{2} \right) \cdots \left(I - \frac{2A}{k} \right) \right] \Bigg\}, \\
 & \qquad \qquad \qquad p = 1, 2, \dots
 \end{aligned} \tag{27}$$

Taking $v = 1 + 10^{-q}$, with $q > 0$ and $c_q \equiv 1/\ln(1 + 10^{-q})$, (26) gives (compare with (24))

$$\begin{aligned}
 e^A &= (1 + 10^{-q})^{-p} \left\{ I + p! \sum_{k=1}^p \frac{10^{-qk}}{k!(p-k)!} \left(I + \frac{c_q A}{p} \right) \left(I + \frac{c_q A}{p-1} \right) \cdots \left(I + \frac{c_q A}{p-k+1} \right) \right. \\
 &+ 10^{-(p+1)q} c_q A(I+c_q A) \left(I + \frac{c_q A}{2} \right) \cdots \left(I + \frac{c_q A}{p} \right) \left[\frac{1}{p+1} I \right. \\
 &\left. + p! \sum_{k=1}^{\infty} \frac{(-1)^k 10^{-qk}}{(k+1)(k+2) \cdots (k+p+1)} (I-c_q A) \left(I - \frac{c_q A}{2} \right) \cdots \left(I - \frac{c_q A}{k} \right) \right] \Bigg\}, \\
 & \qquad \qquad \qquad p = 1, 2, \dots
 \end{aligned} \tag{28}$$

Notice that, in the new formulae derived from (26) the infinite series are very rapidly convergent, with their rate of convergence increasing when the parameter p increases. Highly accurate numerical results can be generated with only a small number of terms retained in the infinite series of these formulae (see Section 4).

Note. All the formulae presented in this section remain valid if A is changed in $-A$. Obviously, in all these expressions A can be replaced by a real number and the identity matrix I by 1, yielding a few novel identities and summation formulae for series of real numbers. Also, the expression in the brackets of (26) for $v \rightarrow 0$ is just $(I + A/p)^{-1}/p$ if $I + A/p$ is positive definite and, thus, we obtain another identity, i.e.,

$$\begin{aligned}
 & p! \sum_{k=1}^p \frac{(-1)^{k+1}}{k!(p-k)!} \left(I + \frac{A}{p} \right) \left(I + \frac{A}{p-1} \right) \cdots \left(I + \frac{A}{p-k+1} \right) \\
 & = I + \frac{(-1)^{p+1}}{p} A(I+A) \left(I + \frac{A}{2} \right) \cdots \left(I + \frac{A}{p-1} \right), \quad p=2,3,\dots
 \end{aligned} \tag{29}$$

which reduces for $A = 0$ to an elementary binomial sum.

4. Solution of general linear systems

In what follows, we apply the matrix exponential formulae from the previous section and present a new iteration procedure and a matrix product formula for the solution of large systems of linear algebraic equations.

4.1 An iterative method

Equation (16) can be written in the form

$$x(v) = x_o + \left(I - e^{A[g(v)-g(v_o)]} \right) (x_s - x_o) \tag{30}$$

where $x_s = x(v_s)$ is the solution of (11), $x_o = x(v_o)$, A is a positive definite matrix and $g(v)$ is a function of the real variable v such that $g(v) \rightarrow -\infty$ for $v \rightarrow v_s$ (see Subsection 2.2). To get $x(v)$ for values of v very close to v_s we choose an adequate $g(v)$ and a formula for the matrix exponential from Section 3. When $g(v_o) = 0$, applying (22) for instance gives

$$\begin{aligned}
 x(v) = x_o + (1 - e^{-1})g(v) & \left[I + \sum_{k=1}^{\infty} \frac{(1 - e^{-1})^k}{k+1} (I + Ag(v)) \right. \\
 & \left. \times \left(I + \frac{A}{2}g(v) \right) \cdots \left(I + \frac{A}{k}g(v) \right) \right] (Ax_o - b)
 \end{aligned} \tag{31}$$

If $g(v) \equiv \ln v$, with $v_o = 1$ and $v_s = 0$, we compute $x(e^{-N})$ for $N \gg 1$ which is closer to the solution x_s ,

$$\begin{aligned}
 x(e^{-N}) \equiv x_s^{(1)} = x_s^{(0)} - N(1 - e^{-1}) & \left[I + \sum_{k=1}^{\infty} \frac{(1 - e^{-1})^k}{k+1} (I - NA) \right. \\
 & \left. \times \left(I - \frac{N}{2}A \right) \cdots \left(I - \frac{N}{k}A \right) \right] (Ax_s^{(0)} - b)
 \end{aligned} \tag{32}$$

where $x_s^{(0)} \equiv x_o$. This equation is applied iteratively by replacing $x_s^{(1)}$ and $x_s^{(0)}$, respectively, with $x_s^{(i)}$ and $x_s^{(i-1)}$, $i = 2, 3, \dots$, until $x_s^{(i)}$ satisfies (11) with a desired accuracy.

To evaluate the amount of computation necessary to obtain the solution of (11) with a certain accuracy, let us take N such that $N\|A\| = 10$ when one needs about 30 terms in the infinite series, i.e., 30 matrix-vector multiplications. The number of iterations increases with the condition number of A . To see this and to determine the corresponding number of iterations, consider (11) with $b = 0$ and A replaced with a diagonal matrix whose entries are positive numbers, the greatest of these being 1, and whose condition is the same as that of A . The solution of this system is

$x_S = 0$ and the components of the solution of (1), with $f(v) = 1/v$, are $x_{ok}v^{\lambda_k}$ where $\lambda_k, k = 1, 2, \dots, n$, are the entries in the diagonal matrix. In order to make the magnitudes of all these components at least 1%, e.g., of the corresponding magnitudes of the initial components, one needs no iteration if the condition number is less than 2, but 5 and, respectively, 46 iterations are needed if the condition number is 10 and 100.

What is remarkable in the iterative method based on (32) is that, for matrices with same condition number and same norm, the number of iterations required is the same, independently of the size of the matrices. Considering approximately $2n^2$ arithmetic operations for one matrix-vector multiplication, where n is the number of equations and unknowns in (11), the total number of arithmetic operations required is, thus, proportional to only n^2 . In the examples given above one has to perform, respectively, $60n^2, 300n^2$ and $2760n^2$ arithmetic operations. Assuming only $2n^3/3$ arithmetic operations for the Gaussian elimination procedure, the method presented in this subsection requires less computation for the same examples if, respectively, $n > 90, n > 450$ and $n > 4140$. One can also notice that the application of Eq. (32) leads to the actual solution of (11) independently of the small error introduced in the computation at each iteration.

4.2 A matrix product formula

The original general system (11) is replaced with an equivalent system such that its solution is obtained in terms of matrix exponentials for which highly convergent and accurate series formulae have been derived in Section 3.

Namely, instead of (11) we use the system

$$(I - e^{-\alpha A})x = b_\alpha \tag{33}$$

where α is a real scalar to be chosen, $\alpha \neq 0$, and

$$b_\alpha = \alpha \left[1 - \frac{(\alpha A)}{2!} + \frac{(\alpha A)^2}{3!} - \frac{(\alpha A)^3}{4!} + \dots \right] b \tag{34}$$

Assuming A to be positive definite, α is taken positive. Then, since $\|e^{-\alpha A}\| < 1$ for a normal matrix, the solution can be expressed as

$$x_s = (I - e^{-\alpha A})^{-1} b_\alpha = \left[\sum_{k=0}^{\infty} e^{-k\alpha A} \right] b_\alpha \tag{35}$$

with the norm of the matrix exponentials decreasing when k increases [6]

$$\|e^{-k\alpha A}\| < e^{-k\alpha\lambda} \tag{36}$$

where λ is the smallest eigenvalue of A . b_α can be accurately computed by using instead of (34) an equivalent expression, for instance (see (22))

$$b_\alpha = (1 - e^{-1})\alpha \left[I + \sum_{k=1}^{\infty} \frac{(1 - e^{-1})^k}{k+1} (I - \alpha A) \left(I - \frac{\alpha A}{2} \right) \dots \left(I - \frac{\alpha A}{k} \right) \right] b \tag{37}$$

If the infinite series in (35) is truncated to $k = N_S$ the rest of the series has a norm

$$\|R_{N_S}\| < \frac{e^{-(N_S+1)\alpha\lambda}}{1 - e^{-\alpha\lambda}} \tag{38}$$

Much less numerical computation (see below) is needed if the infinite series in (35) is transformed into an infinite product using the identity [5]

$$(1-u)^{-1} = \prod_{k=0}^{\infty} (1+u^{2^k}), \quad 0 < u < 1 \tag{39}$$

which is also valid for matrices whose norm is less than 1. Thus, (35) becomes

$$x_s = \left[\prod_{k=0}^{\infty} (1 + e^{-2^k \alpha A}) \right] b_\alpha \tag{40}$$

with the norm of the exponentials $e^{-2^k \alpha A}$ decreasing very rapidly when k increases. Truncating the infinite product to $k = N_P$, i.e., $N_P + 1$ factors, leaves a remaining factor

$$F_{N_P} = \prod_{k=N_P+1}^{\infty} (I + e^{-2^k \alpha A}) \tag{41}$$

whose norm is

$$\|F_{N_P}\| < \prod_{k=N_P+1}^{\infty} (1 + e^{-2^k \alpha \lambda}) = \frac{1}{1 - e^{-\alpha \lambda}} \frac{1}{\prod_{k=0}^{N_P} (1 + e^{-2^k \alpha \lambda})} \tag{42}$$

Let us compare the maximum value of the norm of the truncated matrix in the brackets of (35) and (40) with that of the corresponding untruncated matrix in order to get a rough estimate of the numbers N_S and N_P of matrix exponentials involved in the numerical computation to achieve a certain accuracy. This will also allow to estimate the total number of matrix-vector multiplications necessary to obtain the solution. The ratio of the maximum norm of the truncated matrix to the maximum norm of the untruncated matrix in the brackets of (35) and (40) is, respectively, (see (38) and (42))

$$\rho = 1 - e^{-(N_S+1)\alpha\lambda} \tag{43}$$

and

$$\rho = (1 - e^{-\alpha\lambda}) \prod_{k=0}^{N_P} (1 + e^{-2^k \alpha \lambda}) \tag{44}$$

To illustrate the computation complexity when using (35) or (40), assume that $\|A\| = 1$ and $\alpha = 20$. If ρ is imposed to be ≈ 0.99 , for example, one needs $N_S = 2$, i.e., three terms in (35) and $N_P = 1$, i.e., two factors in (40) if $\lambda = 10^{-1}$. If $\lambda = 10^{-2}$ these numbers increase to $N_S = 23$ and $N_P = 4$, and when $\lambda = 10^{-3}$ one gets $N_S = 230$, but N_P only increases to $N_P = 7$. It is clear that applying the formula (40) the number of

exponentials needed in the numerical computation is much smaller than that if formula (35) would be applied. For all the matrix exponentials involved in the numerical computation we use the formula (28) containing a highly convergent series, such that

$$\begin{aligned}
 e^{-\alpha A} = & (1 + 10^{-q})^{-p} \left\{ I + p! \sum_{k=1}^p \frac{10^{-qk}}{k!(p-k)!} \left(I - \frac{c_q \alpha A}{p} \right) \left(I - \frac{c_q \alpha A}{p-1} \right) \cdots \left(I - \frac{c_q \alpha A}{p-k+1} \right) \right. \\
 & - 10^{-(p+1)q} c_q \alpha A (I - c_q \alpha A) \left(I - \frac{c_q \alpha A}{2} \right) \cdots \left(I - \frac{c_q \alpha A}{p} \right) \left[\frac{I}{p+1} \right. \\
 & \left. \left. + p! \sum_{k=1}^{\infty} \frac{(-1)^k 10^{-qk} k!}{(k+p+1)!} (I + c_q \alpha A) \left(I + \frac{c_q \alpha A}{2} \right) \cdots \left(I + \frac{c_q \alpha A}{k} \right) \right] \right\}, \\
 & p = 1, 2, \dots
 \end{aligned} \tag{45}$$

where $q > 0$ and $c_q \equiv 1 / \ln(1 + 10^{-q})$. With $\alpha \|A\| = 20$ and choosing $q = 1$ and $p = 10$, for instance, $e^{-\alpha A}$ is determined accurately by retaining 50 terms in the infinite series and, thus, to multiply $e^{-\alpha A}$ with a vector one needs 71 matrix-vector multiplications. To compute x_s from (40) one has to use repeatedly the multiplication of e^{-20A} with a vector. For a matrix A with $\lambda = 10^{-1}$ one has to retain two factors in (40) and, thus, to multiply e^{-20A} and $e^{-2 \times 20A}$ with a vector. This means to use repeatedly 3 times the multiplication of e^{-20A} with a vector which requires, therefore, $3 \times 71 = 213$ matrix-vector multiplications. When $\lambda = 10^{-2}$ the infinite product in (40) is truncated at $k = N_p = 4$ and this requires the multiplication of $e^{-2^k \times 20A}$, $k = 0, 1, 2, 3, 4$, with a vector, i.e., to use repeatedly 31 times the multiplication of e^{-20A} with a vector, for a total of $31 \times 71 = 2201$ matrix-vector multiplications. We also have to add the matrix-vector multiplications required to compute b_α in (37). A very accurate result for b_α when $\alpha = 20$ can be achieved by applying four times the series in the brackets of (37) for $\alpha = 5$, each time retaining 30 terms. This requires a total of about 120 multiplications of a matrix $I - 5A/k$ with a vector. In all the matrix-vector multiplications involved when applying (45), the matrices are in the form $I \pm c_q \alpha A/k$ and become better and better conditioned as k increases.

Adding up the number of arithmetic operations involved shows that, with respect to the classical Gaussian elimination method, the procedure presented in this subsection is advantageous for very large systems (11). Namely, assuming same accuracy and only $2n^3/3$ arithmetic operations for the Gaussian elimination, with the data given above, one has to have $n > 3 \times (213 + 120) = 999$ equations and unknowns if $\lambda = 10^{-1}$ and $n > 3 \times (2201 + 120) = 6963$ equations and unknowns if $\lambda = 10^{-2}$ for the proposed method to be more advantageous. For a given $\alpha \|A\|$, one application of $e^{-\alpha A}$ requires a determined finite number of matrix-vector multiplications, independently of the size of A . It is remarkable, as for the iterative method in the previous subsection, that for a given condition of A , one has to apply $e^{-\alpha A}$ a well-determined number of times and, thus, the total number of arithmetic operations necessary to compute the solution with an imposed accuracy is proportional to only n^2 .

It should be noted that, since the infinite series in the expression (45) is truncated and thus determined with a finite accuracy, the accuracy of the solution x_s becomes compromised after a too big a number of matrix exponential-vector multiplications. This is why, the worse conditioned systems (11) should be

appropriately preconditioned. Practically, the computation with (40) is continued factor by factor and the accuracy of x is checked after each step.

5. Solution of general linear systems by numerical integration of differential equations

In this section, we introduce first order differential equations whose numerical integration allows to efficiently find the solution of linear systems of algebraic equations. Differential equations of the type of those in (1), with $f(v) = -1$ or $f(v) = 1/v$, cannot be used for this purpose due to the fact that the first and higher order derivatives of $x(v)$ tend to infinite values as x tends to the solution x_S of (11) (see Section 2).

Here below, we construct ordinary differential equations for $x(v)$ which satisfy the condition that the first few derivatives are finite when $x(v)$ tends to x_S and, therefore, are particularly useful for an accurate computation of x_S . Let us consider the system (11) with a symmetric positive definite matrix. A quadratic functional

$$F(x) = \frac{1}{2}x^T Ax - x^T b + \frac{1}{2}b^T A^{-1}b \tag{46}$$

is associated with (11) [6] whose minimum value is $F(x_S) = 0$. Define now a real variable $v, v \geq 0$, such that

$$F(x) = v^r \tag{47}$$

where r is a real number to be chosen, $r > 0$, with $v = 0$ corresponding to the solution $x = x_S$ and $v = v_o$ to an initial point $x_o, F(x_o) = v_o^r$. Then,

$$\frac{dF}{dv} = rv^{r-1} = (Ax - b)^T \frac{dx}{dv} \tag{48}$$

and, thus,

$$\frac{dx}{dv} = rv^{r-1} \frac{Ax - b}{\|Ax - b\|^2} \tag{49}$$

This is the differential equation to be integrated from $v = v_o$ to $v = v_S = 0$. The second derivative of x is obtained in the form

$$\frac{d^2x}{dv^2} = r(r-1)v^{r-2} \frac{Ax - b}{\|Ax - b\|^2} + (rv^{r-1})^2 \frac{1}{\|Ax - b\|^4} \left\{ A - 2 \frac{[(Ax - b)^T A (Ax - b)]}{\|Ax - b\|^2} \right\} (Ax - b) \tag{50}$$

Higher order derivatives can be worked out if needed.

In order to see the behaviour of the derivatives close to the solution x_S , Eqs. (49) and (50) are rewritten as

$$\frac{dx}{dv} = r \frac{(F(x))^{1-\frac{1}{r}} (Ax - b)}{\|Ax - b\|^2} \tag{51}$$

and

$$\frac{d^2x}{dv^2} = r(r-1) \frac{(F(x))^{1-\frac{2}{r}} (Ax - b)}{\|Ax - b\|^2} + r^2 \frac{(F(x))^{2-\frac{2}{r}}}{\|Ax - b\|^4} \left\{ A - 2 \frac{[(Ax - b)^T A (Ax - b)]}{\|Ax - b\|^2} \right\} (Ax - b) \tag{52}$$

with $F(x)$ in (46) put in the form

$$F(x) = \frac{1}{2} (Ax - b)^T (x - x_s) \tag{53}$$

Notice that as x tends to x_s , when $\|Ax - b\| \equiv \varepsilon$ tends to zero,

$$\left\| \frac{dx}{dv} \right\| \sim K_1(v) \varepsilon^{1-\frac{1}{r}}, \quad \left\| \frac{d^2x}{dv^2} \right\| \sim K_2(v) \varepsilon^{1-\frac{4}{r}} \tag{54}$$

where $K_1(v)$ and $K_2(v)$ are finite when $v \rightarrow 0$. Therefore, as $x \rightarrow x_s$, $\|dx/dv\| \rightarrow 0$ if $r > 2$ and $\|d^2x/dv^2\| \rightarrow 0$ if $r > 4$.

Another differential equation we present here is

$$\frac{dx}{ds} = \frac{Ax - b}{\|Ax - b\|} \tag{55}$$

with the second derivative

$$\frac{d^2x}{ds^2} = \frac{1}{\|Ax - b\|^2} \left\{ A - \frac{[(Ax - b)^T A (Ax - b)]}{\|Ax - b\|^2} \right\} (Ax - b) \tag{56}$$

In this case, always

$$\left\| \frac{dx}{ds} \right\| = 1 \tag{57}$$

even for $x \rightarrow x_s$, but the second derivative tends to an infinite value when $x \rightarrow x_s$

$$\left\| \frac{d^2x}{ds^2} \right\| \sim K(s) \varepsilon^{-1} \tag{58}$$

where $K(s)$ is finite and $\varepsilon \equiv \|Ax - b\|$. The relationship between the differentials of the variables v and s in (49) and (55) is

$$ds = \frac{r(F(x))^{1-\frac{1}{r}}}{\|Ax - b\|} dv \quad (59)$$

and for $x \rightarrow x_S$ we have (see (54))

$$\frac{ds}{dv} \sim K_1(v) \varepsilon^{1-\frac{2}{r}} \quad (60)$$

The differential Eqs. (49) in v and (55) in s require practically the same amount of computation for their right-hand sides, i.e., one matrix-vector multiplication. The first derivatives dx/dv for $r = 2$ and dx/ds remain finite when x tends to x_S , while the second derivatives increase to infinite values as in (54) and (58). For $r = 4$ the second derivative in (52) remains finite when x tends to x_S (see (54)), while the first derivative and the ratio ds/dv tend to zero as in (54) and (60), respectively. If $r > 4$ even $\|d^2x/dv^2\|$ tends to zero as in (54).

Equations (49) and (55) can be integrated by classical numerical methods. Since we are not looking for an accurate solution of these equations all along from x_o to x_S but for finding accurately the final value $x = x_S$, we can use a lower order method, for instance, even Euler's method [7]. This yields an approximate value of x_S which is to be used as initial point for repeating the numerical integration procedure. As we get closer to the solution x_S , we decrease the step size in order to reduce the error. In the case of Euler's method the error is determined in terms of the norm of the second derivative. Higher order numerical integration methods can also be used in order to increase the computation efficiency.

To find a starting point for the integration procedure which is reasonably close to the solution point, one can minimize $F(x)$ in (46) along the normal direction, followed by a minimization of the distance to the solution point x_S along the direction of the normal to F [8]. These two preliminary steps are repeated a few times as needed.

Numerical experiments have been performed applying Euler's method to (49) for $r = 2$, $r = 4$ and $r = 8$, and to (55). Systems (11) of various sizes have been automatically generated and the differential Eqs. (49) for $r = 2$ and $r = 4$, and (55) have produced results with the least amount of computation when imposing an accuracy of 1%.

For matrices which are not symmetric positive definite, (46) is replaced with $F(x) = 1/2 (Ax - b)^T (Ax - b)$.

6. Conclusions

A special type of matrix series are used in Section 2 to express the relationship between some first order ordinary differential equations and systems of linear algebraic equations and, also, in Section 3 to derive efficient formulae for matrix exponentials that allow accurate and stable numerical computations in various applications. The main feature of these series consists in the fact that, starting with their first term which is already a matrix substantially better conditioned than the original problem matrix, each of the subsequent terms is obtained through a multiplication with a better and better conditioned matrix that tends to the identity matrix. The new matrix exponential formulae contain very rapidly convergent series and can be applied to general, arbitrarily conditioned, positive definite or not matrices. They are used in Section 4 for two new methods of solution for general

linear algebraic systems. One is an iterative method which corresponds to the solution of the differential Eq. (1) with $f(v) = 1/v$. It is based on the exact analytical expressions (30)–(32) that always yield results converging finally to the exact solution of the system (11). In a second method, the original algebraic system (11) is replaced with an equivalent system containing a matrix exponential $e^{-\alpha A}$ such that instead of inverting the system matrix A we have now to invert $I - e^{-\alpha A}$. The exact analytical solution is obtained in the form of a series of matrix exponentials which is transformed into an infinite matrix product in order to reduce substantially the necessary amount of computation. It should be remarked that, since the number of matrix-vector multiplications required for the application of one matrix exponential-vector multiplication only depends on the norm of the matrix while the number of matrix exponential-vector multiplications depends on the condition of the system matrix, the total number of arithmetic operations needed to achieve an imposed accuracy when applying each of the two methods is practically proportional to n^2 , where n is the dimension of the matrix. The two methods require a comparable total amount of computation. It is also remarkable that for both methods the necessary amount of computation can be roughly predicted beforehand in terms of the system size, the system condition and the desired accuracy.

In Section 5, a powerful method is presented based on the numerical integration of specially constructed ordinary differential equations.



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Optimal Control of Evolution Differential Inclusions with Polynomial Linear Differential Operators

Elimhan N. Mahmudov

Abstract

In this chapter, we studied a new class of problems in the theory of optimal control defined by polynomial linear differential operators. As a result, an interesting Mayer problem arises with higher order differential inclusions. Thus, in terms of the Euler-Lagrange and Hamiltonian type inclusions, sufficient optimality conditions are formulated. In addition, the construction of transversality conditions at the endpoints of the considered time interval plays an important role in future studies. To this end, the apparatus of locally adjoint mappings is used, which plays a key role in the main results of this chapter. The presented method is demonstrated by the example of the linear optimal control problem, for which the Weierstrass-Pontryagin maximum principle is derived.

Keywords: Euler-Lagrange, differential inclusion, set-valued mapping, polynomial differential operators, linear problem, transversality, Weierstrass-Pontryagin maximum principle

1. Introduction

This chapter concerns with the special kind of optimal control problem with differential inclusions, where the left-hand side of the evolution inclusion is polynomial linear differential operators with variable coefficients; in fact, the main difficulty in the considered problems is to construct the Euler-Lagrange type higher order adjoint inclusions and the transversality conditions. That is why in the whole literature, only the qualitative properties of second-order differential inclusions are investigated (see [1–3] and references therein).

The paper [1] gives necessary and sufficient conditions ensuring the existence of a solution to the second-order differential inclusion with Cauchy initial value problem. Furthermore, second-order interior tangent sets are introduced and studied to obtain such conditions. The paper [2] studies, in the context of Banach spaces, the problem of three boundary conditions for both second-order differential inclusions and second-order ordinary differential equations. The results are obtained in several new settings of Sobolev type spaces involving Bochner and Pettis integrals. In the paper [3], the existence of viable solutions to the Cauchy problem $x'' \in F(x, x'), x(0) = x_0, x'(0) = y_0$ is proved, where F is a set-valued map

defined on a locally compact set $M \subset \mathbb{R}^{2n}$, contained in the Frechet subdifferential of a φ -convex function of order two.

Some qualitative properties and optimization of first-order discrete and continuous time processes with lumped and distributed parameters have been expanding in all directions at an astonishing rate during the last few decades (see [4–13] and their references).

The optimization of higher order differential inclusions was first developed by Mahmudov in [14–21]. Since then this problem has attracted many author's attentions (see [22] and their references). The paper [14] studies a new class of problems of optimal control theory with Sturm-Liouville type differential inclusions involving second-order linear self-adjoint differential operators. By using the discretization method guaranteeing transition to continuous problem, the discrete and discrete-approximate inclusions are investigated. Necessary and sufficient conditions, containing both the Euler-Lagrange and Hamiltonian type inclusions, and "transversality" conditions are derived. The paper [15] deals with the optimization of the Bolza problem with third-order differential inclusions and arbitrary higher order discrete inclusions. The work [16] is devoted to the Bolza problem of optimal control theory given by second-order convex differential inclusions with second-order state variable inequality constraints. According to the proposed discretization method, problems with discrete-approximate inclusions and inequalities are investigated. Necessary and sufficient conditions of optimality including distinctive "transversality" condition are proved in the form of Euler-Lagrange inclusions. The paper [17] is concerned with the necessary and sufficient conditions of optimality for second-order polyhedral optimization described by polyhedral discrete and differential inclusions. The paper [18] is devoted to the study of optimal control theory with higher order differential inclusions and a varying time interval. Essentially, under a more general setting of problems and endpoint constraints, the main goal is to establish sufficient conditions of optimality for higher order differential inclusions. Thus with the use of Euler-Lagrange and Hamiltonian type of inclusions and transversal conditions on the "initial" sets, the sufficient conditions are formulated. The paper [21] studies a new class of problems of optimal control theory with state constraints and second-order delay discrete and delay differential inclusions. Under the "regularity" condition by using discrete approximations as a vehicle, in the forms of Euler-Lagrange and Hamiltonian type inclusions, the sufficient conditions of optimality for delay DFIs, including the peculiar transversality ones, are proved.

The present chapter is ordered in the following manner.

In Section 2 the necessary facts and supplementary results from the book of Mahmudov are given [23]; Hamiltonian function and locally adjoint mapping are introduced, and the problems with initial point constraints for polynomial linear differential operators governed by time-dependent set-valued mapping are formulated. In Section 3, we present the main results; on the basis of "transversality" conditions at the endpoints of the considered time interval, the sufficient conditions of optimality for differential inclusions with polynomial linear differential operators and with initial point constraints are proved. In particular, it is shown that our problems involve optimization of the so-called Sturm-Liouville type differential inclusions. To the best of our knowledge, there is no paper which considers optimality conditions for these problems in the literature, and we aim to fill this gap. Therefore, the novelty of our formulation of the problem is justified. To establish the Euler-Lagrange and Hamiltonian inclusions and the transversality conditions, we use the construction of a suitable rewriting of the primal polynomial linear differential operator and the rearrangement of its integration. The case of variable coefficients of polynomial linear differential operators turns out to be more

complicated, unless transversality assumptions at the endpoints of the considered time interval are applied. It should be noted that the main proof can be easily generalized to the nonconvex case. Then, using the new approach given in Section 4 of this chapter, we construct the Weierstrass-Pontryagin maximum condition [24] for the linear optimal control problem. Consequently, in the particular case, the maximum principle follows from the Euler-Lagrange inclusion.

In Section 5 the optimality conditions are given for convex problem with second-order differential inclusions and endpoint constraints. By using second-order suitable Euler-Lagrange type adjoint inclusions and transversality conditions, Theorem 5.1 is proved.

The main results in this section can be extended to the case of Hilbert spaces ℓ_2, L_2^n . We remind that a Hilbert space H is a real or complex inner product space that is also a complete metric space with respect to the distance function induced by the inner product [2]. By definition, every Hilbert space is also a Banach space. Furthermore, in every Hilbert space, the following parallelogram identity $\|x + y\|^2 + \|x - y\|^2 = 2(\|x\|^2 + \|y\|^2)$ holds. Conversely, every Banach space in which the parallelogram identity holds is a Hilbert space. Remember that ℓ_2 is a space of numerical sequences, such that if $x = \{x_i\}$, then $\sum_{i=1}^{\infty} x_i^2 < \infty$. In fact ℓ_2 is an infinity dimensional coordinate-wise Hilbert space with the corresponding inner product $\langle x, y \rangle = \sum_{i=1}^{\infty} x_i y_i$. Endowing a relevant norm, we have a Banach space. Obviously, optimization of problem with PLDOs can be reduced to problem with geometric constraints in such finite-dimensional Hilbert space. As is known with all the pairs of elements of this space, a certain finite number is associated, i.e., inner product, existence of which is guaranteed by applying the familiar Cauchy Schwarz-Bunyakovskii [25] inequality. We remark that in our case for $x = \{x_0, x_1, x_2, \dots\} \in \ell_2$ and $x^* = \{x_0^*, x_1^*, x_2^*, \dots\} \in \ell_2^*$, the inner product $\langle x, x^* \rangle = \sum_{i=1}^{\infty} x_i x_i^*$ is finite numbers since this series is convergent. Besides it is known [25] that ℓ_2 is a self-adjoint space, i.e., $\ell_p = \ell_q^*$, and $1/p + 1/q = 1$, and so $\ell_2 = \ell_2^*$ for $p = 2$. Thus a dual cone constructed can be defined. The set of square integrable functions $L_2^n([0, 1])$ is a Hilbert space with inner product $\langle x(t), y(t) \rangle = \int_0^1 x(t)y(t)dt$.

2. Preliminaries and problem statements

The basic concepts given in this section can be found in the book [23]; let \mathbb{R}^n be a n -dimensional Euclidean space, $\langle x, v \rangle$ be an inner product of elements $x, v \in \mathbb{R}^n$, and (x, v) be a pair of x, v . Let $F : \mathbb{R}^n \rightrightarrows \mathbb{R}^n$ be a set-valued mapping from \mathbb{R}^n into the set of subsets of \mathbb{R}^n . Therefore F is a convex set-valued mapping, if its graph $gph F = \{(x, v) : v \in F(x)\}$ is a convex subset of \mathbb{R}^{2n} . A set-valued mapping F is called closed if its $gph F$ is a closed subset in \mathbb{R}^{2n} . The domain of a set-valued mapping F is denoted by $dom F$ and is defined as $dom F = \{x : F(x) \neq \emptyset\}$. A set-valued mapping F is convex-valued if $F(x)$ is a convex set for each $x \in dom F$.

The Hamiltonian function and argmaximum set corresponding to a set-valued mapping F are defined by the relations correspondingly:

$$H_F(x, v^*) = \sup_v \{ \langle v, v^* \rangle : v \in F(x) \}, \quad v^* \in \mathbb{R}^n,$$

$$F_A(x, v^*) = \{ v \in F(x) : \langle v, v^* \rangle = H_F(x, v^*) \}$$

We set $H_F(x, v^*) = -\infty$ if $F(x) = \emptyset$. The interior and relative interior of a set $M \subset \mathbb{R}^{2n}$ are denoted by $intM$ and riM , respectively.

A convex cone $K_M(z_0), z_0 \in M$ is a cone of tangent directions if from $\bar{z} = (\bar{x}, \bar{v}) \in K_M(z_0)$ it follows that \bar{z} is a tangent vector to the set M at a point $z_0 \in M$, i.e., there exists such function $q : \mathbb{R}^1 \rightarrow \mathbb{R}^{2n}$ that $z_0 + \alpha\bar{z} + q(\alpha) \in M$ for sufficiently small $\alpha > 0$ and $\alpha^{-1}q(\alpha) \rightarrow 0$, as $\alpha \downarrow 0$.

For a set-valued mapping F , the set-valued mapping $F^* : \mathbb{R}^n \rightrightarrows \mathbb{R}^n$ is defined by

$$F^*(v^*; (x, v)) := \{x^* : (x^*, -v^*) \in K^*_{gph F}(x, v)\}$$

$$K_{gph F}(x, v) = \text{cone} [gph F - (x, v)], \forall (x^1, v^1) \in gph F.$$

It is called the LAM to F at a point $(x, v) \in gph F$, where $K^* = \{z^* : \langle \bar{z}, z^* \rangle \geq 0, \forall \bar{z} \in K\}$ denotes the dual cone to the cone K , as usual. Below by using the Hamiltonian function, associated to a set-valued mapping F , we will define another LAM. Thus, the LAM to “nonconvex” mapping F is defined as follows:

$$F^*(v^*; (x, v)) := \{x^* : H_F(x^1, v^*) - H_F(x, v^*) \leq \langle x^*, x^1 - x \rangle, \forall x^1 \in \mathbb{R}^n\},$$

$$(x, v) \in gph F, v \in F_A(x, v^*).$$

Clearly, for the convex mapping F , the Hamiltonian function $H_F(\cdot, \cdot, v^*)$ is concave, and the latter definition of LAM coincides with the previous definition of LAM ([23], p. 62). Note that prior to the LAM, the notion of coderivative has been introduced for set-valued mappings in terms of the basic normal cone to their graphs by Mordukhovich [26] and for the smooth convex maps, the two notions are equivalent.

The aim of Section 3 is to obtain the Euler-Lagrange type adjoint inclusion and sufficient optimality conditions for a problem with polynomial linear differential operators:

$$\text{Minimize } \varphi(x(1), x'(1), \dots, x^{(s-1)}(1)), \tag{1}$$

$$(P_V) Lx(t) \in F(x(t), t), \text{ a.e. } t \in [0, 1], \tag{2}$$

$$x(0) \in Q_0, x'(0) \in Q_1, x''(0) \in Q_2, \dots, x^{(s-1)}(0) \in Q_{s-1} \tag{3}$$

where $Lx = \sum_{k=1}^s p_k(t) D^k x$ is a PLDO of degree s with variable coefficients $p_k : [0, 1] \rightarrow \mathbb{R}^1$ and $D^k, k = 1, \dots, s$ is the operator of k th-order derivatives. In what follows for each k , a scalar function p_k is k th-order continuously differentiable function, $p_s(t) \neq 0$ on $[0, 1]$ identically, $F(\cdot, t) : \mathbb{R}^n \rightrightarrows \mathbb{R}^n$ is time-dependent set-valued mapping, $\varphi : (\mathbb{R}^n)^s \rightarrow \mathbb{R}^1$ is continuous function, $Q_j \subseteq \mathbb{R}^n, j = 0, 1, \dots, s - 1$ are nonempty subsets of \mathbb{R}^n , and $s (s \geq 2)$ is an arbitrary fixed natural number. It is required to find an arc $\tilde{x}(t)$ of the problem Eqs. (1)–(3) for the s th-order differential inclusions satisfying Eq. (2) almost everywhere (a.e.) on a considered time interval and minimizing the functional $\varphi(x(1), \dots, x^{(s-1)}(1))$. An arc $x(\cdot)$ is absolutely continuous and $s - 1$ order differentiable function, where $x^{(s)}(\cdot) \equiv \frac{d^s x(\cdot)}{dt^s} \in L^n_1([0, 1])$. Obviously, such class of functions is a Banach space, endowed with the different equivalent norms.

Remark 2.1. Notice that to get sufficient condition of optimality for the Mayer problem (P_V) described by ordinary evolution differential inclusions with PLDOs and with initial point constraints, using the discretized method, we consider the following s th-order discrete-approximate problem instead of the problem (P_V) :

$$\begin{aligned} & \text{minimize } \varphi(x(1 - (s - 1)h), \Delta x(1 - (s - 1)h), \dots, \Delta^{s-1}x(1 - (s - 1)h)), \\ & \sum_{k=1}^s p_k(t) \Delta^k x(t) \in F(x(t), t), t = 0, h, 2h, \dots, 1 - sh; \\ & \Delta^k x(0) \in Q_k, k = 0, \dots, s - 1. \end{aligned}$$

Here k th-order difference operator is defined as follows:

$$\Delta^k x(t) = \frac{1}{h^k} \sum_{s=0}^k (-1)^s C_k^s x(t + (k - s)h), \quad C_k^s = \frac{k!}{s!(k - s)!}, \quad t = 0, h, \dots, 1 - h.$$

Thus by using the method of approximation [23, 26, 27], we can establish necessary and sufficient conditions for the rather complicated s th-order discrete-approximate problem. Then by passing to the limit in necessary and sufficient conditions of this problem as $h \rightarrow 0$, we can construct the optimality condition for the Mayer problem (P_V) described by higher order differential inclusions with PLDOs and with initial point constraints. But in this chapter to avoid long calculations, derivations of these conditions are omitted.

3. Optimization of evolution differential inclusions with PLDOs

In the present section, we study sufficient optimality conditions for the problem (P_V) . Before all, we formulate the so-called s th-order Euler-Lagrange type differential inclusion and the transversality conditions:

$$\text{i. } L^* x^*(t) \in F^*(x^*(t); (\tilde{x}(t), L\tilde{x}(t)), t), \text{ a.e. } t \in [0, 1],$$

where $L^* x^*(t) = \sum_{k=1}^s (-1)^k D^k [p_k(t)x^*(t)]$ is the adjoint PLDO of the primal operator L .

$$\text{ii. } \sum_{k=0}^{s-1} (-1)^{s-k} D^{s-k-1} [p_{s-k}(0)x^*(0)] \in K_{Q_0}^*(\tilde{x}(0));$$

$$\sum_{k=0}^{s-2} (-1)^{s-k-1} D^{s-k-2} [p_{s-k}(0)x^*(0)] \in K_{Q_1}^*(\tilde{x}'(0));$$

$$D[p_s(0)x^*(0)] - p_{s-1}(0)x^*(0) \in K_{Q_{s-2}}^*(\tilde{x}^{(s-2)}(0));$$

$$-p_s(0)x^*(0) \in K_{Q_{s-1}}^*(\tilde{x}^{(s-1)}(0)).$$

$$\begin{aligned} \text{iii. } & \left(\sum_{k=0}^{s-1} (-1)^{s-k} D^{s-k-1} [p_{s-k}(1)x^*(1)], \sum_{k=0}^{s-2} (-1)^{s-k-1} D^{s-k-2} [p_{s-k}(1)x^*(1)], \dots, \right. \\ & \left. D[p_s(1)x^*(1)] - p_{s-1}(1)x^*(1), p_s(1)x^*(1) \right) \in \partial\varphi(\tilde{x}(1), \tilde{x}'(1), \dots, \tilde{x}^{(s-1)}(1)). \end{aligned}$$

Later on we assume that $x^*(t), t \in [0, 1]$ is absolutely a continuous function with the higher order derivatives until $s - 1$ and $x^{*(s)}(\cdot) \in L_1^n([0, 1])$. The following condition ensures that the LAM F^* is nonempty:

$$\text{iv. } L\tilde{x}(t) \in F_A(\tilde{x}(t), x^*(t), t), \text{ a.e. } t \in [0, 1] \text{ or, equivalently,}$$

$$\langle L\tilde{x}(t), x^*(t) \rangle = H_F(\tilde{x}(t), x^*(t)), L\tilde{x}(t) \in F(\tilde{x}(t), t).$$

The following are sufficient optimality conditions for evolution differential inclusions with PLDOs.

Theorem 3.1. Let φ be a continuous and convex function and $F(\cdot, t)$ a convex set-valued mapping. Moreover, let

$$Q_j, j = 0, \dots, s - 1$$

be convex sets. Then for optimality of the trajectory $\tilde{x}(\cdot)$ in the problem (P_V) with evolution differential inclusions and PLDOs, it is sufficient that there exists an absolutely continuous function $x^*(\cdot)$ with the higher order derivatives until $s - 1$, satisfying a.e. the Euler-Lagrange type differential inclusion with PLDOs Eqs. (i) and (iv) and transversality conditions Eqs. (ii) and (iii) at the endpoints $t = 0$ and $t = 1$.

Proof. Using Theorem 2.1 ([23], p. 62), the definition of the Hamiltonian function and condition Eq. (i), we obtain

$$H_F(x(t), x^*(t), t) - H_F(\tilde{x}(t), x^*(t), t) \leq \langle L^* x^*(t), x(t) - \tilde{x}(t) \rangle$$

which can be rewritten as follows

$$H_F(x(t), x^*(t), t) - H_F(\tilde{x}(t), x^*(t), t) \leq \left\langle \sum_{k=1}^s (-1)^k (p_k(t)x^*(t))^{(k)}, x(t) - \tilde{x}(t) \right\rangle. \tag{4}$$

Further using the definition of the Hamiltonian function, Eq. (4) can be converted to the inequality

$$0 \geq \langle Lx(t) - L\tilde{x}(t), x^*(t) \rangle - \langle L^* x^*(t), x(t) - \tilde{x}(t) \rangle$$

or

$$0 \geq \left\langle \sum_{k=1}^s p_k(t)(x(t) - \tilde{x}(t))^{(k)}, x^*(t) \right\rangle - \left\langle \sum_{k=1}^s (-1)^k (p_k(t)x^*(t))^{(k)}, x(t) - \tilde{x}(t) \right\rangle. \tag{5}$$

Integrating Eq. (5) over the interval $[0, 1]$, we have

$$0 \geq \int_0^1 \left[\left\langle \sum_{k=1}^s p_k(t)(x(t) - \tilde{x}(t))^{(k)}, x^*(t) \right\rangle - \left\langle \sum_{k=1}^s (-1)^k (p_k(t)x^*(t))^{(k)}, x(t) - \tilde{x}(t) \right\rangle \right] dt \tag{6}$$

Let us denote

$$B = \sum_{k=1}^s \left\langle (x(t) - \tilde{x}(t))^{(k)}, p_k(t)x^*(t) \right\rangle - \sum_{k=1}^s \left\langle (-1)^k (p_k(t)x^*(t))^{(k)}, x(t) - \tilde{x}(t) \right\rangle$$

In what follows our approach lies in reducing B in a relationship consisting of s sums from k ($k = 1, \dots, s$) to s of suitable derivatives of scalar products; thus, after some transformations we can deduce an important representation for a first term of B as follows:

$$\begin{aligned} \sum_{k=1}^s \left\langle (x(t) - \tilde{x}(t))^{(k)}, p_k(t)x^*(t) \right\rangle &= \sum_{k=1}^s \left[\frac{d}{dt} \left\langle x^{(k-1)}(t) - \tilde{x}^{(k-1)}(t), p_k(t)x^*(t) \right\rangle \right] \\ &\quad - \sum_{k=2}^s \left[\frac{d}{dt} \left\langle x^{(k-2)}(t) - \tilde{x}^{(k-2)}(t), (p_k(t)x^*(t))' \right\rangle \right] \\ &\quad + \sum_{k=3}^s \left[\frac{d}{dt} \left\langle x^{(k-3)}(t) - \tilde{x}^{(k-3)}(t), (p_k(t)x^*(t))'' \right\rangle \right] \\ &\quad - \dots + \sum_{k=s-2}^s \left[\frac{d}{dt} \left\langle x^{(k-s+2)}(t) - \tilde{x}^{(k-s+2)}(t), (-1)^{m-3} (p_k(t)x^*(t))^{(s-3)} \right\rangle \right] \quad (7) \\ &\quad + \sum_{k=s-1}^s \left[\frac{d}{dt} \left\langle x^{(k-s+1)}(t) - \tilde{x}^{(k-s+1)}(t), (-1)^{s-2} (p_k(t)x^*(t))^{(s-2)} \right\rangle \right] \\ &\quad + \frac{d}{dt} \left\langle x(t) - \tilde{x}(t), (-1)^{m-1} (p_s(t)x^*(t))^{(s-1)} \right\rangle \\ &\quad + \sum_{k=1}^s \left[\left\langle x(t) - \tilde{x}(t), (-1)^k (p_k(t)x^*(t))^{(k)} \right\rangle \right]. \end{aligned}$$

Then in view of Eq. (7) in the definition of B , we have an efficient formula:

$$\begin{aligned} B &= \sum_{k=1}^s \left[\frac{d}{dt} \left\langle x^{(k-1)}(t) - \tilde{x}^{(k-1)}(t), p_k(t)x^*(t) \right\rangle \right] \\ &\quad - \sum_{k=2}^s \left[\frac{d}{dt} \left\langle x^{(k-2)}(t) - \tilde{x}^{(k-2)}(t), (p_k(t)x^*(t))' \right\rangle \right] \quad (8) \\ &\quad + \sum_{k=3}^s \left[\frac{d}{dt} \left\langle x^{(k-3)}(t) - \tilde{x}^{(k-3)}(t), (p_k(t)x^*(t))'' \right\rangle \right] \\ &\quad - \dots + \sum_{k=s-2}^s \left[\frac{d}{dt} \left\langle x^{(k-s+2)}(t) - \tilde{x}^{(k-s+2)}(t), (-1)^{s-3} (p_k(t)x^*(t))^{(s-3)} \right\rangle \right] \\ &\quad + \sum_{k=s-1}^s \left[\frac{d}{dt} \left\langle x^{(k-s+1)}(t) - \tilde{x}^{(k-s+1)}(t), (-1)^{s-2} (p_k(t)x^*(t))^{(s-2)} \right\rangle \right] \\ &\quad + \frac{d}{dt} \left\langle x(t) - \tilde{x}(t), (-1)^{s-1} (p_s(t)x^*(t))^{(s-1)} \right\rangle. \end{aligned}$$

Then taking into account the structure of B in Eq. (8), we can compute the integral on the right-hand side of Eq. (6) as follows:

$$\int_0^1 B dt = \sum_{k=1}^s \left[\int_0^1 d \left\langle x^{(k-1)}(t) - \tilde{x}^{(k-1)}(t), p_k(t)x^*(t) \right\rangle \right]$$

$$\begin{aligned}
 & - \sum_{k=2}^s \left[\int_0^1 d \left\langle x^{(k-2)}(t) - \tilde{x}^{(k-2)}(t), (p_k(t)x^*(t))' \right\rangle \right] \\
 & + \sum_{k=3}^s \left[\int_0^1 d \left\langle x^{(k-3)}(t) - \tilde{x}^{(k-3)}(t), (p_k(t)x^*(t))'' \right\rangle \right] \\
 & - \dots + \sum_{k=s-2}^s \left[\int_0^1 d \left\langle x^{(k-s+2)}(t) - \tilde{x}^{(k-s+2)}(t), (-1)^{s-3} (p_k(t)x^*(t))^{(s-3)} \right\rangle \right] \\
 & + \sum_{k=s-1}^s \left[\int_0^1 d \left\langle x^{(k-s+1)}(t) - \tilde{x}^{(k-s+1)}(t), (-1)^{s-2} (p_k(t)x^*(t))^{(s-2)} \right\rangle \right] \\
 & + \int_0^1 d \left\langle x(t) - \tilde{x}(t), (-1)^{s-1} (p_s(t)x^*(t))^{(s-1)} \right\rangle.
 \end{aligned}$$

Thus, integrating B, we can obtain

$$\begin{aligned}
 & \int_0^1 B dt = \sum_{k=1}^s \left[\left\langle x^{(k-1)}(1) - \tilde{x}^{(k-1)}(1), p_k(1)x^*(1) \right\rangle \right. \\
 & \quad \left. - \left\langle x^{(k-1)}(0) - \tilde{x}^{(k-1)}(0), p_k(0)x^*(0) \right\rangle \right] \\
 & - \sum_{k=2}^s \left[\left\langle x^{(k-2)}(1) - \tilde{x}^{(k-2)}(1), (p_k(1)x^*(1))' \right\rangle - \left\langle x^{(k-2)}(0) - \tilde{x}^{(k-2)}(0), (p_k(0)x^*(0))' \right\rangle \right] \\
 & \quad + \sum_{k=3}^s \left[\left\langle x^{(k-3)}(1) - \tilde{x}^{(k-3)}(1), (p_k(1)x^*(1))'' \right\rangle \right. \\
 & \quad \left. - \left\langle x^{(k-3)}(0) - \tilde{x}^{(k-3)}(0), (p_k(0)x^*(0))'' \right\rangle \right] \\
 & - \dots + \sum_{k=s-2}^s \left[\left\langle x^{(k-s+2)}(1) - \tilde{x}^{(k-s+2)}(1), (-1)^{s-3} (p_k(1)x^*(1))^{(s-3)} \right\rangle \right. \\
 & \quad \left. - \left\langle x^{(k-s+2)}(0) - \tilde{x}^{(k-s+2)}(0), (-1)^{s-3} (p_k(0)x^*(0))^{(s-3)} \right\rangle \right] \\
 & + \sum_{k=s-1}^s \left[\left\langle x^{(k-s+1)}(1) - \tilde{x}^{(k-s+1)}(1), (-1)^{s-2} (p_k(1)x^*(1))^{(s-2)} \right\rangle \right. \\
 & \quad \left. - \left\langle x^{(k-s+1)}(0) - \tilde{x}^{(k-s+1)}(0), (-1)^{s-2} (p_k(0)x^*(0))^{(s-2)} \right\rangle \right] \\
 & + \left\langle x(1) - \tilde{x}(1), (-1)^{s-1} (p_s(1)x^*(1))^{(s-1)} \right\rangle \\
 & - \left\langle x(0) - \tilde{x}(0), (-1)^{s-1} (p_s(0)x^*(0))^{(s-1)} \right\rangle.
 \end{aligned}$$

Here by suitable rearrangement and necessary simplification, we have

$$\begin{aligned}
 \int_0^1 Bdt &= \sum_{k=1}^s \left[\left\langle x^{(k-1)}(1) - \tilde{x}^{(k-1)}(1), p_k(1)x^*(1) \right\rangle \right. \\
 &\quad - \sum_{k=2}^s \left[\left\langle x^{(k-2)}(1) - \tilde{x}^{(k-2)}(1), (p_k(1)x^*(1))' \right\rangle \right] \\
 &\quad + \sum_{k=3}^s \left[\left\langle x^{(k-3)}(1) - \tilde{x}^{(k-3)}(1), (p_k(1)x^*(1))'' \right\rangle \right] \\
 &\quad - \dots + \sum_{k=s-2}^s \left[\left\langle x^{(k-s+2)}(1) - \tilde{x}^{(k-s+2)}(1), (-1)^{s-3} (p_k(1)x^*(1))^{(s-3)} \right\rangle \right] \\
 &\quad + \sum_{k=s-1}^s \left[\left\langle x^{(k-s+1)}(1) - \tilde{x}^{(k-s+1)}(1), (-1)^{s-2} (p_k(1)x^*(1))^{(s-2)} \right\rangle \right] \\
 &\quad + \left\langle x(1) - \tilde{x}(1), (-1)^{s-1} (p_s(1)x^*(1))^{(s-1)} \right\rangle \\
 &\quad + \left\langle x(0) - \tilde{x}(0), \sum_{k=0}^{s-1} (-1)^{s-k} D^{s-k-1} [p_{s-k}(0)x^*(0)] \right\rangle \\
 &\quad + \left\langle x'(0) - \tilde{x}'(0), \sum_{k=0}^{s-2} (-1)^{s-k-1} D^{s-k-2} [p_{s-k}(0)x^*(0)] \right\rangle \\
 &\quad + \left\langle x''(0) - \tilde{x}''(0), \sum_{k=0}^{s-3} (-1)^{s-k-2} D^{s-k-3} [p_{s-k}(0)x^*(0)] \right\rangle \\
 &\quad + \dots - \left\langle x^{(s-2)}(0) - \tilde{x}^{(s-2)}(0), -D[p_s(0)x^*(0)] + p_{s-1}(0)x^*(0) \right\rangle \\
 &\quad - \left\langle x^{(s-1)}(0) - \tilde{x}^{(s-1)}(0), p_s(0)x^*(0) \right\rangle.
 \end{aligned} \tag{9}$$

In order to make use of the transversality condition Eq. (ii), we rewrite it in a more relevant form:

$$\begin{aligned}
 &\left\langle x(0) - \tilde{x}(0), \sum_{k=0}^{s-1} (-1)^{s-k} D^{s-k-1} [p_{s-k}(0)x^*(0)] \right\rangle \\
 &\quad + \left\langle x'(0) - \tilde{x}'(0), \sum_{k=0}^{s-2} (-1)^{s-k-1} D^{s-k-2} [p_{s-k}(0)x^*(0)] \right\rangle \\
 &\quad + \left\langle x''(0) - \tilde{x}''(0), \sum_{k=0}^{s-3} (-1)^{s-k-2} D^{s-k-3} [p_{s-k}(0)x^*(0)] \right\rangle \\
 &\quad + \dots - \left\langle x^{(s-2)}(0) - \tilde{x}^{(s-2)}(0), -D[p_s(0)x^*(0)] + p_{s-1}(0)x^*(0) \right\rangle \\
 &\quad - \left\langle x^{(s-1)}(0) - \tilde{x}^{(s-1)}(0), p_s(0)x^*(0) \right\rangle \geq 0; \forall x^{(k)}(0) \in K_{Q_k}(\tilde{x}^{(k)}(0)),
 \end{aligned}$$

$k = 0, \dots, s - 1$. Thus, from Eqs. (6) and (9), we have

$$\begin{aligned}
 0 \geq & \sum_{k=1}^s \left[\left\langle x^{(k-1)}(1) - \tilde{x}^{(k-1)}(1), p_k(1)x^*(1) \right\rangle \right. \\
 & - \sum_{k=2}^s \left\langle x^{(k-2)}(1) - \tilde{x}^{(k-2)}(1), \frac{d(p_k(1)x^*(1))}{dt} \right\rangle \\
 & + \sum_{k=3}^s \left[\left\langle x^{(k-3)}(1) - \tilde{x}^{(k-3)}(1), \frac{d^2(p_k(1)x^*(1))}{dt^2} \right\rangle \right] - \dots \\
 & + \sum_{k=s-2}^s \left[\left\langle x^{(k-s+2)}(1) - \tilde{x}^{(k-s+2)}(1), (-1)^{s-3} \frac{d^{s-3}(p_k(1)x^*(1))}{dt^{s-3}} \right\rangle \right] \\
 & + \sum_{k=s-1}^s \left[\left\langle x^{(k-s+1)}(1) - \tilde{x}^{(k-s+1)}(1), (-1)^{s-2} \frac{d^{s-2}(p_k(1)x^*(1))}{dt^{s-2}} \right\rangle \right] \\
 & + \left\langle x(1) - \tilde{x}(1), (-1)^{s-1} \frac{d^{s-1}(p_s(1)x^*(1))}{dt^{s-1}} \right\rangle.
 \end{aligned}$$

Using the derivative operator D , it is not hard to see that the relation described above can be expressed in a more compact form:

$$\begin{aligned}
 0 \geq & \left\langle \sum_{k=0}^{s-1} (-1)^{s-k-1} D^{s-k-1} [p_{s-k}(1)x^*(1)], x(1) - \tilde{x}(1) \right\rangle \\
 & + \left\langle \sum_{k=0}^{s-2} (-1)^{s-k-2} D^{s-k-2} [p_{s-k}(1)x^*(1)], x'(1) - \tilde{x}'(1) \right\rangle + \dots \quad (10) \\
 & - \left\langle D[p_s(1)x^*(1)] - p_{s-1}(1)x^*(1), x^{(s-2)}(1) - \tilde{x}^{(s-2)}(1) \right\rangle \\
 & + \left\langle p_s(1)x^*(1), x^{(s-1)}(1) - \tilde{x}^{(s-1)}(1) \right\rangle.
 \end{aligned}$$

Furthermore, applying the definition of the transversality condition Eq. (iii) for all feasible arc $x(\cdot)$, we have

$$\begin{aligned}
 & \varphi(x(1), x'(1), \dots, x^{(s-1)}(1)) - \varphi(\tilde{x}(1), \tilde{x}'(1), \dots, \tilde{x}^{(s-1)}(1)) \\
 & \geq \left\langle \sum_{k=0}^{s-1} (-1)^{s-k} D^{s-k-1} [p_{s-k}(1)x^*(1)], x(1) - \tilde{x}(1) \right\rangle \\
 & + \left\langle \sum_{k=0}^{s-2} (-1)^{s-k-1} D^{s-k-2} [p_{s-k}(1)x^*(1)], x'(1) - \tilde{x}'(1) \right\rangle + \dots \\
 & + \left\langle D[p_s(1)x^*(1)] - p_{s-1}(1)x^*(1), x^{(s-2)}(1) - \tilde{x}^{(s-2)}(1) \right\rangle \\
 & - \left\langle p_s(1)x^*(1), x^{(s-1)}(1) - \tilde{x}^{(s-1)}(1) \right\rangle \quad (11)
 \end{aligned}$$

Then from the last two inequalities Eqs. (10) and (11) for all feasible arc $x(\cdot)$, we have immediately $\varphi(x(1), x'(1), \dots, x^{(s-1)}(1)) \geq \varphi(\tilde{x}(1), \tilde{x}'(1), \dots, \tilde{x}^{(s-1)}(1))$, that is, $\tilde{x}(\cdot)$ is an optimal trajectory. \square

Remark 3.1. It can be noted that in the particular case, if $p_2(t) = p(t), p_1(t) = p'(t)$, where $p(\cdot) : [0, 1] \rightarrow (0, \infty)$, the second-order linear differential operator $Lx = p_2(t)x'' + p_1(t)x'$ is a well-known self-adjoint Sturm-Liouville operator $Lx \equiv (px')'$.

Corollary 3.1. Let $F(\cdot, t)$ be a closed set-valued mapping. Then under the assumptions of Theorem 3.1, the conditions Eqs. (i) and (iii) can be rewritten in terms of Hamiltonian function as follows:

$$L^* x^*(t) \in \partial_x H_F(\tilde{x}(t), x^*(t), t); L\tilde{x}(t) \in \partial_{v^*} H(\tilde{x}(t), x^*(t), t), \text{ a.e. } t \in [0, 1]$$

Proof. Indeed, by Theorem 2.1 ([23], p. 62) and Lemma 5.1 [14], we can write.

$$F^*(v^*; (x, v), t) = \partial_x H_F(x, v^*, t), \text{ and } F_A(x, v^*, t) = \partial_{v^*} H_F(x, v^*, t)$$

respectively. Then it is easy to see that the result of corollary are equivalent with the conditions Eqs. (i) and (iv) of Theorem 3.1. \square

Below nonconvexity of a set-valued mapping $F(\cdot, t)$ means that its Hamilton function in general is a nonconcave function satisfying the condition Eq. (a).

Theorem 3.2. Suppose that we have the “nonconvex” problem (P_V) , that is, $\varphi : (\mathbb{R}^n)^s \rightarrow \mathbb{R}^1$ and $F(\cdot, t) : \mathbb{R}^n \rightrightarrows \mathbb{R}^n$ in general are nonconvex function and set-valued mapping, respectively. Moreover, suppose that $K_{Q_j}(\tilde{x}^{(j)}(0)), \tilde{x}^{(j)}(0) \in Q_j$ is the cones of tangent directions to $Q_j, j = 0, \dots, s-1$.

Then for optimality of the trajectory $\tilde{x}(\cdot)$, it is sufficient that there exists an absolutely continuous function $x^*(\cdot)$, satisfying the following conditions:

a. $L^* x^*(t) \in F^*(x^*(t); (\tilde{x}(t), L\tilde{x}(t)), t)$, a.e. $t \in [0, 1]$,

b. $\sum_{k=0}^{s-1-j} (-1)^{s-k-j} D^{s-k-1-j} [p_{s-k}(0)x^*(0)] \in K_{Q_j}^*(\tilde{x}^{(j)}(0)), j = 0, \dots, s-1$,

c. $\varphi(v_0, v_1, \dots, v_{s-1}) - \varphi(\tilde{x}(1), \tilde{x}'(1), \dots, \tilde{x}^{(s-1)}(1))$

$$\geq \sum_{j=0}^{s-1} \left\langle \sum_{k=0}^{s-1-j} (-1)^{s-k-j} D^{s-k-1-j} [p_{s-k}(1)x^*(1)], v_j - \tilde{x}^{(j)}(1) \right\rangle,$$

$\forall v_j \in \mathbb{R}^n, j = 0, \dots, s-1$,

d. $\langle L\tilde{x}(t), x^*(t) \rangle = H_F(\tilde{x}(t), x^*(t), t)$, a.e. $t \in [0, 1]$.

Proof. In the proof of Theorem 3.1, we have used the following inequality:

$$\begin{aligned} & H_F(x(t), x^*(t), t) - H_F(\tilde{x}(t), x^*(t), t) \\ & \leq \left\langle \sum_{k=1}^s (-1)^k (p_k(t)x^*(t))^{(k)}, x(t) - \tilde{x}(t) \right\rangle. \end{aligned} \quad (12)$$

Hence, from the inequality Eq. (12), immediately we have the inequality Eq. (10). Moreover, setting $v_j = \tilde{x}^{(j)}(1) (j = 0, \dots, s-1)$ for all feasible trajectories $x(\cdot)$, it is not hard to see that for nonconvex φ the following inequality holds:

$$\begin{aligned} & \varphi\left(x(1), x'(1), \dots, x^{(s-1)}(1)\right) - \varphi\left(\tilde{x}(1), \tilde{x}'(1), \dots, \tilde{x}^{(s-1)}(1)\right) \\ & \geq \sum_{j=0}^{s-1} \left\langle \sum_{k=0}^{s-1-j} (-1)^{s-k-j} D^{s-k-1-j} [p_{s-k}(1)x^*(1)], x^{(j)}(1) - \tilde{x}^{(j)}(1) \right\rangle, j = 0, 1, \dots, s-1, \end{aligned}$$

Then for the furthest proof, we proceed by analogy with the preceding derivation of Theorem 3.1. \square

4. Some applications to optimal control problems with PLDOs

In this section we give two applications of our results. The first one is the particular Mayer problem for differential inclusions involving PLDOs with constant coefficients, and the second one concerns optimization of “linear” differential inclusions with PLDOs and constant coefficients. Thus, suppose now we have the following optimization problem (for simplicity we consider a convex problem) with s th-order PLDO with constant coefficients:

$$\begin{aligned} & \text{Minimize } \varphi_0(x(1)), \\ & \text{(P}_C\text{)} \quad Lx(t) \in F(x(t), t), \text{ a.e. } t \in [0, 1], Lx = D^s x + p_1 D^{s-1} x + \dots + p_{s-1} D x \\ & \quad x(0) = \alpha_0, x'(0) = \alpha_1, x''(0) = \alpha_2, \dots, x^{(s-1)}(0) = \alpha_{s-1}, \end{aligned} \quad (13)$$

where L is the s th-order polynomial operator, $p_k, k = 1, \dots, s-1$ are some real constants, $F(\cdot, t) : \mathbb{R}^n \rightrightarrows \mathbb{R}^n$ is a convex set-valued mapping, $\varphi_0 : \mathbb{R}^n \rightarrow \mathbb{R}^1$ is a continuous convex function, and $\alpha_j \in \mathbb{R}^n, j = 0, \dots, s-1$ are fixed n -dimensional vectors. It is known that the multiplication operation is commutative for polynomial linear differential operators with constant coefficients. On the other hand, the s th-order adjoint operator is defined as follows:

$$L^* x^* = (-1)^s D^s x^* + (-1)^{s-1} p_1 D^{s-1} x^* + \dots - p_{s-1} D x^*.$$

Corollary 4.1. Let φ_0 and $F(\cdot, t)$ be a convex function and a set-valued mapping, respectively. Then, for the trajectory $\tilde{x}(\cdot)$ to be optimal in the problem (P_C), it is sufficient that there exists an absolutely continuous function $x^*(\cdot)$ satisfying the Euler-Lagrange type differential inclusion.

$$\begin{aligned} & L^* x^*(t) \in F^*(x^*(t); (\tilde{x}(t), L\tilde{x}(t)), t); \langle L\tilde{x}(t), x^*(t) \rangle = H_F(\tilde{x}(t), x^*(t)), \\ & \text{a.e. } t \in [0, 1], L^* x^* = (-1)^s x^{*(s)} + (-1)^{s-1} p_1 x^{*(s-1)} + \dots - p_{s-1} x^{*'} \end{aligned}$$

and transversality condition at the endpoint $t = 1$.

$$(-1)^s x^{*(s-1)}(1) \in \partial\varphi_0(\tilde{x}(1)), x^{*(j)}(1) = 0, j = 0, \dots, m-2.$$

Proof. We conclude this proof by returning to the conditions Eqs. (i)–(iii) of Theorem 3.1. Clearly, a problem (P_C) can be reduced to the problem of form (P_V), where

$$\varphi\left(x(1), x'(1), \dots, x^{(s-1)}(1)\right) \equiv \varphi_0(x(1)).$$

It follows that $\partial\varphi(x(1), x'(1), \dots, x^{(s-1)}(1)) = \partial_x\varphi_0(x(1)) \times \underbrace{(0, \dots, 0)}_{s-1}$. On the other hand, since $p_s(t) \equiv 1, p_j(t) = p_{s-j}$, and $j = 1, \dots, s-1$ are constants, by sequentially substitution in the transversality condition Eq. (iii), we derive that

$$\sum_{k=0}^{s-1-j} (-1)^{s-k-j} D^{s-k-1-j} [p_{s-k}(1)x^*(1)] = x^{*(s-1-j)}(1) = 0, j = 1, \dots, s-1,$$

and therefore for $j = 0$.

$$\sum_{k=0}^{s-1} (-1)^{s-k} D^{s-k-1} [p_{s-k}(1)x^*(1)] = (-1)^s x^{*(s-1)}(1) \in \partial\varphi_0(\tilde{x}(1)). \square$$

Suppose now that we have the so-called linear Mayer problem with PLDOs:

$$\text{Minimize } \varphi_0(x(1)), \quad (14)$$

$$Lx(t) \in F(x(t), t), \text{ a.e. } t \in [0, 1],$$

$$x^{(j)}(0) = \alpha_j, j = 0, \dots, s-1, F(x, t) = A(t)x + B(t)U \quad (15)$$

where φ_0 differentiable convex function; $A(t)$ and $B(t)$ are $n \times n$ and $n \times r$ continuous matrices, respectively; U is a convex compact of \mathbb{R}^r ; $\alpha_j, j = 0, \dots, s-1$ are constant vectors. It is required to find a control function $\tilde{u}(\cdot)$ such that the corresponding trajectory $\tilde{x}(\cdot)$ minimizes the Mayer functional $\varphi_0(x(1))$.

In fact, this is optimization of Cauchy problem for “linear” differential inclusions with PLDO. The controlling parameter $u(\cdot)$ is called admissible if it only takes values in the given control set U which is a nonempty, convex compact.

Theorem 4.1. The arc $\tilde{x}(t)$ corresponding to the controlling parameter $\tilde{u}(t)$ is a solution to Eqs. (14) and (15) if there exists an absolutely continuous function $x^*(\cdot)$, satisfying the Euler-Lagrange type differential equation, the transversality condition, and Weierstrass-Pontryagin maximum principle:

$$L^* x^*(t) = A^*(t)x^*(t), \text{ a.e. } t \in [0, 1],$$

$$(-1)^s x^{*(s-1)}(1) = \varphi'_0(\tilde{x}(1)), x^{*(j)}(1) = 0, j = 0, \dots, s-2$$

$$\langle B(t)\tilde{u}(t), x^*(t) \rangle = \max_{u \in U} \langle B(t)u, x^*(t) \rangle.$$

Proof. Obviously, the Hamiltonian is

$$H_F(x, v^*, t) = \langle A(t)x, v^* \rangle + \max_{u \in U} \langle B(t)u, v^* \rangle.$$

Hence,

$$F^*(v^*; (x, \tilde{v}), t) = \partial_x H_F(x, v^*, t) = A^*(t)v^*, \tilde{v} \in F_A(x, v^*, t)\tilde{v} = A(t)x + B(t)\tilde{u}$$

where the argmaximum inclusion $\tilde{v} \in F_A(x, v^*, t)$ implies that $\langle B(t)\tilde{u}, v^* \rangle = \max_{u \in U} \langle B(t)u, v^* \rangle$ and $F^*(v^*; (x, \tilde{v}), t) \neq \emptyset$. Thus, applying Theorem 3.1, we obtain

$$L^* x^*(t) = A^*(t)x^*(t), L\tilde{x}(t) \in F_A(\tilde{x}(t), x^*(t), t),$$

$$\langle B(t)\tilde{u}(t), x^*(t) \rangle = \max_{u \in U} \langle B(t)u, x^*(t) \rangle.$$

Consequently, the transversality condition Eq. (ii) of Theorem 3.1 is unnecessary and by Corollary 4.1 $(-1)^s D^{s-1} x^*(1) = \varphi'_0(\tilde{x}(1))$, $D^j x^*(1) = 0, j = 0, \dots, s-2$. \square

Remark 4.1. Suppose that in the definition of s th-order PLDO (see Eq. (13)) $s = 1, p_i = 0, i = 1, \dots, s-1$ and U is a convex closed polyhedron. Then we have linear equations with variable coefficients $x' = A(t)x + B(t)u, u \in U$ in the finite time interval $t \in [0, 1]$. Obviously, for such problems an adjoint Euler-Lagrange type differential equation and transversality condition at a point $t = 1$ consist of the following: $x^{*'}(t) = -A^*(t)x^*(t), x^*(1) = -\varphi'_0(\tilde{x}(1))$. We remind that along with Pontryagin's maximum principle (see, e.g., [24]) under the condition for generality of position for time-optimal problem, the existence results of optimal control are proved.

Example 4.1. Let us consider the following Mayer problem with second-order PLDO $Lx = D^2x = x''$:

$$\text{Infimum } \varphi(x(1), x'(1)) \text{ is subject to } x'' = u, u \in [-1, 1], x(0) \in Q_0, x'(0) \in Q_1. \quad (16)$$

Here $\varphi(x(1), x'(1)) = x'^2(1) - x(1)$ and $Q_0 = \{0\}, Q_1 = \{1\}$.

It should be noted that substituting $F(t) = u(t), x''(t) = a(t), m = 1$ into Newton's second law $F(t) = ma(t)$, we have $x'' = u$.

Obviously, in this problem $F(x, t) \equiv F(x) = \{u : |u| \leq 1\}, s = 2$.

Then Eq. (16) has the form:

$$\text{Infimum } \varphi(x(1), x'(1)) \text{ is subject to } Lx \in F(x), t \in [0, 1], x(0) = 0, x'(0) = 1. \quad (17)$$

It can be easily seen that in the adjoint inclusion Eq. (i).

$-D(p_1(t)x^*(t)) + D^2(p_2(t)x^*(t)) \in F^*(x^*(t); (\tilde{x}(t), L\tilde{x}(t)))$
of Corollary 3.1 $p_2(t) \equiv 0$ and $p_2(t) \equiv 1$, and so we have

$$x^{*''}(t) \in F^*(x^*(t); (\tilde{x}(t), \tilde{x}''(t))).$$

Now, it is not hard to see that

$$H_F(x, v^*) = \max_u \{uv^* : |u| \leq 1\} = |v^*| \quad (18)$$

and

$$F^*(v^*; (x, v)) = \partial_x H_F(x, v^*) \equiv 0, v \in F_A(x, v^*) = \{-1, +1\}. \quad (19)$$

Then taking into account $Lx^* = d^2x^*/dt^2$, as a result of Theorem 3.1 (see also Corollary 3.1) from Eq. (19), we deduce that

$$x^{*''} = 0, t \in [0, 1],$$

for which the solution is a linear function of the form $x^*(t) = C_1t + C_2$, where C_1, C_2 are arbitrary constants. Then Eq. (18) implies that $\tilde{u}(t)x^*(t) = |x^*(t)|$ or

$$\tilde{u}(t) = \begin{cases} \operatorname{sgn} x^*(t), & \text{if } x^*(t) \neq 0, \\ \forall u_0 \in [-1, 1], & \text{if } x^*(t) = 0. \end{cases} \quad (20)$$

Further, from the linearity of $x^*(\cdot)$ and from Eq. (20), we insure that each optimal control function is a piecewise constant function.

In addition, by the transversality condition Eq. (iii) of Corollary 3.1, we can write.

$(x^{*'}(1), -x^*(1)) \in \partial\varphi(\tilde{x}(1), \tilde{x}'(1))$. On the other hand, it is not hard to see that $\varphi(x, y) = y^2 - x$ is a convex function; in fact, the 2×2 Hessian matrix

$$\varphi''(x, y) = \begin{bmatrix} \varphi''_{xx}(x, y) & \varphi''_{xy}(x, y) \\ \varphi''_{yx}(x, y) & \varphi''_{yy}(x, y) \end{bmatrix} = \begin{bmatrix} 0 & 0 \\ 0 & 2 \end{bmatrix}$$

is a positive semidefinite, that is, all eigenvalues of $\varphi''(x, y)$ are nonnegative. Indeed, denoting this matrix by A , we see that the characteristic equation $|A - \lambda E| = \lambda^2 - 2\lambda = 0$ (E is a 2×2 unique square matrix) has two real nonnegative eigenvalues $\lambda_1 = 0, \lambda_2 = 2$. Consequently, $\varphi(x, y)$ is convex and $\partial\varphi(x, y) = (-1, 2y)$. It follows that $\partial\varphi(\tilde{x}(1), \tilde{x}'(1)) = (-1, 2\tilde{x}'(1))$. Comparing this relation with $(x^{*'}(1), -x^*(1)) \in \partial\varphi(\tilde{x}(1), \tilde{x}'(1))$, we immediately have $x^{*'}(1) = -1, x^*(1) = -2\tilde{x}'(1)$. Then from a general solution of the adjoint Euler-Lagrange type inclusion (equation) $x^*(t) = C_1 t + C_2$, we have $-2\tilde{x}'(1) = x^*(1) = C_1 + C_2, -1 = x^{*'}(1) = C_1$ (C_1, C_2 are arbitrary constants), and so $x^*(t) = 1 - t - 2\tilde{x}'(1)$, whence $x^*(t) \neq 0$, if $t \neq \tau = 1 - 2\tilde{x}'(1)$. Therefore, Eq. (20) implies that for optimal control $\tilde{u}(\cdot)$, there are four possibilities:

$$\tilde{u}(t) = 1, \quad x^*(t) > 0, \quad t \in [0, 1]. \quad (21)$$

$$\tilde{u}(t) = -1, \quad x^*(t) < 0, \quad t \in [0, 1]. \quad (22)$$

$$\tilde{u}(t) = \begin{cases} 1, & \text{if } 0 \leq t < \tau, \\ -1, & \text{if } \tau < t \leq 1. \end{cases} \quad (23)$$

$$\tilde{u}(t) = \begin{cases} -1, & \text{if } 0 \leq t < \tau, \\ 1, & \text{if } \tau < t \leq 1. \end{cases} \quad (24)$$

(observe that τ is a point of discontinuity of $\tilde{u}(\cdot)$ and the values of the control functions $\tilde{u}(\cdot)$ at a point of discontinuity τ are unessential). As a consequence, it follows that either the sign of the linear function $x^*(t)$ does not change for the whole interval $[0, 1]$ or $x^*(t) > 0, 0 \leq t < \tau; x^*(t) < 0, \tau < t \leq 1$ for a some τ in the interval $0 < \tau < 1$ (the case Eq. (24) is excluded). Therefore, since $\tilde{u}(t)$ is a piecewise constant function, having not more than two intervals of constancy, we have either the cases Eqs. (21) and (22) or the case Eq. (23). In general, using Eqs. (21)–(23), by solving the Cauchy problem

$$x''(t) = u(t), \quad x(0) = 0, \quad x'(0) = 1 \quad (25)$$

we have a unique solution of the initial value problem Eq. (25). Thus for the time interval on which $u = 1$, we have $x'(t) = t + c_1; x(t) = t^2/2 + c_1 t + c_2$ (c_1, c_2 are constants). From Eq. (25) we obtain.

$$x'(t) = t + 1; \quad x(t) = t^2/2 + t. \quad (26)$$

By analogy, for $u = -1$ we have.

$$x'(t) = 1 - t; x(t) = -t^2/2 + t. \tag{27}$$

Now, let $x_1(t)$ and $x_2(t)$ be parabolas of Eqs. (26) and (27), respectively. Here, in the case Eq. (21), $\tilde{u}(t) = 1, t \in [0, 1]$, and so from Eq. (26), we have $\tilde{x}_1(1) = 0.5 + 1 = 1.5; \tilde{x}'_1(1) = 2$. Consequently, the value of problem Eq. (16) is $\varphi(\tilde{x}_1(1), \tilde{x}'_1(1)) = \tilde{x}_1'^2(1) - \tilde{x}_1(1) = 2^2 - (1.5) = 2.5$, if $\tilde{u}(t) = 1, t \in [0, 1]$. By a similar way, for a control function $\tilde{u}(t) = -1, t \in [0, 1]$ from Eq. (27), we obtain that $\tilde{x}_2(1) = -1^2/2 + 1 = 0.5, \tilde{x}'_2(1) = 0$ and $\varphi(\tilde{x}_2(1), \tilde{x}'_2(1)) = 0^2 - (0.5) = -0.5$.

On the other hand, in the case Eq. (23), the control function $\tilde{u}(t)$ first is equal to $+1$ and then equal to -1 , and the trajectory $\tilde{x}(t)$ consists of two pieces of parabolas $\tilde{x}_1(t)$ and $\tilde{x}_2(t)$ ($\tilde{x}(t)$ is continuous and piecewise smooth on the interval $0 \leq t \leq 1$). Then the solution of the equation Eq. (25) on the interval $0 \leq t \leq \tau$ is given by Eq. (26); at a point τ are satisfied $x_1(\tau) = \tau^2/2 + \tau, x'_1(\tau) = 1 + \tau$. Consider now the initial value problem:

$$x_2''(t) = -1, x_2(\tau) = \tau^2/2 + \tau, x'_2(\tau) = 1 + \tau, t \in [\tau, 1]. \tag{28}$$

It is clear that $(\tau^2/2) + \tau = x_2(\tau) = -(\tau^2/2) + c_1\tau + c_2$ and $1 + \tau = x'_2(\tau) = -\tau + c_1$ from which we obtain that the solution of the initial value problem Eq. (28) is $\tilde{x}_2(t) = -(t^2/2) + (1 + 2\tau)t - \tau^2$. Substituting the value $\tau = 1 - 2\tilde{x}'_1(1)$ into equation $\tilde{x}'_2(t) = 1 - t + 2\tau$, we have $5\tilde{x}'_2(1) = 2, \tilde{x}'_2(1) = \tilde{x}'_1(1) = 0.4$ (it follows that $\tau = 0.2$). Moreover, $\tilde{x}_2(1) = 2\tau - \tau^2 + (0.5)$ and $\tilde{x}_2(1) = \tilde{x}(1) = (3/2) - 4\tilde{x}'_1^2(1) = 0.86$. Thus, the value of our Mayer problem is $\varphi(\tilde{x}(1), \tilde{x}'(1)) = \tilde{x}'^2(1) - \tilde{x}(1) = (2/5)^2 - (43/50) = -0.7$, where $\tilde{u}(t)$ is defined as in Eq. (23). Comparing the values 2.5, $-0.5, -0.7$, we believe that the value of Mayer problem is -0.7 .

5. Sufficient conditions of optimality for second-order evolution differential inclusions with endpoint constraints

Note that in this section the optimality conditions are given for second-order convex differential inclusions (P_M) with convex endpoint constraints. These conditions are more precise than any previously published ones since they involve useful forms of the Weierstrass-Pontryagin condition and second-order Euler-Lagrange type adjoint inclusions. In the reviewed results, this effort culminates in Theorem 5.1:

$$\begin{aligned} &\text{Minimize } g(x(1), x'(1)), \\ &(P_M) \ x''(t) \in F(x(t), x'(t), t), \text{ a.e. } t \in [0, 1], \\ &x(0) = x_0, x'(0) = x_1; x(1) \in M_0, x'(1) \in M_1, \end{aligned}$$

where g is a convex continuous function, $F(\cdot, t) : \mathbb{R}^{2n} \rightrightarrows \mathbb{R}^n$ is convex set-valued mapping, and $M_0, M_1 \subseteq \mathbb{R}^n$ are convex sets.

The following adjoint inclusion is the second-order Euler-Lagrange type inclusion for the problem (P_M) :

$$a_1. (x^{*''}(t) + v^{*'}(t), v^*(t)) \in F^*(x^*(t); (\tilde{x}(t), \tilde{x}'(t), \tilde{x}''(t)), t), \text{ a.e. } t \in [0, 1],$$

where

$$b_1. \tilde{x}''(t) \in F_A(\tilde{x}(t), \tilde{x}'(t); x^*(t), t), \text{ a.e. } t \in [0, 1].$$

In what follows we assume that $x^*(t)$, $t \in [0, 1]$ is absolutely continuous function together with the first-order derivatives for which $x^{*''}(\cdot) \in L_1^n([0, 1])$. Besides the auxiliary function $v^*(t)$, $t \in [0, 1]$ is absolutely continuous and $v^{*'}(\cdot) \in L_1^n([0, 1])$.

The transversality conditions at the endpoint $t = 1$ consist of the following:

$$c_1. (v^*(1) + x^{*'}(1), -x^*(1)) \in \partial_{(x,u)} g(\tilde{x}(1), \tilde{x}'(1)) - K_{M_0}^*(\tilde{x}(1)) \times K_{M_1}^*(\tilde{x}'(1)).$$

Now we are ready to formulate the following theorem of optimality.

Theorem 5.1. Suppose that g is a continuous and convex function, $F(\cdot, t)$ is a convex set-valued mapping, and M_0, M_1 are convex sets. Then for optimality of the feasible trajectory $\tilde{x}(t)$ in the problem (P_M) , it is sufficient that there exists a pair of absolutely continuous functions:

$$\{x^*(t), v^*(t)\}, t \in [0, 1]$$

satisfying a.e. the second-order Euler-Lagrange type inclusions Eqs. (a_1) and (b_1) and the transversality condition Eq. (c_1) at the endpoint $t = 1$.

Proof. By the proof idea of Theorem 3.1 from Eqs. (a_1) and (b_1) , we obtain the adjoint differential inclusion of second order:

$$(x^{*''}(t) + v^{*'}(t), v^*(t)) \in \partial_{(x,u)} H_F(\tilde{x}(t), \tilde{x}'(t), x^*(t), t), t \in [0, 1].$$

On the definition of subdifferential set of the Hamiltonian function $H_F(\cdot, t)$ for all feasible trajectory $x(t)$, $t \in [0, 1]$, we rewrite the last relation in the equivalent form:

$$\begin{aligned} & H_F(x(t), x'(t), x^*(t), t) - H_F(\tilde{x}(t), \tilde{x}'(t), x^*(t), t) \\ & \leq \langle x^{*''}(t) + v^{*'}(t), x(t) - \tilde{x}(t) \rangle + \langle v^*(t), x'(t) - \tilde{x}'(t) \rangle. \end{aligned} \quad (29)$$

Now by using definition of the Hamiltonian function, the inequality Eq. (29) can be reduced to the inequality

$$0 \geq \langle (x(t) - \tilde{x}(t))'', x^*(t) \rangle - \langle x^{*''}(t), x(t) - \tilde{x}(t) \rangle - \frac{d}{dt} \langle v^*(t), x(t) - \tilde{x}(t) \rangle. \quad (30)$$

Integrating of the inequality Eq. (30) over the interval $[0, 1]$, we derive that

$$\begin{aligned} 0 & \geq \int_0^1 [\langle (x(t) - \tilde{x}(t))'', x^*(t) \rangle - \langle x^{*''}(t), x(t) - \tilde{x}(t) \rangle] dt \\ & \quad + \langle v^*(0), x(0) - \tilde{x}(0) \rangle - \langle v^*(1), x(1) - \tilde{x}(1) \rangle. \end{aligned} \quad (31)$$

For convenience we transform the expression in the square parentheses on the right-hand side of Eq. (31) as follows

$$\langle (x(t) - \tilde{x}(t))'', x^*(t) \rangle - \langle x^{*''}(t), x(t) - \tilde{x}(t) \rangle$$

$$= \frac{d}{dt} \langle (x(t) - \tilde{x}(t))', x^*(t) \rangle - \frac{d}{dt} \langle x^{*'}(t), x(t) - \tilde{x}(t) \rangle.$$

Thus by elementary property of the definite integrals, we can compute the integral on the right-hand side of Eq. (31):

$$\begin{aligned} & \int_0^1 [\langle (x(t) - \tilde{x}(t))'', x^*(t) \rangle - \langle x^{*''}(t), x(t) - \tilde{x}(t) \rangle] dt \\ &= \langle x'(1) - \tilde{x}'(1), x^*(1) \rangle - \langle x'(0) - \tilde{x}'(0), x^*(0) \rangle \\ & \quad - \langle x^{*'}(1), x(1) - \tilde{x}(1) \rangle + \langle x^{*'}(0), x(0) - \tilde{x}(0) \rangle. \end{aligned} \quad (32)$$

Then substituting Eq. (32) into Eq. (31), we have

$$\begin{aligned} 0 &\geq \langle x'(1) - \tilde{x}'(1), x^*(1) \rangle - \langle x'(0) - \tilde{x}'(0), x^*(0) \rangle \\ & \quad - \langle v^*(1) + x^{*'}(1), x(1) - \tilde{x}(1) \rangle + \langle v^*(0) + x^{*'}(0), x(0) - \tilde{x}(0) \rangle. \end{aligned} \quad (33)$$

Now, remember that $x(\cdot), \tilde{x}(\cdot)$ are feasible trajectories and $x(0) = \tilde{x}(0) = x_0$ and $x'(0) = \tilde{x}'(0)$.

$= x_1$. Then it follows from Eq. (33) that

$$0 \geq \langle x'(1) - \tilde{x}'(1), x^*(1) \rangle - \langle v^*(1) + x^{*'}(1), x(1) - \tilde{x}(1) \rangle. \quad (34)$$

Now, thanks to the transversality conditions Eq. (c₁) at the endpoint $t = 1$, we can rewrite

$$\begin{aligned} & g(x(1), x'(1)) - g(\tilde{x}(1), \tilde{x}'(1)) \geq \langle v^*(1) + x^{*'}(1) + x^*(1), x(1) - \tilde{x}(1) \rangle \\ & + \langle x^{*'}(1) - x^*(1), x'(1) - \tilde{x}'(1) \rangle, \quad x^*(1) \in K_{M_0}^*(\tilde{x}(1)), \quad x^{*'}(1) \in K_{M_1}^*(\tilde{x}'(1)) \end{aligned}$$

or, in other words

$$\begin{aligned} & g(x(1), x'(1)) - g(\tilde{x}(1), \tilde{x}'(1)) \\ & \geq \langle v^*(1) + x^{*'}(1), x(1) - \tilde{x}(1) \rangle - \langle x^*(1), x'(1) - \tilde{x}'(1) \rangle \end{aligned} \quad (35)$$

Thus, summing the inequalities Eqs. (34) and (35) for all feasible trajectories $x(\cdot)$, satisfying the initial conditions $x(0) = x_0$ and $x'(0) = x_1$ and endpoint constraints $x(1) \in M_0, x'(1) \in M_1$, we have the needed inequality:

$$g(x(1), x'(1)) - g(\tilde{x}(1), \tilde{x}'(1)) \geq 0 \text{ or } g(x(1), x'(1)) \geq g(\tilde{x}(1), \tilde{x}'(1)). \quad \square$$

6. Conclusion

According to proposed method, the problem with the differential inclusions described by polynomial linear differential operators is investigated. Obviously, this problem is an important generalization of problems with first-order differential inclusions. Thus, sufficient conditions of optimality for such problems are deduced. Here the existence of nonfunctional initial point or endpoint constraints generates different kinds of transversality conditions. Besides, there can be no doubt that investigations of optimality conditions of problems with second- and fourth-order Sturm-Liouville type differential inclusions can play an important role in the development of modern optimization and there is every reason to believe that this role

will be even more significant in the future. Thus, the suggested problem with linear differential operators and variable coefficients can be used in various forms in applied problems.



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Spectral Analysis and Numerical Investigation of a Flexible Structure with Nonconservative Boundary Data

Marianna A. Shubov and Laszlo P. Kindrat

Abstract

Analytic and numerical results of the Euler-Bernoulli beam model with a two-parameter family of boundary conditions have been presented. The co-diagonal matrix depending on two control parameters (k_1 and k_2) relates a two-dimensional input vector (the shear and the moment at the right end) and the observation vector (the time derivatives of displacement and the slope at the right end). The following results are contained in the paper. First, high accuracy numerical approximations for the eigenvalues of the discretized differential operator (the dynamics generator of the model) have been obtained. Second, the formula for the number of the deadbeat modes has been derived for the case when one control parameter, k_1 , is positive and another one, k_2 , is zero. It has been shown that the number of the deadbeat modes tends to infinity, as $k_1 \rightarrow 1^+$ and $k_2 = 0$. Third, the existence of *double* deadbeat modes and the asymptotic formula for such modes have been proven. Fourth, numerical results corroborating all analytic findings have been produced by using Chebyshev polynomial approximations for the continuous problem.

Keywords: matrix differential operator, eigenvalues, Chebyshev polynomials, numerical scheme, boundary control

1. Introduction

The present paper is concerned with the spectral analysis and numerical investigation of the eigenvalues of the Euler-Bernoulli beam model. The beam is clamped at the left end and subject to linear feedback-type conditions with a non-dissipative feedback matrix [1, 2]. Depending on the boundary parameters k_1 and k_2 , the model can be either conservative, dissipative, or completely non-dissipative. We focus on the non-dissipative case, i.e., when the energy of a vibrating system is not a decreasing (or nonincreasing) function of time. In our approach, the initial-boundary value problem describing the beam dynamics is reduced to the first order in time evolution equation in the state Hilbert space \mathcal{H} . The evolution of the system is completely determined by the dynamics generator \mathcal{L}_{k_1, k_2} , which is an unbounded non-self-adjoint matrix differential operator (see Eqs. (2), (3), and (8)).

The eigenmodes and the mode shapes of the flexible structure are defined as the eigenvalues (up to a multiple i) and the generalized eigenvectors of \mathcal{L}_{k_1, k_2} .

Based on the results of [1, 2], the dynamics generator has a purely discrete spectrum, whose location on the complex plane is determined by the controls k_1 and k_2 . Having in mind the practical applications of the asymptotic formulas [3–5], we discuss the case of $k_1 \geq 0$ and $k_2 \geq 0$, such that $|k_1| + |k_2| > 0$ (see Proposition 2). As shown in [2], even though the operator \mathcal{L}_{k_1, k_2} is non-dissipative, for the case $k_1 > 0$ and $k_2 = 0$ (or $k_1 = 0$ and $k_2 > 0$), the entire set of eigenvalues is located in the closed upper half of the complex plane \mathbb{C} , which means that all eigenmodes are stable or neutrally stable. (We recall that to obtain an elastic mode from an eigenvalues of \mathcal{L}_{k_1, k_2} , one should multiply the eigenvalue by a factor i).

In the paper we address the question of accuracy of the asymptotic formulas for the eigenvalues. *Namely, under what conditions the leading asymptotic terms in formulas (20) and (21) can be used for practical estimation of the actual frequencies of the flexible beam?* Numerical simulations show that the accuracy of the asymptotic formulas is really high; the leading asymptotic terms can be used by practitioners almost immediately, i.e., almost from the first vibrational mode. The second question is concerned with the role of the *deadbeat modes*. A deadbeat mode is a purely negative elastic mode that generates a solution of the evolution equation exponentially decaying in time. The deadbeat modes are important in engineering applications. As we prove in the paper, when the boundary parameter k_1 is close to 1 (while $k_2 = 0$), the number of the deadbeat modes is so large that the corresponding mode shapes become important for the description of the beam dynamics. More precisely, the number of deadbeat modes tends to infinity as $k_1 \rightarrow 1^+$.

We have also shown that there exists a sequence of values of the parameter k_1 , i.e., $\{k_1^{(n)}\}_{n=1}^{\infty}$, such that for each $k_1 = k_1^{(n)}$ there exist a finite number of deadbeat modes and each corresponds to a *double eigenvalue* of the dynamics generator \mathcal{L}_{k_1, k_2} . For each value $k_1^{(n)}$, the operator \mathcal{L}_{k_1, k_2} has a two-dimensional root subspace spanned by an eigenvector and an associate vector. This result means that for a double deadbeat mode (corresponding to $k_1^{(n)}$), there exists a mode shape and an associate mode shape. This fact indicates that for some values of k_1 and k_2 , there exists a significant number of associate vectors of \mathcal{L}_{k_1, k_2} . Therefore, if one can prove that the set of the generalized eigenvectors (eigenvectors and associate vectors together) forms an unconditional basis for the state space, then construction of the bi-orthogonal basis [6] would be a more complicated problem than for the case when no associate vectors exist.

Finally, we mention that the feedback control of beams is a well-studied area [6], with multiple applications to the control of robotic manipulators, long and slender aircraft wings, propeller blades, large space structure [7, 8], and the dynamics of carbon nanotubes [9]. The analysis of a classical beam model with nonstandard feedback control law that originated in engineering literature [4, 10–12] may be of interest for both analysts and practitioners.

This paper is organized as follows. In Section 1 we formulate the initial-boundary value problem for the Euler-Bernoulli beam model. In Section 2, we reformulate the problem as an evolution equation in the Hilbert space of Cauchy data (the energy space). The dynamics generator \mathcal{L}_{k_1, k_2} , which is a *non-self-adjoint matrix differential operator depending on two parameters, k_1 and k_2 , is the main object of interest*. The eigenvalues and the generalized eigenvectors of \mathcal{L}_{k_1, k_2} correspond to the modes and the mode shapes of the beam. We also give numerical approximations and graphical representations of the eigenvalues of a discrete approximation of the main operator (see **Tables 1** and **2** and **Figures 1** and **2**). In Section 3,

$k_1 = 0, k_2 = 0.5, EI = 1, \rho = 0.1, N = 64, \varepsilon_f = 10^{-20}$					
No.	Numerical	Analytic	No.	Numerical	Analytic
1.	$-7988.1 + 237.00i$	$-7988.1 + 237.00i$	18.	$28.860 + 13.515i$	$29.453 + 14.812i$
2.	$-7020.6 + 222.19i$	$-7020.6 + 222.19i$	19.	$123.16 + 29.723i$	$123.08 + 29.625i$
3.	$-6115.5 + 207.37i$	$-6115.5 + 207.37i$	20.	$279.13 + 44.431i$	$279.14 + 44.437i$
4.	$-5272.8 + 192.56i$	$-5272.8 + 192.56i$	21.	$497.61 + 59.250i$	$497.61 + 59.250i$
5.	$-4492.5 + 177.75i$	$-4492.5 + 177.75i$	22.	$778.50 + 74.062i$	$778.50 + 74.062i$
6.	$-3774.7 + 162.94i$	$-3774.7 + 162.94i$	23.	$1121.8 + 88.875i$	$1121.8 + 88.875i$
7.	$-3119.3 + 148.12i$	$-3119.3 + 148.12i$	24.	$1527.6 + 103.69i$	$1527.6 + 103.69i$
8.	$-2526.3 + 133.31i$	$-2526.3 + 133.31i$	25.	$1995.7 + 118.50i$	$1995.7 + 118.50i$
9.	$-1995.7 + 118.50i$	$-1995.7 + 118.50i$	26.	$2526.3 + 133.31i$	$2526.3 + 133.31i$
10.	$-1527.6 + 103.69i$	$-1527.6 + 103.69i$	27.	$3119.3 + 148.12i$	$3119.3 + 148.12i$
11.	$-1121.8 + 88.875i$	$-1121.8 + 88.875i$	28.	$3774.7 + 162.94i$	$3774.7 + 162.94i$
12.	$-778.5 + 74.062i$	$-778.5 + 74.062i$	29.	$4492.5 + 177.75i$	$4492.5 + 177.75i$
13.	$-497.61 + 59.250i$	$-497.61 + 59.250i$	30.	$5272.8 + 192.56i$	$5272.8 + 192.56i$
14.	$-279.13 + 44.431i$	$-279.14 + 44.437i$	31.	$6115.5 + 207.37i$	$6115.5 + 207.37i$
15.	$-123.16 + 29.723i$	$-123.08 + 29.625i$	32.	$7020.6 + 222.19i$	$7020.6 + 222.19i$
16.	$-28.860 + 13.515i$	$-29.453 + 14.812i$	33.	$7988.1 + 237.00i$	$7988.1 + 237.00i$
17.	$-2.2 \cdot 10^{-17} + 4.6007i$				

Table 1.

Approximations of the eigenvalues for the discrete and “continuous” operators ($K > 1$).

we study the deadbeat modes and derive the estimates for the number of the deadbeat modes from below and above for different values of the boundary parameters (see **Figure 5**). Section 4 is concerned with the asymptotic approximation for the set of double deadbeat modes (see **Tables 3 and 4** and **Figures 6 and 7**). In Section 5, we outline the numerical scheme used for the spectral analysis of the finite-dimensional approximation of the dynamics generator.

1.1 The initial-boundary value problem for the Euler-Bernoulli beam model of a unit length

The Lagrangian of the system is defined by [10, 11]

$$\frac{1}{2} \int_0^1 [q(x)A(x)h_t^2(x, t) - E(x)I(x)h_{xx}^2(x, t)] dx, \quad (1)$$

where $h(x, t)$ is the transverse deflection, $E(x)$ is the modulus of elasticity, $I(x)$ is the area moment of inertia, $q(x)$ is the linear density, and $A(x)$ is the cross-sectional area of the beam.

Assuming that the beam is clamped at the left end ($x = 0$) and free at the right end ($x = 1$), and applying Hamilton’s variational principle to the action functional defined by (1), we obtain the equation of motion

$$q(x)A(x)h_{tt}(x, t) + (E(x)I(x)h_{xx}(x, t))_{xx} = 0, \quad 0 \leq x \leq 1, \quad t > 0, \quad (2)$$

$k_1 = 1.3, k_2 = 1.2, EI = 10, \rho = 0.1, N = 64, \epsilon_f = 10^{-20}$

No.	Numerical	Analytic	No.	Numerical	Analytic
1.	-25266 - 229.07i	-25266 - 229.07i	18.	99.467 - 19.816i	98.177 - 14.317i
2.	-22206 - 214.76i	-22206 - 214.76i	19.	394.17 - 28.149i	394.26 - 28.634i
3.	-19344 - 200.44i	-19344 - 200.44i	20.	887.75 - 42.983i	887.75 - 42.951i
4.	-16679 - 186.12i	-16679 - 186.12i	21.	1578.6 - 57.267i	1578.6 - 57.268i
5.	-14212 - 171.81i	-14212 - 171.81i	22.	2466.9 - 71.586i	2466.9 - 71.585i
6.	-11942 - 157.49i	-11942 - 157.49i	23.	3552.5 - 85.902i	3552.5 - 85.903i
7.	-9869.1 - 143.17i	-9869.1 - 143.17i	24.	4835.6 - 100.22i	4835.6 - 100.22i
8.	-7993.9 - 128.85i	-7993.9 - 128.85i	25.	6316.0 - 114.54i	6316.0 - 114.54i
9.	-6316.0 - 114.54i	-6316.0 - 114.54i	26.	7993.9 - 128.85i	7993.9 - 128.85i
10.	-4835.6 - 100.22i	-4835.6 - 100.22i	27.	9869.1 - 143.17i	9869.1 - 143.17i
11.	-3552.5 - 85.902i	-3552.5 - 85.903i	28.	11942 - 157.49i	11942 - 157.49i
12.	-2466.9 - 71.586i	-2466.9 - 71.585i	29.	14212 - 171.81i	14212 - 171.81i
13.	-1578.6 - 57.267i	-1578.6 - 57.268i	30.	16679 - 186.12i	16679 - 186.12i
14.	-887.75 - 42.983i	-887.75 - 42.951i	31.	19344 - 200.44i	19344 - 200.44i
15.	-394.17 - 28.149i	-394.26 - 28.634i	32.	22206 - 214.76i	22206 - 214.76i
16.	-99.467 - 19.816i	-98.177 - 14.317i	33.	25266 - 229.07i	25266 - 229.07i
17.	$1.5 \cdot 10^{-17} + 7.7256i$				

Table 2. Approximations of the eigenvalues for the discrete and “continuous” operators ($K < -1$).

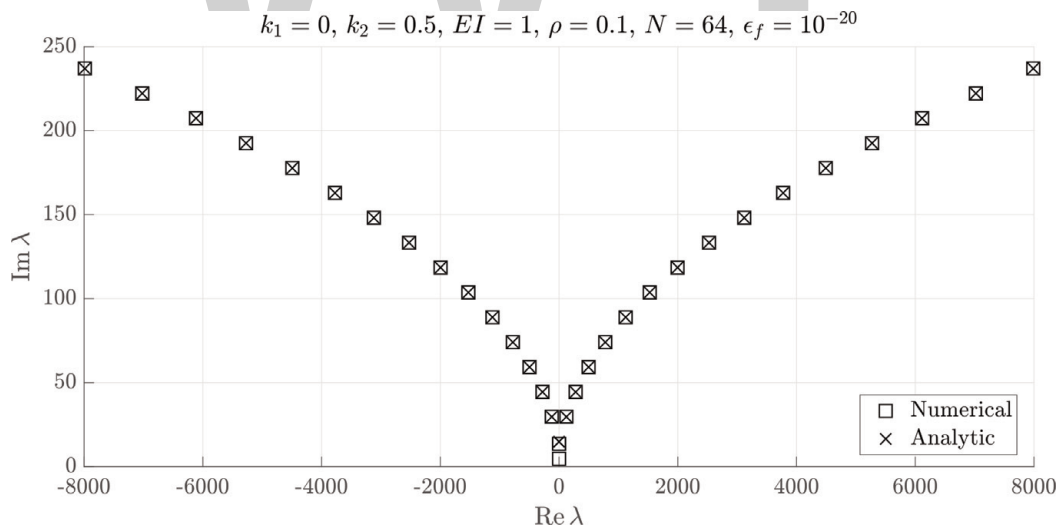


Figure 1. Graphical representation of the eigenvalues of the discrete and “continuous” operators ($K > 1$).

and the boundary conditions

$$h(0, t) = h_x(0, t) = 0 \quad \text{and} \quad M(1, t) = Q(1, t) = 0, \quad (3)$$

where $M(x, t)$ and $Q(x, t)$ are the moment and the shear, respectively [10]:

$$M(x, t) = E(x)I(x)h_{xx}(x, t) \quad \text{and} \quad Q(x, t) = M_x(x, t). \quad (4)$$

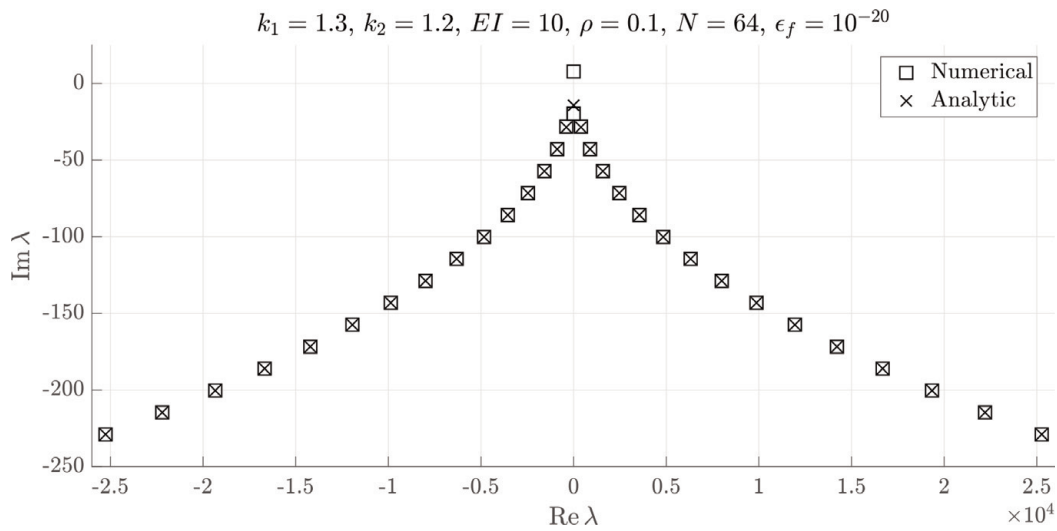


Figure 2. Graphical representation of the eigenvalues of the discrete and “continuous” operators ($K < -1$).

$k_2 = 0, EI = 1, \rho = 1, N = 64, \epsilon_f = 10^{-30}$			
No.	$k_1 = 1 + 10^{-4}$	$k_1 = 1 + 10^{-7}$	$k_1 = 1 + 10^{-10}$
1.	$-222.22 + 155.56i$	$-176.06 + 264.07i$	$-106.90 + 372.41i$
2.	$-133.38 + 124.45i$	$-87.723 + 211.27i$	$-2.4378 \cdot 10^{-16} + 254.44i$
3.	$-64.540 + 93.396i$	$-3.9609 \cdot 10^{-18} + 162.37i$	$-3.0717 \cdot 10^{-17} + 123.66i$
4.	$-9.8081 + 58.559i$	$-5.7977 \cdot 10^{-19} + 116.23i$	$-1.3012 \cdot 10^{-17} + 4.9349i$
5.	$-7.7725 \cdot 10^{-21} + 5.0488i$	$-6.3661 \cdot 10^{-20} + 44.182i$	$-8.9232 \cdot 10^{-18} + 44.421i$
6.	$4.9365 \cdot 10^{-21} + 38.994i$	$-6.0066 \cdot 10^{-20} + 4.9383i$	$8.9143 \cdot 10^{-18} + 44.406i$
7.	$7.8007 \cdot 10^{-21} + 4.8257i$	$6.0114 \cdot 10^{-20} + 4.9313i$	$1.3012 \cdot 10^{-17} + 4.9347i$
8.	$9.8081 + 58.559i$	$6.6801 \cdot 10^{-20} + 44.651i$	$2.9907 \cdot 10^{-17} + 123.09i$
9.	$64.540 + 93.396i$	$2.9381 \cdot 10^{-18} + 139.20i$	$1.1286 \cdot 10^{-16} + 234.06i$
10.	$133.38 + 124.45i$	$87.723 + 211.27i$	$3.2280 \cdot 10^{-16} + 315.13i$
11.	$222.22 + 155.56i$	$176.06 + 264.07i$	$106.90 + 372.41i$

Table 3. Eigenvalues closest to the imaginary axis as $k_1 \rightarrow 1^+$.

Now we replace the free right-end conditions from Eq. (3) with the following boundary feedback control law [2, 4]. Define the input and the output as

$$U(t) = [-Q(1, t), M(1, t)]^T \quad \text{and} \quad Y(t) = [h_t(1, t), h_{xt}(1, t)]^T, \quad (5)$$

where T stands for transposition. The feedback control law is given by

$$U(t) = KY(t), \quad (6)$$

where K is the 2×2 feedback matrix. We select

$$K = \text{codiag}(-k_2, -k_1), \quad k_1, k_2 \geq 0, \quad (7)$$

$k_2 = 0, EI = 1, \rho = 1, N = 64, \epsilon_f = 10^{-30}$			
#	$k_1 = 1 - 10^{-4}$	$k_1 = 1 - 10^{-7}$	$k_1 = 1 - 10^{-10}$
1.	$-175.34 + 140.01i$	$-129.20 + 237.56i$	$-60.143 + 338.80i$
2.	$-96.394 + 108.84i$	$-50.206 + 186.90i$	$-8.6602 + 240.01i$
3.	$-36.896 + 78.778i$	$-8.0431 + 121.15i$	$-0.28608 + 123.37i$
4.	$-6.0769 + 42.171i$	$-0.23455 + 44.410i$	$-0.0074192 + 44.413i$
5.	$-0.11141 + 4.9324i$	$-0.0035253 + 4.9348i$	$-0.00011148 + 4.9348i$
6.	$0.11141 + 4.9324i$	$0.0035253 + 4.9348i$	$0.00011148 + 4.9348i$
7.	$6.0769 + 42.171i$	$0.23455 + 44.410i$	$0.0074192 + 44.413i$
8.	$36.896 + 78.778i$	$8.0431 + 121.15i$	$0.28608 + 123.37i$
9.	$96.394 + 108.84i$	$50.206 + 186.90i$	$8.6602 + 240.01i$
10.	$175.34 + 140.01i$	$129.20 + 237.56i$	$60.143 + 338.80i$

Table 4.
Eigenvalues closest to the imaginary axis as $k_1 \rightarrow 1^-$.

with k_1, k_2 being the control parameters. The feedback (6) can be written as

$$E(1)I(1)h_{xx}(1, t) = -k_1 h_t(1, t) \quad \text{and} \quad (E(x)I(x)h_{xx}(x, t))_x|_{x=1} = k_2 h_{xt}(1, t). \quad (8)$$

Finally, we arrive to the following initial-boundary value problem: the equation of motion (2), the boundary conditions (3), and the standard initial conditions $h(x, 0) = h_0(x)$, $h_t(x, 0) = h_1(x)$.

Notice that the choice of a feedback matrix K defines whether the system is dissipative or not. Indeed, let $\mathcal{E}(t)$ be the energy of the system, defined by representation (1). Evaluating $\mathcal{E}_t(t)$ on the solutions of Eq. (2) satisfying the left-end conditions from Eqs. (3), we obtain

$$\begin{aligned} \mathcal{E}_t(t) &= \int_0^1 [\varrho(x)A(x)h_t(x, t)h_{tt}(x, t) + E(x)I(x)h_{xx}(x, t)h_{xxt}(x, t)]x \\ &= -(E(x)I(x)h_{xx}(xt))_x h_t(xt)|_{x=1} + E(1)I(1)h_{xx}(1, t)h_{xt}(1, t). \end{aligned} \quad (9)$$

Taking into account Eqs. (4) and (6), we represent the right-hand side of Eq. (9) as the dot product in \mathbb{R}^2 :

$$\mathcal{E}_t(t) = -Q(1, t)h_t(1, t) + M(1, t)h_{xt}(1, t) = U(t) \cdot Y(t) = KY(t) \cdot Y(t). \quad (10)$$

With the choice of K as in Eq. (7), we have

$$\mathcal{E}_t(t) = \begin{bmatrix} 0 & -k_2 \\ -k_1 & 0 \end{bmatrix} \begin{bmatrix} 2h_t(1, t) \\ h_{xt}(1, t) \end{bmatrix} \cdot \begin{bmatrix} 2h_t(1, t) \\ h_{xt}(1, t) \end{bmatrix} = -(k_1 + k_2)h_t(1, t)h_{xt}(1, t). \quad (11)$$

Thus the system is not dissipative for all nonnegative values of k_1 and k_2 .

2. Operator form of the problem

In what follows, we incorporate the cross-sectional area $A(x)$ into the density, write $\rho(x)$ instead of $\varrho(x)A(x)$, and also abbreviate $EI(x) \equiv E(x)I(x)$. Let \mathcal{H} be the

Hilbert space of two-component vector functions $U(x) = [u_0(x), u_1(x)]^T$ equipped with the following norm:

$$\|U\|_{\mathcal{H}}^2 = \frac{1}{2} \int_0^1 \left[EI(x) |u_0'(x)|^2 + \rho(x) |u_1(x)|^2 \right] dx. \tag{12}$$

Assuming that $EI, \rho \in C^2(0, 1)$ are positive functions, we obtain that the closure of smooth functions $U(x) = [u_0(x), u_1(x)]^T$ satisfying $u_0(0) = u_0'(0) = 0$ will produce the energy space $\mathcal{H} = H_0^2(0, 1) \times L^2(0, 1)$. Here $H_0^2(0, 1) = \{u \in H^2(0, 1) : u(0) = u'(0) = 0\}$, and the equality of function spaces is understood in the sense of a Hilbert-space isomorphism.

Problem (2) with conditions (3) can be represented as the time evolution problem:

$$U_t(x, t) = i(\mathcal{L}_{k_1, k_2} U)(x, t) \quad \text{and} \quad U(x, 0) = [u_0(x), u_1(x)]^T, \tag{13}$$

where $0 \leq x \leq 1, t \geq 0$. The dynamics generator \mathcal{L}_{k_1, k_2} is given by the following matrix differential expression:

$$\mathcal{L}_{k_1, k_2} = -i \begin{bmatrix} 0 & 1 \\ -\frac{1}{\rho(x)} \frac{\partial^2}{\partial x^2} \left(EI(x) \frac{\partial^2}{\partial x^2} \right) & 0 \end{bmatrix}, \tag{14}$$

defined on the domain

$$\mathcal{D}(\mathcal{L}_{k_1, k_2}) = \{U = (u_0, u_1)^T \in \mathcal{H} : u_0 \in H^4(0, 1), u_1 \in H_0^2(0, 1); u_1(0) = u_1'(0) = 0; EI(1)u_0''(1) = -k_1u_1(1), (EI(x)u_0''(x))'|_{x=1} = k_2u_1'(1)\}. \tag{15}$$

For any $(k_1, k_2) \in \mathbb{R}^2$, the adjoint operator \mathcal{L}_{k_1, k_2}^* [13] is given by

$$\mathcal{L}_{k_1, k_2}^* = \mathcal{L}_{-k_2, -k_1}, \tag{16}$$

i.e., \mathcal{L}_{k_1, k_2}^* is defined by the same differential expression (14) on the domain described in Eq. (15), where k_1 and k_2 are replaced by $(-k_2)$ and $(-k_1)$, respectively. It follows from Eq. (16) that $\mathcal{L}_{0, 0}$ is self-adjoint in \mathcal{H} and thus $\mathcal{L}_{0, 0}$ is the dynamics generator of the clamped-free beam model. For the reader's convenience, we summarize the properties of \mathcal{L}_{k_1, k_2} from [1, 2] needed for the present work.

Proposition 1:

1. \mathcal{L}_{k_1, k_2} is an unbounded operator with compact resolvent, whose spectrum consists of a countable set of normal eigenvalues (i.e., isolated eigenvalues, each of finite algebraic multiplicity [6, 13]).
2. For each $(k_1, k_2) \in \mathbb{R}^2, |k_1| + |k_2| > 0$, the operator \mathcal{L}_{k_1, k_2} is a rank-two perturbation of the self-adjoint operator $\mathcal{L}_{0, 0}$ in the sense that the operators $\mathcal{L}_{k_1, k_2}^{-1}$ and $\mathcal{L}_{0, 0}^{-1}$ exist and are related by the rule

$$\mathcal{L}_{k_1, k_2}^{-1} = \mathcal{L}_{0, 0}^{-1} + \mathcal{T}_{k_1, k_2}, \tag{17}$$

where \mathcal{T}_{k_1, k_2} is a rank-two operator. A similar decomposition is valid for the adjoint operator, i.e.,

$$\left(\mathcal{L}_{k_1, k_2}^{-1}\right)^* = \mathcal{L}_{0, 0}^{-1} + \mathcal{T}_{k_1, k_2}^*, \quad \mathcal{T}_{k_1, k_2}^* = \mathcal{T}_{-k_2, -k_1}. \quad (18)$$

From now on, we assume that the structural parameters are constant. In the case of variable parameters, the spectral asymptotics will have the same leading terms and remainder terms depending on parameter smoothness.

Proposition 2: Assume that $k_1, k_2 > 0$ and $k_1 k_2 \neq EI\rho$. Let

$$\mathcal{K} = \frac{k_1 + k_2}{A - k_1 k_2 / A}, \quad A = \sqrt{EI\rho}, \quad \text{and} \quad |\mathcal{K}| \neq 1. \quad (19)$$

The following asymptotic approximations for the eigenvalues λ_n (as $|n| \rightarrow \infty$) of the operator \mathcal{L}_{k_1, k_2} hold:

1. If $1 < |\mathcal{K}| < \infty$, then for $|n| \rightarrow \infty$ one has

$$\lambda_n = \text{sign}(\mathcal{K}n) \sqrt{\frac{EI}{\rho}} \left[(\pi n)^2 - \frac{1}{4} \ln^2 \left(\frac{\mathcal{K} + 1}{\mathcal{K} - 1} \right) + i\pi n \ln \left(\frac{\mathcal{K} + 1}{\mathcal{K} - 1} \right) \right] + O(ne^{-\pi|n|}). \quad (20)$$

2. If $0 < \mathcal{K} < 1$, then for $n \rightarrow \infty$ one has

$$\lambda_n = \text{sign}(\mathcal{K}n) \sqrt{\frac{EI}{\rho}} \left[\left(\frac{2n+1}{2} \pi \right)^2 - \frac{1}{4} \ln^2 \left(\frac{\mathcal{K} + 1}{\mathcal{K} - 1} \right) + i\pi \left(\frac{2n+1}{2} \right) \ln \left(\frac{\mathcal{K} + 1}{\mathcal{K} - 1} \right) \right] + O(ne^{-\pi|n|}). \quad (21)$$

First of all, we address the question of accuracy of the asymptotic formulas (20) and (21). By its nature, formula (20) (as well as formula (21)) means that for any small $\varepsilon > 0$, one can find a positive integer N , such that all eigenvalues λ_n with $|n| \geq N + 1$ satisfy the estimate

$$\left| \lambda_n - \text{sign}(\mathcal{K}n) \sqrt{\frac{EI}{\rho}} \left[(\pi n)^2 - \frac{1}{4} \ln^2 \left(\frac{\mathcal{K} + 1}{\mathcal{K} - 1} \right) + i\pi n \ln \left(\frac{\mathcal{K} + 1}{\mathcal{K} - 1} \right) \right] \right| \leq \varepsilon \quad (22)$$

for the case when $1 < |\mathcal{K}| < \infty$ and

$$\left| \lambda_n - \text{sign}(\mathcal{K}n) \sqrt{\frac{EI}{\rho}} \left[\left(\frac{2n+1}{2} \pi \right)^2 - \frac{1}{4} \ln^2 \left(\frac{\mathcal{K} + 1}{\mathcal{K} - 1} \right) + i\pi \left(\frac{2n+1}{2} \right) \ln \left(\frac{\mathcal{K} + 1}{\mathcal{K} - 1} \right) \right] \right| \leq \varepsilon \quad (23)$$

for the case when $0 < |\mathcal{K}| < 1$. The following important question holds: *From which index N can the eigenvalues be approximated by the leading asymptotic terms with acceptable accuracy?* In other words, can one claim that the asymptotic formulas (20) and (21) are valuable to practitioners, or are they just important mathematical results of the spectral analysis?

The results of numerical simulations (see **Tables 1** and **2** and **Figures 1** and **2**) show that the asymptotic formulas are indeed quite accurate. That is, if one places on the complex plane the numerically produced sets of the eigenvalues, then the theoretically predicted distribution of eigenvalues can be seen almost immediately. To obtain these results, we used the numerical procedure based on Chebyshev polynomial approximations [14–16], as outlined in Section 5.

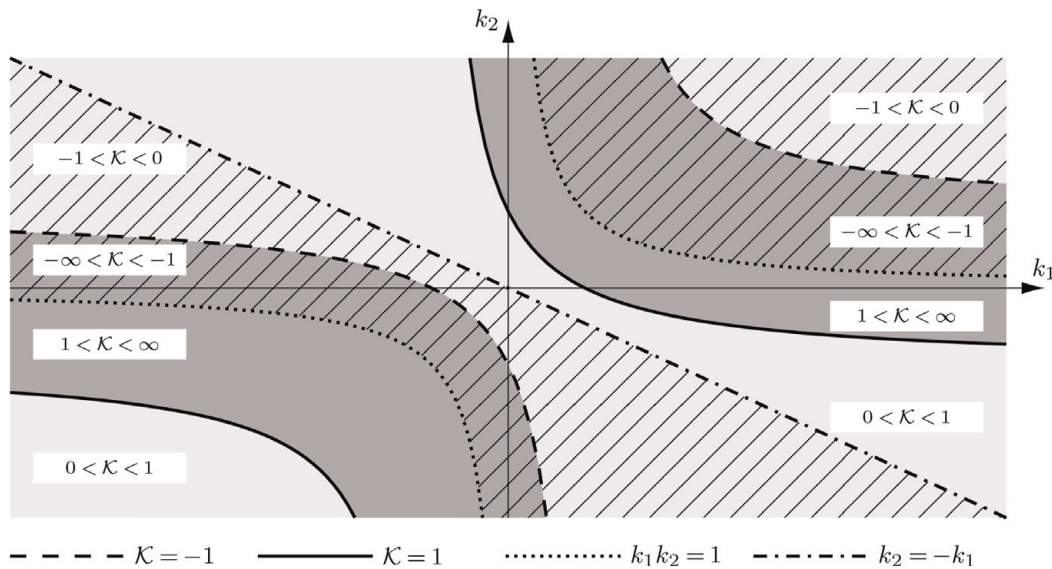


Figure 3.
Regions of \mathcal{K} on the k_1, k_2 -plane, $A = \sqrt{EI\rho} = 1$.

In **Figures 1** and **2**, we represent the graphical distribution of the eigenvalues corresponding to the discretized operator (“numerical” eigenvalues) and the leading asymptotic terms from Eqs. (20) and (21) (“analytic” eigenvalues). In **Tables 1** and **2**, the numerical values of the corresponding graphical points on **Figures 1** and **2** are listed. We have used the following notations: $N = 64$ is the number of grid points on $[0, 1]$, and ε_f is the filtering parameter as described in Eq. (69). It can be easily seen from the graphs and tables that the two sets of data coincide almost immediately, i.e., the leading asymptotic terms in the approximations are very close to the numerically approximated eigenvalues.

Figure 3 shows the sub-domains of the k_1, k_2 -plane, which correspond to different intervals for the values of \mathcal{K} defined by Eq. (19). On the sub-domain with dark gray color \mathcal{K} such that $|\mathcal{K}| > 1$, i.e., to evaluate the asymptotic approximation for the eigenvalues, one needs formula (20), while on the complementary sub-domain, one needs formula (21).

3. The deadbeat modes

An eigenvalue λ_n of the dynamics generator \mathcal{L}_{k_1, k_2} is called a deadbeat mode if $\lambda_n = i\beta_n$, $\beta_n > 0$. If the corresponding eigenfunction is $\Phi_n(x)$, then the evolution problem (13) has a solution given in the form $e^{i\lambda_n t} \Phi_n(x) = e^{-\beta_n t} \Phi_n(x)$, which tends to zero without any oscillation.

As shown in paper [2], for the case when one of the control parameters is zero and the other one is positive, the entire set of the eigenvalues is located in the closed upper half plane. This result is not obvious since the operator is not dissipative; in fact, it requires a fairly nontrivial proof. However, due to this fact, we assume that any deadbeat mode can be given in the form $i\beta$, with $\beta > 0$. To deal with the deadbeat modes analytically, we rewrite the spectral equation $(\mathcal{L}_{k_1, k_2} \Phi)(x) = \lambda \Phi(x)$ in the form of an equivalent problem for an operator pencil [17] as

$$\begin{aligned} EI\varphi''''(x) &= \lambda^2 \rho \varphi(x), & \varphi(0) &= \varphi'(0) = 0, \\ EI\varphi''(1) &= -i\lambda k_1 \varphi(1), & EI\varphi''(1) &= i\lambda k_2 \varphi'(1). \end{aligned} \quad (24)$$

If λ_n and $\varphi_n(x)$ are an eigenvalue and eigenfunction of the pencil (24), then λ_n is also an eigenvalue of \mathcal{L}_{k_1, k_2} with the eigenfunction $\Phi_n(x) = \left[\frac{1}{i\lambda_n} \varphi_n(x), \varphi_n(x) \right]^T$.

To solve problem (24), we first redefine the spectral and control parameters to eliminate ρ and EI from Eq. (24). We define $\tilde{\lambda}$, \tilde{k}_1 , and \tilde{k}_2 by $\lambda = \sqrt{EI/\rho\tilde{\lambda}}$ and $\tilde{k}_j = \sqrt{EI\rho}k_j, j = 1, 2$. Substituting these relations into Eq. (24) and eliminating the “tilde,” we obtain the following Sturm-Liouville eigenvalue problem:

$$\varphi''''(x) = \lambda^2\varphi(x), \quad \varphi(0) = \varphi'(0) = 0, \quad \varphi''(1) = -i\lambda k_1\varphi(1), \quad \varphi'''(1) = i\lambda k_2\varphi'(1). \tag{25}$$

The solution of Eq. (25) satisfying the left-end boundary conditions $\varphi(0) = \varphi'(0) = 0$ can be written in the form

$$\varphi(\lambda, x) = \mathcal{A}(\lambda) \left[\cosh(\sqrt{\lambda}x) - \cos(\sqrt{\lambda}x) \right] + \mathcal{B}(\lambda) \left[\sinh(\sqrt{\lambda}x) - \sin(\sqrt{\lambda}x) \right]. \tag{26}$$

Substituting formula (26) into the right-end boundary conditions of Eq. (25), one gets a system for the coefficients $\mathcal{A}(\lambda)$ and $\mathcal{B}(\lambda)$:

$$\begin{aligned} \mathcal{A}(\lambda) [(1 + ik_1)\cosh \sqrt{\lambda} - (1 - ik_1)\cos \sqrt{\lambda}] + \\ \mathcal{B}(\lambda) [(1 + ik_1)\sinh \sqrt{\lambda} - (1 - ik_1)\sin \sqrt{\lambda}] &= 0, \\ \mathcal{A}(\lambda) [(1 - ik_2)\sinh \sqrt{\lambda} - (1 + ik_2)\sin \sqrt{\lambda}] + \\ \mathcal{B}(\lambda) [(1 - ik_2)\cosh \sqrt{\lambda} + (1 + ik_2)\cos \sqrt{\lambda}] &= 0. \end{aligned} \tag{27}$$

Let $\Delta(\lambda)$ be the determinant of the matrix of coefficients for $\mathcal{A}(\lambda)$ and $\mathcal{B}(\lambda)$ in Eqs. (27). System (27) has nontrivial solutions if and only if $\Delta(\lambda) = 0$, i.e.,

$$(1 + k_1k_2) + (1 - k_1k_2)\cosh \sqrt{\lambda} \cos \sqrt{\lambda} + i(k_1 + k_2)\sinh \sqrt{\lambda} \sin \sqrt{\lambda} = 0. \tag{28}$$

Theorem 1: The following results hold in the case when $k_1 > 0$ and $k_2 = 0$. Similar results hold in the case when $k_1 = 0$ and $k_2 > 0$.

1. For $0 < k_1 < 1$, the deadbeat modes do not exist.
2. For $k_1 = 1$, there exist infinitely many deadbeat modes given explicitly by

$$\lambda_n = \mu_n^2, \quad \mu_n = x_n(1 + i), \quad x_n = \frac{\pi}{2}(2n + 1), \quad n = 0, 1, 2, \dots \tag{29}$$

3. For any $k_1 > 1$, there exist a finite number $\mathcal{N}(k_1)$ of deadbeat modes. Each mode has the form $\lambda = \mu^2, \mu = x(1 + i)$, where x is a root of the function

$$H(x; k_1) \equiv 2 + (1 + k_1)\cos 2x + (1 - k_1)\cosh 2x, \quad x > 0. \tag{30}$$

Let $X(k_1) = \frac{1}{2\pi} \cosh^{-1} \left(1 + \frac{4}{k_1 - 1} \right)$, and then $\mathcal{N}(k_1)$ satisfies the estimate

$$2[X(k_1)] + 1 \leq \mathcal{N}(k_1) \leq 2[X(k_1)] + 3. \tag{31}$$

Hence $\mathcal{N}(k_1) \rightarrow \infty$ as $k_1 \rightarrow 1^+$. (By $[X]$ we denote the greatest integer less than or equal to X).

Proof: Let $\mu = \sqrt{\lambda} = x(1 + i)$, $x > 0$. Taking into account the relations

$$\begin{aligned} 2\cosh \mu \cos \mu &= \cosh (1 + i)\mu + \cosh (1 - i)\mu, \\ 2i \sinh \mu \sin \mu &= \cosh (1 + i)\mu - \cosh (1 - i)\mu, \end{aligned}$$

we reduce Eq. (28) to the following form:

$$2 + (1 + k_1) \cos 2x + (1 - k_1) \cosh 2x = 0, \quad x > 0. \quad (32)$$

It can be readily seen that if $0 < k_1 < 1$, then $2 + (1 + k_1) \cos 2x > 0$ and $(1 - k_1) \cosh 2x > 0$, which means that Eq. (32) has no solutions. Statement (1) is shown. Statement (2) follows immediately if one considers Eq. (32) for $k_1 = 1$.

To prove Statement (3), we rewrite Eq. (32) in the form

$$\cosh 2x = \frac{2}{k_1 - 1} + \left(1 + \frac{2}{k_1 - 1}\right) \cos 2x, \quad x > 0, \quad k_1 > 1. \quad (33)$$

The left-hand side of Eq. (33) is monotonically increasing, while the right-hand side is sinusoidal, with maximum $\left(1 + \frac{4}{k_1 - 1}\right)$ and minimum (-1) , and period π . So the graphs of the left- and right-hand side have intersections only on the interval $\left[0, \frac{1}{2} \cosh^{-1} \left(1 + \frac{4}{k_1 - 1}\right)\right]$. There are two intersections for each full period of the

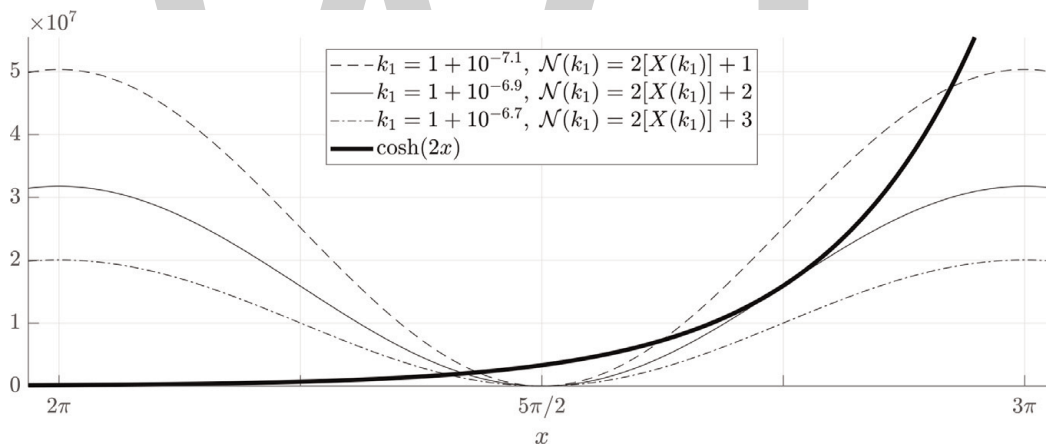


Figure 4. Left- and right-hand side of Eq. (33) for different values of k_1 .

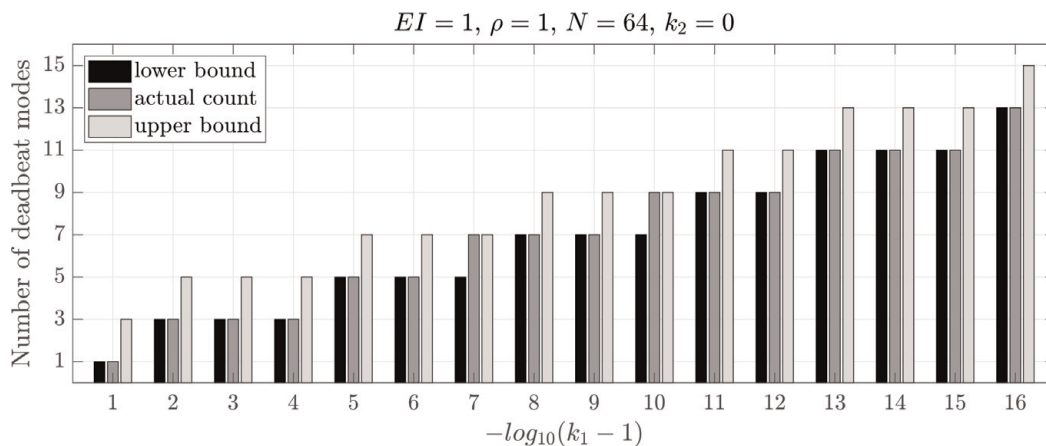


Figure 5. Estimates and actual count of deadbeat modes based on numerical simulations.

right-hand side that fits into the above interval (**Figure 4**). As it can be seen in **Figure 4**, one should add at least one more intersection for the first half-period after the full periods. Depending on the value of k_1 , the two graphs can have two intersections, one tangential intersection or no intersections on the second half-period. This leads to estimate (31). ■

A graphical illustration of the result of Theorem 1 is shown in **Figure 5**.

4. Structure of the deadbeat mode set

The main result on the existence and distribution of double roots of the function $H(x; k_1)$ is presented in the statement below.

Theorem 2: For a given $k_1 > 1$, the multiplicity of each root of $H(x; k_1)$ does not exceed 2. There exists a sequence $\{k_1^{(n)}; n = 0, 1, 2, \dots\}$, such that the function $H(x; k_1)$ has a double root if and only if $k_1 = k_1^{(n)}$ for some n . So the original spectral problem with $k_1 = k_1^{(n)}, k_2 = 0$ has a double deadbeat mode $\lambda_n = \mu_n^2 = 2ix_n^2$. The following asymptotic formulas hold

$$x_n = \frac{3\pi}{4} + \pi n + P_n^{-1} + O(P_n^{-2}), \quad \text{where} \quad P_n = \exp\left\{\frac{3\pi}{2} + 2\pi n\right\}, \quad (34)$$

and

$$k_1^{(n)} = 1 + 4P_n^{-1} + O(P_n^{-2}). \quad (35)$$

Proof: If x is a double root of H , then $H(x; k_1) = H'(x; k_1) = 0$, i.e., separating the real and imaginary parts, we have

$$2 + (k_1 + 1) \cos 2x - (k_1 - 1) \cosh 2x = 0, \quad (36)$$

$$(k_1 + 1) \sin 2x + (k_1 - 1) \sinh 2x = 0. \quad (37)$$

Eliminating k_1 from system given by (36) and (37), we obtain that the following equation has to be satisfied:

$$G(x) \equiv (1 + \cos 2x) \sinh 2x + (1 + \cosh 2x) \sin 2x = 0, \quad x > 0. \quad (38)$$

Rewriting Eq. (37) in the form $(k_1 + 1) \sin 2x = -(k_1 - 1) \sinh 2x$, and taking into account that $k_1 > 1$, we obtain that if x is the solution of Eq. (38), then $\sin 2x < 0$.

Now we show that when $\cos 2x < 0$ and $\sin 2x < 0$, i.e.,

$$\pi(2n + 1) < 2x < \frac{3\pi}{2} + 2\pi n, \quad n \in \{0, 1, 2, \dots\},$$

Eq. (38) does not have any solutions. Indeed, in the above range of x , we have $\cos 2x + \sin 2x = \sqrt{2} \sin(2x + \pi/4) < -1$ and $|\sin 2x - \cos 2x| < 1$. With such estimates we obtain that

$$G(x) = \sin 2x + \sinh 2x + \frac{1}{2} e^{2x} (\cos 2x + \sin 2x) + \frac{1}{2} e^{-2x} (\sin 2x - \cos 2x) < \sin 2x < 0, \quad (39)$$

which mean that Eq. (38) cannot be satisfied.

Now we consider the case when $\cos 2x > 0$ and $\sin 2x < 0$, i.e.,

$$\frac{3\pi}{2} + 2\pi n < 2x < 2\pi(n + 1), \quad n \in \{0, 1, 2, \dots\}.$$

It is convenient to rewrite system given by (36) and (37) in the form $2x = 3\pi/2 + 2\pi n + s$, where $n \in \{0, 1, 2, \dots\}$, $0 < s < \pi/2$. If $g(s) \equiv G(3\pi/4 + \pi n + s)$, then Eq. (38) generates the following equation for g :

$$g(s) \equiv (1 + \sin s) \sinh \left(\frac{3\pi}{2} + 2\pi n + s \right) - \left[1 + \cosh \left(\frac{3\pi}{2} + 2\pi n + s \right) \right] \cos s = 0. \quad (40)$$

Let us show that for each n , Eq. (40) has a unique solution. For $s = 0$ we have

$$g(0) = \sinh \left(\frac{3\pi}{2} + 2\pi n \right) - 1 - \cosh \left(\frac{3\pi}{2} + 2\pi n \right) < 0,$$

and for $s = \pi/2$ we have $g(\pi/2) = 2\sinh(2\pi(n + 1)) > 0$. Evaluating g' we have

$$g'(s) = \sin s + (1 + 2 \sin s) \cosh \left(\frac{3\pi}{2} + 2\pi n + s \right) > 0. \quad (41)$$

Thus $g(s)$ is a monotonically increasing function, such that $g(0) < 0 < g(\pi/2)$, which means that g has a unique root on $[0, \pi/2]$.

Finally we show that the multiplicity of a multiple root cannot exceed 2. Using a contradiction argument, assume that there exists x_0 , such that in addition to Eqs. (36) and (37), one has $H''(x_0; k_1) = 0$, i.e., the multiplicity of x_0 is at least 3. The system $H'(x_0; k_1) = 0$ and $H''(x_0; k_1) = 0$ can be written as

$$\begin{aligned} (k_1 + 1) \sin 2x_0 + (k_1 - 1) \sinh 2x_0 &= 0, \\ (k_1 + 1) \cos 2x_0 + (k_1 - 1) \cosh 2x_0 &= 0. \end{aligned} \quad (42)$$

Since $k_1 > 1$, the second equation of (42) yields $\cos 2x_0 < 0$. Also, since x_0 is a multiple root, we must have $\sin 2x_0 < 0$. Then $2x_0$ is in the third quadrant, which means that $G(x_0) \neq 0$, as we have seen above. This contradicts our assumption that x_0 is a root of Eqs. (36) and (37).

To derive asymptotic distribution of the roots of Eq. (40), we check that with P_n from Eq. (34), the following approximations are valid:

$$\begin{aligned} \sin(2P_n^{-1}) &= 2P_n^{-1} + O(P_n^{-3}), \quad \cos(2P_n^{-1}) = 1 - 2P_n^{-2} + O(P_n^{-4}), \\ 2\sinh\left(\frac{3\pi}{2} + 2\pi n + 2P_n^{-1}\right) &= P_n + 2 + P_n^{-1} + O(P_n^{-2}), \\ 2\cosh\left(\frac{3\pi}{2} + 2\pi n + 2P_n^{-1}\right) &= P_n + 2 + 3P_n^{-1} + O(P_n^{-2}). \end{aligned} \quad (43)$$

Evaluating $g(s)$ from Eq. (40) for $s = 2P_n^{-1}$ and using Eq. (43), we get

$$\begin{aligned} g(2P_n^{-1}) &= [1 + \sin(2P_n^{-1})] \sinh\left(\frac{3\pi}{2} + 2\pi n + 2P_n^{-1}\right) - \\ &\quad \left[1 + \cosh\left(\frac{3\pi}{2} + 2\pi n + 2P_n^{-1}\right) \right] \cos(2P_n^{-1}) = 2P_n^{-1} + O(P_n^{-2}). \end{aligned} \quad (44)$$

Representation (44) implies that there exists n_0 , such that for all $n \geq n_0$, we have $g(2P_n^{-1}) > 0$. Taking into account that $g(0) < 0$, we obtain that the root $s_n, n \geq n_0$, of the function $g(s)$ is located on the interval $(0, 2P_n^{-1})$. To find the location of this root more precisely [18], we use linear interpolation. Namely, substituting Eq. (43) into the expression for $g'(s)$ from Eq. (41) yields

$$g'(2P_n^{-1}) = \frac{P_n}{2} + O(1). \tag{45}$$

Replacing $g(s)$ by the linear function tangential to $g(s)$ at the point $(2P_n^{-1}, g(2P_n^{-1}))$, and finding the root of this function, we get

$$s_n = 2P_n^{-1} - \frac{g(2P_n^{-1})}{g'(2P_n^{-1})} + O(P_n^{-2}) = 2P_n^{-1} + O(P_n^{-2}). \tag{46}$$

Having this approximation for s_n , we immediately get

$$x_n = \frac{3\pi}{4} + \pi n + \frac{s_n}{2} = \frac{3\pi}{4} + \pi n + P_n^{-1} + O(P_n^{-2}). \tag{47}$$

From the equation $H(x_n; k_1^{(n)}) = 0$, we obtain the formula for $k_1^{(n)}$ as

$$k_1^{(n)} = \frac{\sinh(3\pi/2 + 2\pi n + s_n) + \cos s_n}{\sinh(3\pi/2 + 2\pi n + s_n) - \cos s_n}. \tag{48}$$

Substituting formulas (43) and (46) into formula (48), we obtain representation (35). ■

Corollary 1: Let $k_1 = k_1^{(n)}$ for some $n \in \mathbb{N}^+ \cup \{0\}$, and let x_n be the corresponding double root of the function $H(x; k_1^{(n)})$. Then $\lambda_0 = 2ix_n^2$ is an eigenvalue of the operator $\mathcal{L}_{k_1^{(n)}, 0}$, such that the geometric multiplicity of λ_0 is 1 and its algebraic multiplicity is 2. Therefore there exists a unique eigenvector Φ and one associate vector Ψ , such that

$$\mathcal{L}_{k_1^{(n)}, 0} \Phi = \lambda_0 \Phi, \quad \mathcal{L}_{k_1^{(n)}, 0} \Psi - \lambda_0 \Psi = \Phi. \tag{49}$$

Proof: It suffices to show that problem (24) does not have two linearly independent eigenvectors corresponding to λ_0 [18, 19]. Using contradiction argument we assume that for some λ_0 the boundary-value problem (25) with $k_2 = 0$ has two linearly independent solutions ψ and χ . Each function satisfies the problem

$$\varphi''''(x) = \lambda_0^2 \varphi(x), \quad \varphi(0) = \varphi'(0) = 0, \quad \varphi'''(1) = 0, \quad \varphi''(1) = i\lambda_0 k_1 \varphi(1). \tag{50}$$

First we observe that $\psi(1)\chi(1) \neq 0$. Indeed, if $\psi(1) = 0$, then we have

$$\int_0^1 |\psi''(\sigma)|^2 d\sigma = \int_0^1 \psi''''(\sigma) \overline{\psi(\sigma)} d\sigma = \lambda_0^2 \int_0^1 |\psi(\sigma)|^2 d\sigma. \tag{51}$$

Since λ_0 is purely imaginary, Eq. (51) is not valid. We define a new function:

$$g(x) = \psi(x) - \frac{\psi(1)}{\chi(1)}\chi(x). \tag{52}$$

One can readily check that g satisfies the following boundary-value problem:

$$\varphi''''(x) = \lambda_0^2\varphi(x), \quad \varphi(0) = \varphi'(0) = 0, \quad \varphi(1) = \varphi''(1) = 0, \tag{53}$$

and therefore

$$\int_0^1 |g''(\sigma)|^2 d\sigma = \int_0^1 g''''(\sigma)\overline{g(\sigma)}\sigma = \lambda_0^2 \int_0^1 |g(\sigma)|^2 d\sigma. \tag{54}$$

Eq. (54) is valid if and only if $\lambda_0^2 > 0$; however, for a deadbeat mode, $\lambda_0^2 < 0$. The obtained contradiction means that for each double mode, there is one eigenfunction and one associate function. ■

4.1 Deadbeat mode behavior as $k_1 \rightarrow 1$

As k_1 approaches 1, the spectral branches are moving upward and toward the imaginary axis (**Figures 6 and 7**). As a result of this motion, eigenvalues approach the imaginary axis at different rates depending on whether k_1 approaches 1 from above or below.

As follows from **Table 3**, the real parts of the eigenvalues decrease steadily as $k_1 \rightarrow 1^+$, to a point where the eigenvalue becomes a deadbeat mode. An increase in the number of deadbeat modes can be seen as $k_1 \rightarrow 1^+$, which is in agreement with Statement (3) of Theorem 1. One can see from **Table 3** that there are pairs of modes such that the distance between them tends to zero as $k_1 \rightarrow 1^+$. (Compare modes no.5 and no.7 for $|k_1 - 1| = 10^{-4}$, modes no.4 and no.7 for $|k_1 - 1| = 10^{-6}$, modes no.5 and no.8 for $|k_1 - 1| = 10^{-8}$, and modes no.4 and no.7 for $|k_1 - 1| = 10^{-10}$). Such behavior indicates convergence of the two simple deadbeat modes to a double mode, which is consistent with Theorem 2.

Analyzing **Table 4**, one can see that the eigenvalues get closer to the imaginary axis as $k_1 \rightarrow 1^-$. However the rate at which their real parts approach zero is

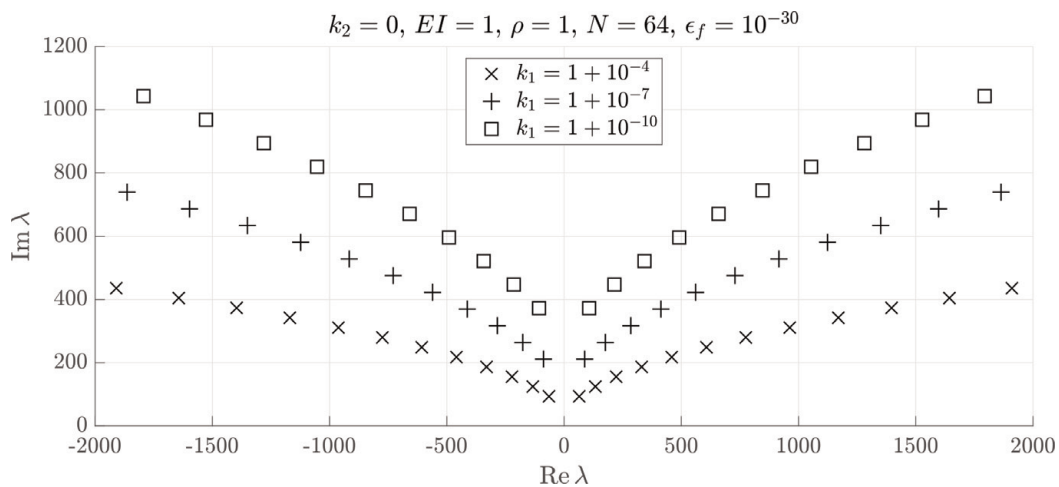


Figure 6.
Eigenvalues with $|\text{Re}\lambda| > 10$ as $k_1 \rightarrow 1^+$.

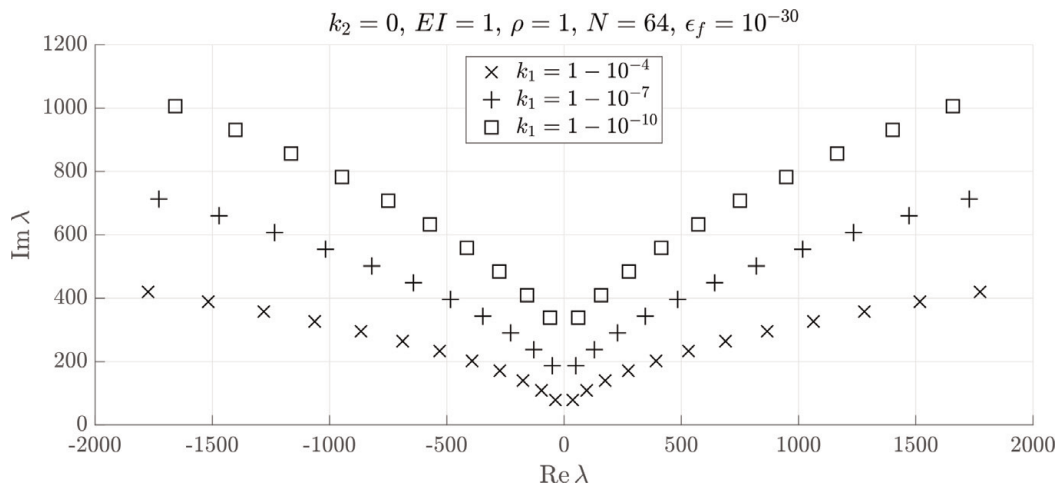


Figure 7.
Eigenvalues with $|\operatorname{Re}\lambda| > 10$ as $k_1 \rightarrow 1^-$.

significantly lower than in the case $k_1 \rightarrow 1^+$. Even at $k_1 = 1 - 10^{-10}$, the eigenvalue closest to the imaginary axis has a real part of about 10^{-4} , which means that it is not a deadbeat mode (see Statement (1) of Theorem 1).

The eigenvalues near the imaginary axis approach the same double deadbeat modes in both cases when $k_1 \rightarrow 1^\pm$ (see Statement (2) of Theorem 1). In conclusion, one can claim that the eigenvalues are indeed approaching the imaginary axis; however, the rate of this approach is different for $k_1 \rightarrow 1^-$ and $k_1 \rightarrow 1^+$. In the former case, an eigenvalue's distance from the imaginary axis decreases very slowly; in the latter case, the eigenvalues quickly “jump” on the imaginary axis and turn into deadbeat modes.

5. Outline of the numerical scheme

To carry out the numerical analysis of the differential operator \mathcal{L}_{k_1, k_2} , we use the Chebyshev collocation method and cardinal functions [14–16].

Recall that the N th Chebyshev polynomial of the first kind is defined by

$$T_N(\xi) = \cos N\theta, \quad -1 \leq \xi \leq 1 \quad \text{where} \quad \xi = \cos \theta, \quad \theta \in [0, \pi]. \quad (55)$$

The cardinal functions, $\psi_k(\xi)$, and the Chebyshev-Gauss-Lobatto (CGL) grid points $\{\xi_k\}$ are defined as follows:

$$\psi_k(\xi) = (-1)^k \frac{(1 - \xi^2) T'_{N-1}(\xi)}{c_k (N-1)^2 (\xi - \xi_k)}, \quad \xi_k = \cos \frac{(k-1)\pi}{N-1}, \quad \text{for } 1 \leq k \leq N, \quad (56)$$

where coefficients c_k are such that $c_1 = c_N = 2$ and $c_k = 1$ for $1 < k < N$. The main property of cardinal functions is $\psi_k(\xi_j) = \delta_{kj}$ (using the Kronecker delta). The family $\{\psi_k\}_{k=1}^N$ forms a basis in the space of polynomials of degree $(N-1)$, i.e., if f is such polynomial, then f and f' can be written in the forms

$$f(\xi) = \sum_{k=1}^N f(\xi_k) \psi_k(\xi) \quad \text{and} \quad f'(\xi) = \sum_{k=1}^N f'(\xi_k) \psi_k(\xi). \quad (57)$$

If $\mathbf{f} = [f(\xi_1), f(\xi_2), \dots, f(\xi_N)]^T$ and $\mathbf{g} = [f'(\xi_1), f'(\xi_2), \dots, f'(\xi_N)]^T$, then $\mathbf{g} = D\mathbf{f}$, where D is the Chebyshev derivative matrix with the elements

$$D_{11} = -D_{NN} = \frac{1 + 2(N - 1)^2}{6}, \quad D_{kk} = -\frac{\xi_k}{2(1 - \xi_k^2)} \quad \text{for } 1 < k < N, \tag{58}$$

$$D_{j,k} = \frac{c_j(-1)^{j+k}}{c_k(\xi_j - \xi_k)} \quad \text{for } j \neq k.$$

5.1 Discretization of \mathcal{L}_{k_1, k_2}

Rescaling the independent variable x as $\xi = 2x - 1$, we rewrite the operator and its domain, representations (14) and (15), in the form

$$\mathcal{L}_{k_1, k_2} = -i \begin{bmatrix} 0 & 1 \\ -\frac{16}{\rho(\xi)} \frac{\partial^2}{\partial \xi^2} \left(EI(\xi) \frac{\partial^2}{\partial \xi^2} \right) & 0 \end{bmatrix}, \tag{59}$$

and

$$\mathcal{D}(\mathcal{L}_{k_1, k_2}) = \{(u_0, u_1)^T \in \mathcal{H} : u_0 \in H^4(-1, 1), u_1 \in H_0^2(-1, 1); u_1(-1) = u_1'(1) = 0; 4EI(1)u_0''(1) = -k_1u_1(1), 4(EI(\xi)u_0''(\xi))'|_{\xi=1} = k_2u_1'(1)\}, \tag{60}$$

where $\mathcal{H} = H_0^2(-1, 1) \times L^2(-1, 1)$, equipped with the norm

$$\|U\|_{\mathcal{H}}^2 = \frac{1}{4} \int_{-1}^1 [16EI(\xi)|u_0''(\xi)|^2 + \rho(\xi)|u_1(\xi)|^2] d\xi. \tag{61}$$

We approximate the action of \mathcal{L}_{k_1, k_2} on the finite-dimensional subspace $\mathcal{H}_N \subset \mathcal{H}$ of polynomials of degree at most $(N - 1)$. Using the CGL grid and the cardinal functions, we substitute for u_0 and u_1 their truncated expansions:

$$u_0(\xi) \approx \sum_{k=1}^N \Phi_k \psi_k(\xi), \quad \Phi_k = u_0(\xi_k), \quad u_1(\xi) \approx \sum_{k=1}^N \Theta_k \psi_k(\xi), \quad \Theta_k = u_1(\xi_k). \tag{62}$$

Let Φ and Θ be N -dim vectors and Ψ be a $2N$ -dim vector defined by

$$\Phi = [\Phi_1, \Phi_2, \dots, \Phi_N]^T, \quad \Theta = [\Theta_1, \Theta_2, \dots, \Theta_N]^T, \quad \Psi = \begin{bmatrix} \Phi \\ \Theta \end{bmatrix}. \tag{63}$$

Let L be the finite-dimensional approximation of the differential operator \mathcal{L}_{k_1, k_2} . The discretized operator L induced by \mathcal{L}_{k_1, k_2} can be given by

$$L = -i \begin{bmatrix} \mathbf{0} & I_{N \times N} \\ -16 \frac{EI}{\rho} D^4 & \mathbf{0} \end{bmatrix}, \tag{64}$$

where $I_{N \times N}$ is the $N \times N$ identity matrix and D is the derivative matrix (58).

5.2 Incorporating the boundary conditions

Discretization of the boundary conditions in the domain description (60) yields

$$\begin{aligned} \Phi_N = 0, \quad [D\Phi]_N = 0, \quad \Theta_N = 0, \quad [D\Theta]_N = 0, \\ 4EI[D^2\Phi]_1 + k_1\Theta_1 = 0, \quad 4EI[D^3\Phi]_1 - k_2[D\Theta]_1 = 0. \end{aligned} \quad (65)$$

Let $r_N, z_N, l_N \in \mathbb{R}^N$ be auxiliary row-vectors

$$r_N = [0 \ 0 \ \dots \ 0 \ 1], \quad z_N = [0 \ 0 \ \dots \ 0 \ 0], \quad l_N = [1 \ 0 \ \dots \ 0 \ 0] \quad (66)$$

and D_j^n designate the j th row of the n th derivative matrix D^n . Using Eqs. (66) we represent Eqs. (65) as the following matrix equation:

$$\mathbb{K}\Psi \equiv \begin{bmatrix} r_N & z_N \\ D_N^1 & z_N \\ z_N & r_N \\ z_N & D_N^1 \\ 4EID_1^2 & k_1l_N \\ 4EID_1^3 & -k_2D_1^1 \end{bmatrix} \begin{bmatrix} \Phi \\ \Theta \end{bmatrix} = 0. \quad (67)$$

\mathbb{K} is called the boundary operator. Let \mathcal{K}_N be the kernel of \mathbb{K} , i.e., $\mathcal{K}_N = \{v \in \mathbb{R}^{2N} : \mathbb{K}v = 0\}$. We have to identify all eigenvalues of the operator L , when its domain is restricted to \mathcal{K}_N . It is clear that \mathcal{K}_N is isomorphic to \mathbb{R}^k with $k \equiv \dim \mathcal{K}_N = \dim \mathcal{H}_N - \text{rank} \mathbb{K} = 2N - 6$. Let B be the matrix consisting of column vectors that form an orthonormal basis in \mathcal{K}_N . It is clear that $B^T B$ is the identity matrix on \mathbb{R}^k and BB^T is the identity matrix on \mathbb{K} . The following result holds: if λ is an eigenvalue of the operator L , and the corresponding eigenvector Ψ satisfies Eq. (67), then the same λ is an eigenvalue of the matrix $(B^T L B)$. However, the inverse statement is not necessarily true. Indeed, we observe that BB^T is the identity in \mathcal{K}_N , which is not equivalent to the identity in \mathcal{H}_N . Assume now that λ is an eigenvalue of $B^T L B$ with corresponding eigenvector $v \in \mathbb{R}^k$. If $\Psi = Bv$, we have

$$BB^T L \Psi = BB^T L B v = \lambda B v = \lambda \Psi, \quad (68)$$

but $BB^T L \neq L$, which indicates that fake eigenvalues may exist.

5.3 Filtering of spurious eigenvalues

In order to decide which eigenvalues of $B^T L B$ should be discarded, we impose the following condition. Let Λ be the spectrum of $B^T L B$ and V be the set of its eigenfunctions. We construct the set of “trusted” eigenvalues [14, 15], for some $\varepsilon_f > 0$ filtering precision, as

$$\Lambda_\varepsilon = \{\lambda \in \Lambda : \|LBv_\lambda - \lambda Bv_\lambda\|_C < \varepsilon_f, \text{ for corresponding eigenvector } v_\lambda \in V\}, \quad (69)$$

where $\|\cdot\|_C$ is a discrete approximation to the integral norm defined in Eq. (61). (The subscript C is short for Chebyshev). Using the CGL quadrature, we obtain the following formula for the norm of a vector Ψ defined as in Eq. (63):

$$\|\Psi\|_C = \frac{\pi/4}{N-1} \sum_{k=1}^N \sqrt{1 - \frac{\xi_k^2}{\xi_k^2}} \left[16EI(\xi_k) | [D^2\Phi]_k |^2 + \rho(\xi_k) |\Theta_k|^2 \right].$$

6. Conclusions

In this work we have considered the spectral properties of the Euler-Bernoulli beam model with special feedback-type boundary conditions. The dynamics generator of the model is a non-self-adjoint matrix differential operator acting in a Hilbert space of two-component Cauchy data. This operator has been approximated by a “discrete” operator using Chebyshev polynomial approximation. We have shown that the eigenvalues of the main operator can be approximated by the eigenvalues of its discrete counterpart with high accuracy. This means that the leading asymptotic terms in formulas (20) and (21) can be used by practitioners who need the elastic modes.

Further results deal with existence and formulas of the deadbeat modes. It has been shown that for the case when one control parameter, k_1 , is such that $k_1 \rightarrow 1^+$ and the other one $k_2 = 0$, the number of deadbeat modes approaches infinity. The formula for the rate at which the number of the deadbeat modes tends to infinity has been derived. It has also been established that there exists a sequence $\{k_1^{(n)}\}_{n=1}^{\infty}$ of the values of parameter k_1 , such that the corresponding deadbeat mode has a multiplicity 2, which yields the existence of the associate mode shapes for the operator \mathcal{L}_{k_1, k_2} . The formulas for the double deadbeat modes and asymptotics for the sequence $\{k_1^{(n)}\}$ as $n \rightarrow \infty$ have been derived.

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