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An objective isogeometric mixed finite element formulation for nonlinear elastodynamic beams with incompatible warping strains

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Abstract

We present a stable mixed isogeometric finite element formulation for geometrically and materially nonlinear beams in transient elastodynamics, where a Cosserat beam formulation with extensible directors is used. The extensible directors yield a linear configuration space incorporating constant in-plane cross-sectional strains. Higher-order (incompatible) strains are introduced to correct stiffness, whose additional degrees of freedom are eliminated by an element-wise condensation. Further, the present discretization of the initial director field leads to the objectivity of approximated strain measures, regardless of the degree of basis functions. For physical stress resultants and strains, we employ a global patch-wise approximation using B-spline basis functions, whose higher-order continuity enables using much fewer degrees of freedom than an element-wise approximation. For time-stepping, we employ implicit energy-momentum consistent scheme, which exhibits superior numerical stability in comparison to standard trapezoidal and mid-point rules. Several numerical examples are presented to verify the present method.

Keywords Beam · Extensible directors · Objectivity · Mixed formulation · Warping · Implicit dynamics

1 Introduction

A Cosserat beam represents a slender object described by a spatial curve with two attached directors at each material point. Those directors span the (planar) cross-section and are often

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assumed to be orthonormal so that they can be parameterized by three rotational parameters. It has been shown that a standard finite element approximation of three-dimensional rotation parameters can lead to a loss of frame-invariance of the underlying continuous strains due to the nonadditive nature of finite rotations, see Crisfield and Jelenić [11] and Jelenić and Crisfield [20] for the relevant investigation and its remedies using the concept of local rotation. For a comprehensive review of different types of rotational parameterizations for nonlinear beams, one may refer to Romero [33]. It has also been shown that a direct finite element approximation of the director vectors can circumvent this issue, which, however, requires an additional constraint for the directors' orthonormality. For example, this constraint can be imposed at the nodal points using a Lagrange multiplier method (Betsch and Steinmann [2]) or introducing nodal rotational parameters (Gruttmann et al. [18], Romero and Armero [34]). On the other hand, this orthonormality constraint can be simply abandoned so that one has two independent extensible (unconstrained) directors, which represent inplane shear and transverse normal strains of the cross-section. This increases the number of degrees of freedom (DOFs) per cross-section from six to nine, but the configuration space becomes linear, \mathbb{R}^9 . The relevant approaches can be found, e.g., in Rhim and Lee [32], Coda [10], Durville [13], and Choi et al. [7]. Note that these approaches based on a direct finite element approximation of the director field may have a singularity in coarse meshes, when the directors' orientation changes abruptly along the length (Romero [33, Sect. 4.3]). The direct finite element approximation of (total) directors implies that the initial director field is also approximated. A continuous (exact) representation of the initial director field, inconsistent with the finite element approximation, may lead to a loss of objectivity. In Choi et al. [9, Sect. 6.2], it turned out that using a reduced degree of basis p_d for the change of directors lower than that of the initial (continuous) director field may result in the inability to represent rigid body rotations. In the present paper, we present a finite element approximation of the initial director field such that the objectivity is satisfied for any degree p_d . For the relevant discussion on the objectivity under Kirchhoff beam kinematics, we refer interested readers to Meier et al. [26].

Out-of-plane warping means a cross-sectional deformation such that an initially planar cross-section does not remain plane. It can be coupled with other deformation modes like bending, shear, or torsion. In order to take this into account in Timohsenko beam kinematics, Simo and Vu-Quoc [38] introduced an additional DOF representing the amplitude of warping, and it is multiplied by a pre-defined warping function to produce an out-of-plane displacement, which has also been applied in the framework of unconstrained directors in Coda [10]. This has been extended by Klinkel and Govindjee [23] to deal with anisotropic linear elastic materials exhibiting a bending-torsion coupling. In these previous works, the additional DOF for warping leads to nonstandard stress resultants, bi-moment, and bi-shear in the equilibrium equations. See also Nukala and White [29] for a relevant approach with simplified shear strains such that the transverse shear strain and the warping shear strain due to nonuniform torsion are neglected. In the framework of an enhanced assumed strain (EAS) method (Simo and Rifai [36]), one can enhance the cross-sectional strains that do not exist in the compatible strain field from the configuration. This requires additional constraints like orthogonality between the enhanced strain and the stress field. Wackerfuß and Gruttmann [41] presented a construction of global polynomial basis functions of arbitrary order for the enhanced cross-sectional strains. This has been extended to incorporate arbitrary (open) cross-section shapes by using a local polynomial basis in Wackerfuß and Gruttmann [42], which has also been employed in Klarmann et al. [22] for a coupling between beam and brick elements. In the present work, the construction of global polynomial basis functions needs to be more generalized in order to account for the additional deformation modes in the



compatible strain field, such as transverse normal and in-plane shear strains, due to the extensible (unconstrained) directors. An alternative way to incorporate warping, which, however, requires an increased number of global DOFs, is to use higher-order directors (Choi et al. [8], Moustacas et al. [28]). One may also consider using multiple brick (solid beam) elements in the cross-section, e.g., as in Frischkorn and Reese [15].

The total linear momentum, total angular momentum, and total energy are fundamental integral quantities of motion in nonlinear elastodynamics, and their conservation (under suitable boundary conditions) manifests the stability of a numerical time-stepping scheme. For nonlinear problems, the standard trapezoidal rule, corresponding to $\beta=1/4$ and $\gamma=1/2$ in the Newmark family, preserves the total linear momentum but neither the total angular momentum nor the total energy (Simo et al. [39]). The standard mid-point rule preserves the total linear and angular momentums but not the total energy (Gonzalez [16]). It has been shown that time-stepping schemes that preserve the conservation laws underlying the time-continuous form show superior numerical stability compared with standard ones like the trapezoidal and mid-point rules. A class of these energy-momentum-consistent (EMC) algorithms can be obtained by modifying the (implicit) mid-point rule. For example, one may refer to the algorithmic treatment of the stress resultant in Romero and Armero [34], and Betsch and Janz [1] for geometrically nonlinear beam and shell problems, respectively, and Gonzalez [16] for general (nonlinear) hyperelastic materials.

The paper is organized as follows. In Sect. 2, we introduce the beam kinematics based on the extensible directors. Section 3 presents a mixed variational formulation for nonlinear dynamics of beams, which includes an EAS method for enriching higher-order cross-sectional strains. Section 4 presents an implicit time-stepping scheme for energy-momentum consistency. Section 5 presents an isogeometric spatial discretization. In Sect. 6, several numerical examples are presented. Section 7 concludes the paper.

2 Beam kinematics

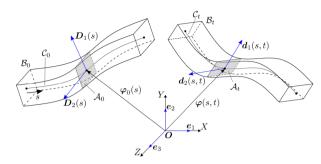
This section presents the kinematical description of Cosserat beams, based on extensible directors, and the resulting configuration–strain relation. A beam characterizes a three-dimensional body whose dimension along the longitudinal direction is much larger than the others such that a suitable kinematical assumption on the cross-sectional deformation can be employed. The origin of the transverse coordinates ζ^{α} ($\alpha \in \{1, 2\}$) is given by the geometrical center of the initial cross-section, which coincides with the mass center, assuming constant mass density in the initial configuration. The line connecting these center points is called an *initial center axis* whose position is represented by the vector $\varphi_0(s)$, $s \in [0, L]$, where s denotes its arc-length coordinate, and s denotes its length. Then, the initial beam configuration can be expressed by

$$\boldsymbol{x}_0(\zeta^1, \zeta^2, s) = \boldsymbol{\varphi}_0(s) + \zeta^{\alpha} \boldsymbol{D}_{\alpha}(s), \tag{1}$$

where the two orthonormal vectors $D_{\alpha}(s) \in \mathbb{R}^3$ ($\alpha \in \{1, 2\}$) are called *initial directors*, and they span the initial cross-sectional plane, \mathcal{A}_0 in Fig. 1. Those initial directors are defined along two principal directions of the cross-section's second moment of inertia. Here and hereafter, unless stated otherwise, repeated Greek indices like α , β , and γ imply summation over 1 to 2, and repeated Latin indices like i and j imply summation over 1 to 3. The initial directors are unit vectors and dimensionless, such that the cross-sectional dimension is associated with the range of the *transverse coordinates* $(\zeta^1, \zeta^2) \in \mathcal{A} \cup \partial \mathcal{A}$, where \mathcal{A}



Fig. 1 A schematic illustration of the beam kinematics. \mathcal{B}_0 and \mathcal{B}_t denote the (open) domains of the initial (undeformed) and current configurations, respectively. e_1 , e_2 , and e_3 represent the global Cartesian base vectors. This figure is redrawn with modifications from Choi et al. [9]



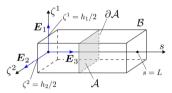


Fig. 2 A schematic illustration of the reference domain of a beam having a rectangular cross-section with dimension $h_1 \times h_2$. \mathcal{B} denotes the (open) domain of the reference configuration. This figure is redrawn with modifications from Choi et al. [9]

denotes the (open) domain of the (initial) cross-section and ∂A denotes its boundary, see Fig. 2. It should be noted that the transverse coordinates ζ^{α} are independent of time. In the initial configuration, we define the covariant basis $G_i := \partial x_0/\partial \zeta^i$ ($i \in \{1, 2, 3\}$) with $\zeta^3 \equiv s$, and the contravariant basis $\{G^1, G^2, G^3\}$ can then be uniquely determined by the orthogonality condition $G_i \cdot G^j = \delta_i^j$, where δ_i^j denotes the Kronecker-delta.

Remark 2.1 Initial curvature. The initial covariant base vectors are obtained by (Choi et al. [7])

$$\begin{cases} G_1 = D_1(s), \\ G_2 = D_2(s), \\ G_3 = \varphi_{0,s}(s) + \zeta^{\alpha} D_{\alpha,s}(s). \end{cases}$$
 (2)

The infinitesimal volume in the initial configuration can be decomposed into

$$d\mathcal{B}_0 = j_0 \, d\mathcal{A} \, ds, \tag{3}$$

where the Jacobian is expressed by

$$j_0 := (\mathbf{G}_1 \times \mathbf{G}_2) \cdot \mathbf{G}_3 = 1 - \zeta^{\alpha} \kappa_{\alpha}, \tag{4}$$

with

$$\kappa_{\alpha} := \boldsymbol{\varphi}_{0,ss} \cdot \boldsymbol{D}_{\alpha}, \ \alpha \in \{1,2\}, \tag{5}$$

which is associated with the initial curvature of the center axis, for a given initial center axis curve $\varphi_0 \equiv \varphi_0(s)$ and the attached initial directors $D_\alpha \equiv D_\alpha(s)$. Note that, for initially straight beams or linear elements, $\kappa_\alpha = 0$ so that $j_0 = 1$.



We introduce the first-order beam kinematics, which expresses the current position of a material point, as

$$\mathbf{x}(\zeta^1, \zeta^2, s, t) = \boldsymbol{\varphi}(s, t) + \zeta^{\alpha} \boldsymbol{d}_{\alpha}(s, t), \tag{6}$$

where $\varphi(s,t)$ denotes the current position of the center axis, and the current cross-sectional plane is spanned by two *current directors* $d_{\alpha}(s,t) \in \mathbb{R}^3$ ($\alpha \in \{1,2\}$), see Fig. 1. Here, $t \in [0,T]$ denotes time, where T represents the terminal time. We obtain the velocity by differentiating Eq. (6) with respect to time, as

$$\dot{\boldsymbol{x}}(\zeta^1, \zeta^2, s, t) = \dot{\boldsymbol{\varphi}}(s, t) + \zeta^\alpha \dot{\boldsymbol{d}}_\alpha(s, t), \tag{7}$$

where the upper dot denotes the material time derivative. Hereafter, for brevity, we often omit the arguments, e.g., $\mathbf{x} \equiv \mathbf{x}(\zeta^1, \zeta^2, s, t)$, $\boldsymbol{\varphi} \equiv \boldsymbol{\varphi}(s, t)$ and $\boldsymbol{d}_{\alpha} \equiv \boldsymbol{d}_{\alpha}(s, t)$. Further, we define a configuration variable $\mathbf{y} := \left[\boldsymbol{\varphi}^{\mathrm{T}}, \boldsymbol{d}_{1}^{\mathrm{T}}, \boldsymbol{d}_{2}^{\mathrm{T}}\right]^{\mathrm{T}} \in \mathbb{R}^{d}$, where d = 9 denotes the number of independent cross-sectional components.

2.1 Geometric strain: strain-configuration relation

We may represent a three-dimensional strain state of a beam by so called *beam strains* evaluated at the center axis. In the current configuration, we define the covariant basis $\mathbf{g}_i := \partial \mathbf{x}/\partial \zeta^i$ ($i \in \{1, 2, 3\}$), and then the contravariant basis $\{\mathbf{g}^1, \mathbf{g}^2, \mathbf{g}^3\}$ can be uniquely determined by the orthogonality condition $\mathbf{g}_i \cdot \mathbf{g}^j = \delta_i^j$. Then, the deformation gradient tensor \mathbf{F} can be expressed by (Wriggers [43, page 31])

$$F := \partial x / \partial x_0 = g_i \otimes G^i. \tag{8}$$

The Green-Lagrange strain tensor $E \equiv E(y)$ is defined by

$$E(\mathbf{y}) := \frac{1}{2} \left(\mathbf{F}^{\mathrm{T}} \mathbf{F} - \mathbf{1} \right) = E_{ij} \, \mathbf{G}^{i} \otimes \mathbf{G}^{j}, \tag{9}$$

where 1 denotes the identity tensor. In Voigt notation, we have

$$\underline{\boldsymbol{E}} := [E_{11}, E_{22}, E_{33}, 2E_{12}, 2E_{13}, 2E_{23}]^{\mathrm{T}} = \boldsymbol{A}(\zeta^{1}, \zeta^{2}) \boldsymbol{\varepsilon}(\boldsymbol{y}), \tag{10}$$

where the underline (\bullet) denotes Voigt notation of a second-order tensor, and we have defined the matrix $A \equiv A(\zeta^{\frac{1}{2}}, \zeta^{2})$ (Choi et al. [7])

and the beam strain components

$$\varepsilon := \frac{1}{2} (\|\boldsymbol{\varphi}_{,s}\|^2 - 1), \tag{12a}$$



$$\rho_{\alpha} := \boldsymbol{\varphi}_{s} \cdot \boldsymbol{d}_{\alpha,s} - \boldsymbol{\varphi}_{0,s} \cdot \boldsymbol{D}_{\alpha,s}, \tag{12b}$$

$$\delta_{\alpha} := \boldsymbol{\varphi}_{s} \cdot \boldsymbol{d}_{\alpha} - \boldsymbol{\varphi}_{0s} \cdot \boldsymbol{D}_{\alpha}, \tag{12c}$$

$$\gamma_{\alpha\beta} := \boldsymbol{d}_{\alpha} \cdot \boldsymbol{d}_{\beta,s} - \boldsymbol{D}_{\alpha} \cdot \boldsymbol{D}_{\beta,s}, \tag{12d}$$

$$\chi_{\alpha\beta} := \frac{1}{2} (\boldsymbol{d}_{\alpha} \cdot \boldsymbol{d}_{\beta} - \boldsymbol{D}_{\alpha} \cdot \boldsymbol{D}_{\beta}), \tag{12e}$$

$$\kappa_{\alpha\beta} := \frac{1}{2} \left(\boldsymbol{d}_{\alpha,s} \cdot \boldsymbol{d}_{\beta,s} - \boldsymbol{D}_{\alpha,s} \cdot \boldsymbol{D}_{\beta,s} \right), \tag{12f}$$

 $\alpha, \beta \in \{1, 2\}$, which are arrayed in the form

$$\boldsymbol{\varepsilon}(\mathbf{y}) := [\varepsilon, \rho_1, \rho_2, \kappa_{11}, \kappa_{22}, 2\kappa_{12}, \delta_1, \delta_2, \gamma_{11}, \gamma_{12}, \gamma_{21}, \gamma_{22}, \chi_{11}, \chi_{22}, 2\chi_{12}]^{\mathrm{T}} \in \mathbb{R}^{d_{\mathrm{p}}},$$
 (13)

where $d_{\rm p}=15$ denotes the number of independent strain components. Those beam strain components are called

 ε : axial strain,

 ρ_1, ρ_2 : bending strain,

 $\kappa_{11}, \kappa_{22}, \kappa_{12}$: higher-order bending strain,

 δ_1, δ_2 : transeverse shear strain,

 γ_{11} , γ_{12} , γ_{21} , γ_{22} : couple shear strain,

 $\chi_{11}, \chi_{22}, \chi_{12}$: in-plane stretching and shear strains.

Note that $\kappa_{12} = \kappa_{21}$ and $\chi_{12} = \chi_{21}$. We call $\varepsilon(y)$, expressed in terms of the configuration variable y, a *geometric strain*, which represents the strain-configuration relation, whose objectivity is proved in Sect. A.1.

3 Mixed variational formulation

This section presents a mixed variational formulation for nonlinear transient dynamics of beams based on the kinematics in Sect. 2.

3.1 Variational forms

3.1.1 Total strain energy: Hu-Washizu principle

Based on the Hu-Washizu variational principle, the total strain energy, $U_{\rm HW} \equiv U_{\rm HW}(y, E_{\rm p}, S_{\rm p})$, can be expressed, using Eq. (3), as

$$U_{\text{HW}} := \int_0^L \int_{\mathcal{A}} \left[\Psi \left(\boldsymbol{E}_{\text{p}} \right) + \left\{ \boldsymbol{E}(\boldsymbol{y}) - \boldsymbol{E}_{\text{p}} \right\} : \boldsymbol{S}_{\text{p}} \right] j_0 \, d\mathcal{A} \, ds, \tag{14}$$

where $\Psi(E_p)$ denotes the strain energy density per unit undeformed volume, expressed in terms of the *physical* (independent) Green-Lagrange strain tensor, E_p . The compatibility condition,

$$E(y) = E_{\rm p},\tag{15}$$



is enforced by a Lagrange multiplier method, where the Lagrange multiplier S_p can be interpreted as the *physical* second Piola-Kirchhoff stress tensor. This type of variational formulation is motivated by the following:

- The introduction of an independent strain E_p aims at an alleviation of numerical locking due to the parasitic terms in the compatible (geometric) strain $E(y^h)$ from the finite element approximation of the configuration variable, $y \approx y^h$.
- The introduction of an independent stress field S_p alleviates the overestimation of internal force at inequilibrium configurations, which turns out to allow much larger load increments compared with the displacement-based formulation.

The relevant investigations can be found in, e.g., Klinkel et al. [24] for elastostatics, and Betsch and Janz [1] for elastodynamics. Further, we may enhance the physical strain field, as (Büchter et al. [3], Simo and Rifai [36])

$$\boldsymbol{E}^{\text{tot}} = \underbrace{\boldsymbol{E}_{\text{p}}}_{\text{kinematic}} + \underbrace{\boldsymbol{\tilde{E}}}_{\text{enhanced}}, \tag{16}$$

where 'tot' means 'total', and the 'kinematic' part means its compatibility with the geometric (compatible) strain, E(y) in Eq. (15). Substituting Eq. (16) into Eq. (14), the total strain energy in Eq. (14) can be rewritten as

$$U_{\text{HW}} = \int_0^L \int_{\mathcal{A}} \left[\Psi \left(\mathbf{E}^{\text{tot}} \right) + \left\{ \mathbf{E}(\mathbf{y}) - \mathbf{E}_{\text{p}} \right\} : \mathbf{S}_{\text{p}} \right] j_0 \, d\mathcal{A} \, ds, \tag{17}$$

where the following orthogonality condition is utilized

$$\int_{\mathcal{A}} \widetilde{\mathbf{E}} : \mathbf{S}_{\mathsf{p}} \, j_0 \, \mathrm{d}\mathcal{A} = 0. \tag{18}$$

The compatibility in Eq. (15) enables us to represent the three-dimensional strain state in terms of the *physical strain* $\boldsymbol{\varepsilon}_p \equiv \boldsymbol{\varepsilon}_p(s,t) \in \mathbb{R}^{d_p}$, as

$$\boldsymbol{E}_{\mathrm{p}} = \boldsymbol{A}(\zeta^{1}, \zeta^{2}) \,\boldsymbol{\varepsilon}_{\mathrm{p}}(s, t), \tag{19}$$

as in Eq. (10). This, so called an enhanced assumed strain (EAS) method is introduced to correct stiffness, which, especially, takes the following into account:

- Correction of the bending stiffness, considering the fact that a bending deformation may lead to (at least) linear in-plane cross-sectional strain due to nonzero Poisson's ratio.
- Correction of the torsional stiffness, considering the fact that torsion may induce crosssectional warping.
- Correction of the transverse shear stiffness.

The enhanced strain can also be represented by the parameters $\boldsymbol{\alpha} \in \mathbb{R}^{d_{\mathrm{a}}}$, as

$$\underline{\underline{\tilde{E}}} = \Gamma(\zeta^1, \zeta^2) \alpha(s, t), \tag{20}$$



with the following matrix of basis functions (Wackerfuß and Gruttmann [41])

$$\Gamma(\zeta^{1}, \zeta^{2}) = \begin{bmatrix} \mathbf{w}_{1} & & & & & \\ & \mathbf{w}_{1} & & & & \\ & & \mathbf{w}_{3} & & & \\ & & & \mathbf{w}_{2} & & \\ & & & & \mathbf{w}_{4,1} \\ & & & & \mathbf{w}_{4,2} \end{bmatrix}_{6 \times d_{2}},$$
(21)

where the blank entries represent zeros, and $(\bullet)_{,\alpha}$ denotes the derivative with respect to ζ^{α} ($\alpha \in \{1, 2\}$). Here, \mathbf{w}_1 , \mathbf{w}_2 , \mathbf{w}_3 , and \mathbf{w}_4 denote row arrays of the *global* polynomial basis functions for enriching the transverse normal $(\widetilde{E}_{11}, \widetilde{E}_{22})$, in-plane shear (\widetilde{E}_{12}) , axial normal (\widetilde{E}_{33}) , and transverse shear $(\widetilde{E}_{13}, \widetilde{E}_{23})$ strains, respectively.

Remark 3.1 For the transverse shear strains $(\widetilde{E}_{13}, \widetilde{E}_{23})$, we employ the derivatives of \mathbf{w}_4 . However, we distinguish \mathbf{w}_4 from \mathbf{w}_3 , which is in contrast to the formulation of Wackerfuß and Gruttmann [41]. This is due to the quadratic basis for E_{33} in Eq. (11), which does not exist for E_{13} and E_{23} . We verify the erroneous results from using $\mathbf{w}_4 = \mathbf{w}_3$ in Figs. 7 and 13a in Sect. 6.3.

The detailed process of constructing $\mathbf{w}_i \in \mathbb{R}^{1 \times d_i}$ $(i \in \{1, 2, 3, 4\})$ is given in Appendix A.3. Then, the total number of independent components in the enhanced strain field is $d_a = 2d_1 + d_2 + d_3 + d_4$, where the dimension d_i is given by Eq. (A.20). It is noted that 'global' implies that the basis functions span the whole domain of the cross-section, $A \ni (\zeta^1, \zeta^2)$. This, however, is not suitable to represent localized warping, e.g., for a beam having an open cross-section. This limits the applicability of the present formulation to convex-shaped cross-sections like a rectangular one. In order to overcome this, one may consider the local polynomial basis functions in Wackerfuß and Gruttmann [42]. Substituting Eqs. (19) and (20) into Eq. (17) gives

$$U_{\text{HW}} = \int_{0}^{L} \left[\psi \left(\boldsymbol{\varepsilon}_{\text{p}}, \boldsymbol{\alpha} \right) + \underbrace{\left\{ \boldsymbol{\varepsilon}(\boldsymbol{y}) - \boldsymbol{\varepsilon}_{\text{p}} \right\} \cdot \boldsymbol{r}_{\text{p}} \right] ds}_{\text{compatibility}}, \tag{22}$$

where we define the line energy density (strain energy per undeformed unit length),

$$\psi(\boldsymbol{\varepsilon}_{p},\boldsymbol{\alpha}) := \int_{\mathcal{A}} \Psi\left(\boldsymbol{E}^{\text{tot}}(\boldsymbol{\varepsilon}_{p},\boldsymbol{\alpha})\right) j_{0} \, d\mathcal{A}. \tag{23}$$

Here, the Lagrange multiplier

$$\boldsymbol{r}_{\mathrm{p}} := \int_{A} A^{\mathrm{T}} \underline{\boldsymbol{S}_{\mathrm{p}}} j_{0} \, \mathrm{d}\mathcal{A} \in \mathbb{R}^{d_{\mathrm{p}}}$$
 (24)

for the compatibility condition, $\varepsilon(y) = \varepsilon_p$, in Eq. (22) represents an array of *physical* stress resultants. Note that each component of the array r_p is an independent variable, a work-conjugate to the corresponding component in the array ε_p . Taking the first variation of Eq. (22), we obtain the internal virtual work (Choi et al. [9])

$$G_{\text{int}}^{\text{HW}} \equiv \delta U_{\text{HW}} = G_{\text{int}}^{\text{y}}(\mathbf{y}, \mathbf{r}_{\text{p}}, \delta \mathbf{y}) + G_{\text{int}}^{\text{r}}(\mathbf{y}, \boldsymbol{\varepsilon}_{\text{p}}, \delta \mathbf{r}_{\text{p}}) + G_{\text{int}}^{\varepsilon}(\mathbf{r}_{\text{p}}, \boldsymbol{\varepsilon}_{\text{p}}, \boldsymbol{\alpha}, \delta \boldsymbol{\varepsilon}_{\text{p}}),$$

$$+ G_{\text{int}}^{a}(\boldsymbol{\varepsilon}_{\text{p}}, \boldsymbol{\alpha}, \delta \boldsymbol{\alpha}), \tag{25}$$



where we have defined

$$G_{\text{int}}^{\mathbf{y}}(\mathbf{y}, \mathbf{r}_{\mathbf{p}}, \delta \mathbf{y}) := \int_{0}^{L} \delta \mathbf{y} \cdot \mathbb{B}(\mathbf{y})^{\mathsf{T}} \mathbf{r}_{\mathbf{p}} \, \mathrm{d}s, \tag{26a}$$

$$G_{\text{int}}^{\text{r}}(\boldsymbol{y}, \boldsymbol{\varepsilon}_{\text{p}}, \delta \boldsymbol{r}_{\text{p}}) := \int_{0}^{L} \delta \boldsymbol{r}_{\text{p}} \cdot \left\{ \boldsymbol{\varepsilon}(\boldsymbol{y}) - \boldsymbol{\varepsilon}_{\text{p}} \right\} \mathrm{d}s, \tag{26b}$$

$$G_{\text{int}}^{\varepsilon}(\boldsymbol{r}_{p},\boldsymbol{\varepsilon}_{p},\boldsymbol{\alpha},\delta\boldsymbol{\varepsilon}_{p}) := \int_{0}^{L} \delta\boldsymbol{\varepsilon}_{p} \cdot \left\{ \partial_{\boldsymbol{\varepsilon}_{p}} \psi(\boldsymbol{\varepsilon}_{p},\boldsymbol{\alpha}) - \boldsymbol{r}_{p} \right\} ds, \tag{26c}$$

$$G_{\text{int}}^{a}(\boldsymbol{\varepsilon}_{p},\boldsymbol{\alpha},\delta\boldsymbol{\alpha}) := \int_{0}^{L} \delta\boldsymbol{\alpha} \cdot \partial_{\boldsymbol{\alpha}} \psi(\boldsymbol{\varepsilon}_{p},\boldsymbol{\alpha}) \, \mathrm{d}s. \tag{26d}$$

Here, $\delta(\bullet)$ denotes the first variation, and we have defined

$$\partial_{\boldsymbol{\varepsilon}_{p}} \psi \left(\boldsymbol{\varepsilon}_{p}, \boldsymbol{\alpha}\right) := \int_{\mathcal{A}} A^{T} \underline{\partial_{E} \Psi|_{E=E^{\text{tot}}}} j_{0} \, d\mathcal{A}, \tag{27a}$$

$$\partial_{\alpha} \psi \left(\boldsymbol{\varepsilon}_{\mathbf{p}}, \boldsymbol{\alpha} \right) := \int_{\mathcal{A}} \mathbf{\Gamma}^{\mathrm{T}} \underline{\partial_{E} \Psi|_{E = E^{\mathrm{tot}}}} j_{0} \, \mathrm{d} \mathcal{A}, \tag{27b}$$

with $\partial_E \Psi := \partial \Psi(E)/\partial E$. The detailed expression of the operator $\mathbb{B}(y)$, defined by

$$\delta \boldsymbol{\varepsilon} = \mathbb{B}(\boldsymbol{y}) \,\delta \boldsymbol{y},\tag{28}$$

is given in Choi et al. [7, Eq. (A.4.5)].

3.1.2 Work function due to external loads

We define a work function for time-dependent external loads,

$$W_{\text{ext}}(\boldsymbol{y},t) := \int_{0}^{L} \boldsymbol{y} \cdot \boldsymbol{\bar{R}}(t) \, \mathrm{d}s + \left[\boldsymbol{y} \cdot \boldsymbol{\bar{R}}_{0}(t) \right]_{s \in \Gamma_{N}}, \ t \in [0,T], \tag{29}$$

where (Choi et al. [7])

$$\bar{\mathbf{R}} := \left\{ \begin{array}{l} \bar{\mathbf{n}} \\ \bar{\tilde{\mathbf{m}}}^1 \\ \bar{\tilde{\mathbf{m}}}^2 \end{array} \right\}, \ \bar{\mathbf{R}}_0 := \left\{ \begin{array}{l} \bar{\mathbf{n}}_0 \\ \bar{\tilde{\mathbf{m}}}_0^1 \\ \bar{\tilde{\mathbf{n}}}_0^2 \end{array} \right\}. \tag{30}$$

Here, \bar{n} and $\bar{\tilde{m}}^{\alpha}$ ($\alpha \in \{1,2\}$) denote the given external (distributed) force and director couple along the center axis, respectively, that follow from the body force and the lateral-surface tractions. Further, \bar{n}_0 and $\bar{\tilde{m}}_0^{\alpha}$ denote the prescribed force and director couple at the Neumann boundary, $\Gamma_N \subset \emptyset \cup \{0, L\}$, due to a given surface traction acting on the end faces. \emptyset denotes the empty set. Then, by taking the first variation of the work function in Eq. (29), the external virtual work at time $t \in [0, T]$ can be expressed as

$$G_{\text{ext}}(\delta \mathbf{y}, t) = \delta W_{\text{ext}} = \int_0^L \delta \mathbf{y} \cdot \bar{\mathbf{R}}(t) ds + \left[\delta \mathbf{y} \cdot \bar{\mathbf{R}}_0(t) \right]_{s \in \Gamma_{\text{N}}}.$$
 (31)

Note that, here, we assume that the external load is deformation-independent. If the external load has no explicit time dependence, the system is *conservative*, i.e., the total energy of



the system is conserved (Lanczos [25, Sects. 1.7 and 1.8]), and we may define an external potential energy as

$$U_{\text{ext}}(\mathbf{y}) = -W_{\text{ext}}(\mathbf{y}). \tag{32}$$

3.1.3 Kinetic energy

We define an independent (material) velocity vector \mathbf{v} that satisfies

$$\mathbf{v} = \dot{\mathbf{x}}.\tag{33}$$

From the beam kinematics in Eq. (7), we may also decompose the independent velocity field into two parts: the center axis velocity $\mathbf{v}_{\varphi} \equiv \mathbf{v}_{\varphi}(s,t) \in \mathbb{R}^3$ and the director velocity $\mathbf{v}_{\alpha} \equiv \mathbf{v}_{\alpha}(s,t) \in \mathbb{R}^3$ ($\alpha \in \{1,2\}$) that satisfy the *compatibility* condition

$$V = \dot{\mathbf{y}},\tag{34}$$

where, for brevity, we define a *generalized velocity* $V := \left[\mathbf{v}_{\varphi}^{\mathrm{T}}, \mathbf{v}_{1}^{\mathrm{T}}, \mathbf{v}_{2}^{\mathrm{T}}\right]^{\mathrm{T}} \in \mathbb{R}^{d}$. Then, we can express the *total linear momentum* as

$$L(V) = \int_0^L p(V) \, \mathrm{d}s,\tag{35}$$

where we have defined the local linear momentum per undeformed unit arc-length,

$$p(V) := \rho_A \mathbf{v}_{\varphi} + I_{\varrho}^{\alpha} \mathbf{v}_{\alpha}, \tag{36}$$

with the mass per undeformed unit arc-length, i.e., the line mass density,

$$\rho_A := \int_A \rho_0 \, j_0 \, \mathrm{d}\mathcal{A},\tag{37}$$

and the first mass moment of inertia per undeformed unit arc-length,

$$I_{\rho}^{\alpha} := \int_{A} \rho_0 \, \zeta^{\alpha} \, j_0 \, \mathrm{d}\mathcal{A}, \, \, \alpha \in \{1, 2\}.$$
 (38)

We can then express the total angular momentum as

$$J(y, V) = \int_0^L \{ \boldsymbol{\varphi} \times \boldsymbol{p}(V) + \boldsymbol{d}_{\alpha} \times \boldsymbol{\mu}^{\alpha}(V) \} \, \mathrm{d}s, \tag{39}$$

where we have defined the local director momentum per undeformed unit arc-length,

$$\boldsymbol{\mu}^{\alpha}(V) := I_{\rho}^{\alpha} \boldsymbol{v}_{\varphi} + I_{\rho}^{\alpha\beta} \boldsymbol{v}_{\beta}, \ \alpha \in \{1, 2\}, \tag{40}$$

with the second mass moment of inertia per undeformed unit arc-length,

$$I_{\rho}^{\alpha\beta} := \int_{\mathcal{A}} \rho_0 \, \zeta^{\alpha} \zeta^{\beta} j_0 \, \mathrm{d}\mathcal{A}, \ \alpha, \beta \in \{1, 2\}.$$
 (41)



For brevity, we combine Eqs. (36) and (40) into

with the mass moment of inertia

$$\mathcal{M} := \begin{bmatrix} \rho_A \mathbf{1} & I_{\rho}^1 \mathbf{1} & I_{\rho}^2 \mathbf{1} \\ & I_{\rho}^{11} \mathbf{1} & I_{\rho}^{12} \mathbf{1} \\ \text{sym.} & I_{\rho}^{22} \mathbf{1} \end{bmatrix}, \tag{42b}$$

which is time-independent. It should be noted that \mathcal{M} is symmetric (i.e., $I_{\rho}^{12} = I_{\rho}^{21}$), regardless of the cross-section's shape and initial mass distribution. Further, we define the total kinetic energy as

$$\mathcal{K}(V) = \frac{1}{2} \int_0^L V \cdot \mathcal{M}V \, \mathrm{d}s. \tag{43}$$

Accordingly, we have an inertial contribution to the virtual work,

$$G_{\text{iner}}(\dot{V}, \delta y) := \int_{0}^{L} \delta y \cdot \mathcal{M} \dot{V} \, \mathrm{d}s. \tag{44}$$

3.2 Euler-Lagrange equations: Livens' theorem

Based on Livens' theorem (Pars [30, Sect. 26.2]), we define the Lagrangian with the independent velocity field V as

$$\mathcal{L} := \mathcal{K}(\mathbf{V}) - U_{\text{HW}}\left(\boldsymbol{\varepsilon}(\mathbf{y}), \boldsymbol{\varepsilon}_{\text{p}}, \boldsymbol{r}_{\text{p}}, \boldsymbol{\alpha}\right) + W_{\text{ext}}(\mathbf{y}, t) + \int_{0}^{L} (\dot{\mathbf{y}} - \mathbf{V}) \cdot \begin{Bmatrix} \boldsymbol{\lambda}_{\varphi} \\ \boldsymbol{\lambda}^{1} \\ \boldsymbol{\lambda}^{2} \end{Bmatrix} ds, \qquad (45)$$

where the last term represents a weak enforcement of the compatibility condition in Eq. (34). Here, the Lagrange multipliers λ_{φ} , $\lambda^{\alpha} \in \mathbb{R}^3$, $\alpha \in \{1, 2\}$ can be identified by the stationarity of the Lagrangian with respect to the independent velocities v_{φ} and v_{α} ($\alpha \in \{1, 2\}$), as (Betsch and Janz [1, Sect. 3.2])

$$\lambda_{\varphi} = p(V), \tag{46a}$$

$$\lambda^{\alpha} = \mu^{\alpha}(V), \ \alpha \in \{1, 2\}. \tag{46b}$$

Substituting Eqs. (46a)–(46b) into Eq. (45), the Lagrangian can be rewritten as

$$\mathcal{L} = \int_{0}^{L} \left(\dot{\boldsymbol{y}} - \frac{1}{2} \boldsymbol{V} \right) \cdot \boldsymbol{\mathcal{M}} \boldsymbol{V} \, \mathrm{d}s - U_{\mathrm{HW}} \left(\boldsymbol{\varepsilon}(\boldsymbol{y}), \boldsymbol{\varepsilon}_{\mathrm{p}}, \boldsymbol{r}_{\mathrm{p}}, \boldsymbol{\alpha} \right) + W_{\mathrm{ext}}(\boldsymbol{y}, t), \tag{47}$$

from which we define an action integral in the time span $[0, T] \ni t$,

$$S := \int_0^T \mathcal{L}(\boldsymbol{V}, \boldsymbol{y}, \boldsymbol{\varepsilon}_{\mathbf{p}}, \boldsymbol{r}_{\mathbf{p}}, \boldsymbol{\alpha}, t) \, \mathrm{d}t. \tag{48}$$

By the stationarity condition of the action integral, we obtain the following weak form of the initial-boundary-value problem: Find the solution $\{V, y, r_p, \varepsilon_p, \alpha\} \in \overline{\mathcal{V}} \times \mathcal{V} \times \mathcal{V}_p \times \mathcal{V}_p \times \mathcal{V}_a$ at time $t \in (0, T)$ such that

$$G_{\text{iner}}(\dot{\mathbf{V}}, \delta \mathbf{y}) + G_{\text{int}}^{y}(\mathbf{y}, \mathbf{r}_{p}, \delta \mathbf{y}) = G_{\text{ext}}(\delta \mathbf{y}, t), \tag{49a}$$

$$G_{\text{int}}^{\text{r}}(\mathbf{y}, \boldsymbol{\varepsilon}_{\text{p}}, \delta \boldsymbol{r}_{\text{p}}) = 0,$$
 (49b)

$$G_{\text{int}}^{\varepsilon}(\mathbf{r}_{\text{p}}, \boldsymbol{\varepsilon}_{\text{p}}, \boldsymbol{\alpha}, \delta \boldsymbol{\varepsilon}_{\text{p}}) = 0,$$
 (49c)

$$G_{\rm int}^{\rm a}(\boldsymbol{\varepsilon}_{\rm p},\boldsymbol{\alpha},\delta\boldsymbol{\alpha})=0,$$
 (49d)

are satisfied for all $\{\delta y, \delta r_p, \delta \varepsilon_p, \delta \alpha\} \in \bar{\mathcal{V}} \times \mathcal{V}_p \times \mathcal{V}_a$, together with the *compatibility* condition from Eq. (34),

$$\int_{0}^{L} \delta \mathbf{V} \cdot \mathbf{\mathcal{M}} \left(\dot{\mathbf{y}} - \mathbf{V} \right) \, \mathrm{d}\mathbf{s} = 0, \ \forall \delta \mathbf{V} \in \bar{\mathcal{V}}, \tag{50}$$

and the initial conditions at $s \in (0, L)$,

$$\varphi(s,0) = \varphi_0(s), \quad d_{\alpha}(s,0) = D_{\alpha}(s), \quad \alpha \in \{1,2\},$$
 (51a)

$$\mathbf{v}_{\varphi}(s,0) = \bar{\mathbf{v}}_{\varphi}(s), \quad \mathbf{v}_{\alpha}(s,0) = \bar{\mathbf{v}}_{\alpha}(s), \quad \alpha \in \{1,2\}.$$
 (51b)

Here, $\bar{v}_{\varphi}(s)$ and $\bar{v}_{\alpha}(s)$ denote the given initial velocity fields for the center axis and directors, respectively. We define the kinematically admissible solution space for the configuration variable \mathbf{v} as

$$\mathcal{V} := \left\{ y \in \left[H^1(0, L) \right]^d \middle| \varphi = \bar{\varphi}, \ d_1 = \bar{d}_1, \text{ and } d_2 = \bar{d}_2 \text{ at } s \in \Gamma_D \right\}, \tag{52}$$

where $\Gamma_D \subset \emptyset \cup \{0, L\}$ denotes the Dirichlet boundary such that $\Gamma_D \cup \Gamma_N = \{0, L\}$ and $\Gamma_D \cap \Gamma_N = \emptyset$. The variational space is defined as

$$\bar{\mathcal{V}} := \left\{ \delta \mathbf{y} \in \left[H^1(0, L) \right]^d \middle| \delta \mathbf{\varphi} = \delta \mathbf{d}_1 = \delta \mathbf{d}_2 = \mathbf{0} \text{ at } s \in \Gamma_{\mathbf{D}} \right\}.$$
 (53)

Further, we employ the solution spaces for the physical variables, $V_p := [L^2(0, L)]^{d_p}$, and $V_a := [L^2(0, L)]^{d_a}$ such that those solution functions may have discontinuity in the domain. Further comment on the selection of the approximation space is given in Sects. 5.2.2 and 5.2.3.

4 Temporal discretization: an EMC scheme

We divide a given time span [0, T] into N intervals, and find approximate solutions at discrete times $\{t_0 = 0, t_1, \dots, t_n, t_{n+1}, \dots, t_N = T\}$, where n is a nonnegative integer. Note that time increment $\Delta t := t_{n+1} - t_n$ can vary adaptively. We denote the approximation of the independent solution variables at time t_n by ${}^n(\bullet)$. In the given time interval $[t_n, t_{n+1}]$, our goal is to find an approximate solution at time t_{n+1} for a given solution at time t_n . We present an implicit time-stepping scheme, based on the previous work in Betsch and Janz [1], which exactly preserves the total energy for linear constitutive laws. A further extension to non-linear constitutive laws, e.g., using the discrete derivative of Gonzalez [16] remains future



work. The present time-stepping scheme turns out to exhibit superior numerical stability in comparison to standard trapezoidal and mid-point rules. We first approximate the first-order time derivatives at the mid-point as

$$\dot{\mathbf{y}}\left(t_n + \frac{1}{2}\Delta t\right) \approx \frac{1}{\Delta t} \left(^{n+1}\mathbf{y} - {}^{n}\mathbf{y}\right),$$
 (54a)

and

$$\dot{V}\left(t_n + \frac{1}{2}\Delta t\right) \approx \frac{1}{\Delta t} \left({}^{n+1}V - {}^{n}V\right). \tag{54b}$$

Further, we define the mid-point approximations

$$y\left(t_n + \frac{1}{2}\Delta t\right) \approx \frac{1}{2} \binom{n+1}{y} + \binom{n}{y} =: \binom{n+\frac{1}{2}}{y}, \tag{55a}$$

$$\mathbf{r}_{\mathrm{p}}\left(t_{n}+\frac{1}{2}\Delta t\right)\approx\frac{1}{2}\left(^{n+1}\mathbf{r}_{\mathrm{p}}+^{n}\mathbf{r}_{\mathrm{p}}\right)=:^{n+\frac{1}{2}}\mathbf{r}_{\mathrm{p}},$$
 (55b)

$$\boldsymbol{\varepsilon}_{\mathrm{p}}\left(t_{n}+\frac{1}{2}\Delta t\right) \approx \frac{1}{2}\left(^{n+1}\boldsymbol{\varepsilon}_{\mathrm{p}}+^{n}\boldsymbol{\varepsilon}_{\mathrm{p}}\right) =: ^{n+\frac{1}{2}}\boldsymbol{\varepsilon}_{\mathrm{p}},$$
 (55c)

$$\alpha \left(t_n + \frac{1}{2} \Delta t \right) \approx \frac{1}{2} \left({}^{n+1} \alpha + {}^{n} \alpha \right) =: {}^{n+\frac{1}{2}} \alpha,$$
 (55d)

and

$$V\left(t_n + \frac{1}{2}\Delta t\right) \approx \frac{1}{2} \left(^{n+1}V + {}^nV\right) =: {}^{n+\frac{1}{2}}V.$$
 (55e)

From the compatibility condition in Eq. (50) at the mid-point, combining Eqs. (54a) and (55e) gives (Betsch and Janz [1, Sect. 3.8])

$${}^{n+1}V = \frac{2}{\Delta t} \left({}^{n+1}y - {}^{n}y \right) - {}^{n}V.$$
 (56)

Then, substituting Eq. (56) into Eq. (54b) yields

$$\dot{V}\left(t_n + \frac{1}{2}\Delta t\right) \approx \frac{2}{\Delta t^2} \binom{n+1}{y} - \binom{n}{y} - \frac{2}{\Delta t} \binom{n}{y} V. \tag{57}$$

Substituting Eq. (57) into Eq. (49a), a time-discrete form of the momentum balance equation at the mid-point can be stated as

$$\frac{2}{\Delta t^{2}}G_{\text{iner}}\left(^{n+1}\boldsymbol{y},\delta\boldsymbol{y}\right) + G_{\text{int}}^{\boldsymbol{y}}\left(^{n+\frac{1}{2}}\boldsymbol{y},^{n+\frac{1}{2}}\boldsymbol{r}_{p},\delta\boldsymbol{y}\right)$$

$$= G_{\text{ext}}\left(\delta\boldsymbol{y},t_{n} + \frac{1}{2}\Delta t\right) + \frac{2}{\Delta t}G_{\text{iner}}\left(^{n}\boldsymbol{V} + \frac{1}{\Delta t}^{n}\boldsymbol{y},\delta\boldsymbol{y}\right),$$
(58a)

which is subject to

$$G_{\text{int}}^{\text{r}}\left(^{n+1}\boldsymbol{y},^{n+1}\boldsymbol{\varepsilon}_{\text{p}},\delta\boldsymbol{r}_{\text{p}}\right)=0,$$
 (58b)

$$G_{\text{int}}^{\varepsilon} \left({}^{n+\frac{1}{2}} \boldsymbol{r}_{p}, {}^{n+\frac{1}{2}} \boldsymbol{\varepsilon}_{p}, {}^{n+\frac{1}{2}} \boldsymbol{\alpha}, \delta \boldsymbol{\varepsilon}_{p} \right) = 0, \tag{58c}$$

$$G_{\text{int}}^{a}\left(^{n+\frac{1}{2}}\boldsymbol{\varepsilon}_{p},^{n+\frac{1}{2}}\boldsymbol{\alpha},\delta\boldsymbol{\alpha}\right)=0,$$
 (58d)

 $\forall \{\delta y, \delta r_p, \delta \varepsilon_p, \delta \alpha\} \in \bar{\mathcal{V}} \times \mathcal{V}_p \times \mathcal{V}_p \times \mathcal{V}_a$. It should be noted that, in Eq. (58b), we have imposed the compatibility between the displacement and the physical strain at time $t = t_{n+1}$ instead of $t = t_n + \frac{1}{2}\Delta t$, which is sufficient for the exact conservation of total energy for linear constitutive laws (Betsch and Janz [1]). Note that this treatment is in line with using the algorithmic stress resultant in Romero and Armero [34], which evaluates the stress resultant by taking an average between two subsequent discrete times instead of evaluating it from the nonlinear strain-configuration relation at the mid-point.

4.1 Linearization

Due to the geometrical and material nonlinearities in the internal virtual work term of Eq. (25), we need an incremental-iterative (Newton-Raphson) solution process. It can be stated as: In the (n + 1)th time step, for the given solution at the previous nth time step $\{{}^{n}V, {}^{n}y, {}^{n}r_{p}, {}^{n}e_{p}, {}^{n}\alpha\}$, and that in the previous iteration

$$\left\{ ^{n+1}\boldsymbol{y}^{(i-1)},^{n+1}\boldsymbol{r}_{\mathrm{p}}^{(i-1)},^{n+1}\boldsymbol{\varepsilon}_{\mathrm{p}}^{(i-1)},^{n+1}\boldsymbol{\alpha}^{(i-1)}\right\} ,$$

find the solution increment

$$\{\Delta y, \Delta r_{p}, \Delta \varepsilon_{p}, \Delta \alpha\} \in \bar{\mathcal{V}} \times \mathcal{V}_{p} \times \mathcal{V}_{p} \times \mathcal{V}_{a}$$

such that

$$\int_{0}^{L} \begin{cases} \delta \mathbf{y} \\ \delta \mathbf{r}_{p} \\ \delta \boldsymbol{\varepsilon}_{p} \\ \delta \boldsymbol{\alpha} \end{cases} \cdot \mathbf{k} \begin{cases} \Delta \mathbf{y} \\ \Delta \mathbf{r}_{p} \\ \Delta \boldsymbol{\varepsilon}_{p} \\ \Delta \boldsymbol{\alpha} \end{cases} ds + \frac{4}{\Delta t^{2}} \int_{0}^{L} \delta \mathbf{y} \cdot \mathcal{M} \Delta \mathbf{y} ds$$

$$= 2 G_{\text{ext}} \left(\delta \mathbf{y}, t_{n} + \frac{1}{2} \Delta t \right) + \frac{4}{\Delta t} G_{\text{iner}} \left({}^{n} \mathbf{V} - \frac{1}{\Delta t} \left({}^{n+1} \mathbf{y}^{(i-1)} - {}^{n} \mathbf{y} \right), \delta \mathbf{y} \right)$$

$$- 2 G_{\text{int}}^{\mathbf{y}} \left(\bar{\mathbf{y}}, \bar{\mathbf{r}}_{p}, \delta \mathbf{y} \right) - G_{\text{int}}^{\mathbf{r}} \left({}^{n+1} \mathbf{y}^{(i-1)}, {}^{n+1} \boldsymbol{\varepsilon}_{p}^{(i-1)}, \delta \boldsymbol{r}_{p} \right)$$

$$- 2 G_{\text{int}}^{\varepsilon} \left(\bar{\mathbf{r}}_{p}, \bar{\boldsymbol{\varepsilon}}_{p}, \bar{\boldsymbol{\alpha}}, \delta \boldsymbol{\varepsilon}_{p} \right)$$

$$- 2 G_{\text{int}}^{\varepsilon} \left(\bar{\boldsymbol{\varepsilon}}_{p}, \bar{\boldsymbol{\alpha}}, \delta \boldsymbol{\alpha} \right), \tag{59}$$

 $\forall \left\{ \delta \textbf{\textit{y}}, \delta \textbf{\textit{r}}_{p}, \delta \boldsymbol{\textit{e}}_{p}, \delta \boldsymbol{\alpha} \right\} \in \bar{\mathcal{V}} \times \mathcal{V}_{p} \times \mathcal{V}_{p} \times \mathcal{V}_{a}, \text{ and update}$

$$^{n+1}y^{(i)} = ^{n+1}y^{(i-1)} + \Delta y, \quad ^{n+1}y^{(0)} \equiv {}^{n}y,$$
 (60a)

$${}^{n+1}\boldsymbol{r}_{\rm p}^{(i)} = {}^{n+1}\boldsymbol{r}_{\rm p}^{(i-1)} + \Delta \boldsymbol{r}_{\rm p}, {}^{n+1}\boldsymbol{r}_{\rm p}^{(0)} \equiv {}^{n}\boldsymbol{r}_{\rm p},$$
 (60b)

$$^{n+1}\boldsymbol{\varepsilon}_{\mathrm{p}}^{(i)} = ^{n+1}\boldsymbol{\varepsilon}_{\mathrm{p}}^{(i-1)} + \Delta\boldsymbol{\varepsilon}_{\mathrm{p}}, \, ^{n+1}\boldsymbol{\varepsilon}_{\mathrm{p}}^{(0)} \equiv {}^{n}\boldsymbol{\varepsilon}_{\mathrm{p}},$$
 (60c)

$$^{n+1}\boldsymbol{\alpha}^{(i)} = ^{n+1}\boldsymbol{\alpha}^{(i-1)} + \Delta\boldsymbol{\alpha}, \quad ^{n+1}\boldsymbol{\alpha}^{(0)} \equiv {}^{n}\boldsymbol{\alpha},$$
 (60d)



for i = 1, 2, ..., until a convergence criteria is satisfied. After the iteration finishes, we should also update the independent velocity field using Eq. (56). In Eq. (59), we have the tangent operator

$$\boldsymbol{k} := \begin{bmatrix} \boldsymbol{Y}^{\mathrm{T}} \boldsymbol{k}_{\mathrm{G}} (\bar{\boldsymbol{r}}_{\mathrm{p}}) \boldsymbol{Y} & \mathbb{B}(\bar{\boldsymbol{y}})^{\mathrm{T}} & \boldsymbol{0}_{d \times d_{\mathrm{p}}} & \boldsymbol{0}_{d \times d_{\mathrm{a}}} \\ \mathbb{B}^{(n+1} \boldsymbol{y}^{(i-1)}) & \boldsymbol{0}_{d_{\mathrm{p}} \times d_{\mathrm{p}}} & -\boldsymbol{1}_{d_{\mathrm{p}} \times d_{\mathrm{p}}} & \boldsymbol{0}_{d_{\mathrm{p}} \times d_{\mathrm{a}}} \\ \boldsymbol{0}_{d_{\mathrm{p}} \times d} & -\boldsymbol{1}_{d_{\mathrm{p}} \times d_{\mathrm{p}}} & \mathbb{C}^{\varepsilon}_{\mathrm{p}\varepsilon}_{\mathrm{p}}(\bar{\boldsymbol{e}}_{\mathrm{p}}, \bar{\boldsymbol{\alpha}}) & \mathbb{C}^{\varepsilon}_{\mathrm{p}\varepsilon}_{\mathrm{p}\varepsilon}(\bar{\boldsymbol{e}}_{\mathrm{p}}, \bar{\boldsymbol{\alpha}})^{\mathrm{T}} \\ \boldsymbol{0}_{d_{\mathrm{a}} \times d} & \boldsymbol{0}_{d_{\mathrm{a}} \times d_{\mathrm{p}}} & \mathbb{C}^{\varepsilon}_{\mathrm{p}\varepsilon}_{\mathrm{p}}(\bar{\boldsymbol{e}}_{\mathrm{p}}, \bar{\boldsymbol{\alpha}}) & \mathbb{C}^{\mathrm{aa}}_{\mathrm{p}}(\bar{\boldsymbol{e}}_{\mathrm{p}}, \bar{\boldsymbol{\alpha}}) \end{bmatrix},$$

$$(61)$$

where we have defined the constitutive matrices as (Choi et al. [9])

$$\frac{\partial^2 \psi}{\partial \boldsymbol{\varepsilon}_{\mathbf{p}} \partial \boldsymbol{\varepsilon}_{\mathbf{p}}} = \int_{\mathcal{A}} \boldsymbol{A}^{\mathrm{T}} \underline{\underline{\boldsymbol{\mathcal{C}}}_{\mathbf{p}}} \boldsymbol{A} \, j_0 \, \mathrm{d} \mathcal{A} =: \mathbb{C}_{\mathbf{p}}^{\varepsilon \varepsilon} (\boldsymbol{\varepsilon}_{\mathbf{p}}, \boldsymbol{\alpha}) \,, \tag{62a}$$

$$\frac{\partial^2 \psi}{\partial \boldsymbol{\alpha} \partial \boldsymbol{\alpha}} = \int_{\mathcal{A}} \boldsymbol{\Gamma}^{\mathrm{T}} \underline{\underline{\boldsymbol{\mathcal{C}}}}_{p} \boldsymbol{\Gamma} j_0 \, \mathrm{d} \mathcal{A} =: \mathbb{C}_{p}^{\mathrm{aa}} (\boldsymbol{\varepsilon}_{p}, \boldsymbol{\alpha}), \tag{62b}$$

$$\frac{\partial^2 \psi}{\partial \boldsymbol{\alpha} \partial \boldsymbol{\varepsilon}_{p}} = \int_{\mathcal{A}} \boldsymbol{\Gamma}^{T} \underline{\underline{\boldsymbol{\mathcal{C}}}_{p}} \boldsymbol{A} \ j_{0} \, d\boldsymbol{\mathcal{A}} =: \mathbb{C}_{p}^{a\varepsilon} \left(\boldsymbol{\varepsilon}_{p}, \boldsymbol{\alpha} \right), \tag{62c}$$

based on the material elasticity tensor

$$C_{p} := \frac{\partial^{2} \Psi(E_{p})}{\partial E_{p} \partial E_{p}}.$$
 (63)

Here, $\underline{\bullet}$ denotes Voigt notation of a fourth-order tensor with major and minor symmetries. The detailed expressions of the operators k_G and Y can be found in Eq. (A.4.9) and (A.4.11) of Choi et al. [7], respectively. In Eq. (61), for brevity, we also employ the following notations for averaged solution variables

$$\bar{\mathbf{y}} := \frac{1}{2} \left({}^{n+1} \mathbf{y}^{(i-1)} + {}^{n} \mathbf{y} \right),$$
 (64a)

$$\bar{\boldsymbol{r}}_{p} := \frac{1}{2} \left({}^{n+1} \boldsymbol{r}_{p}^{(i-1)} + {}^{n} \boldsymbol{r}_{p} \right),$$
 (64b)

$$\bar{\boldsymbol{\varepsilon}}_{\mathbf{p}} := \frac{1}{2} \left({}^{n+1} \boldsymbol{\varepsilon}_{\mathbf{p}}^{(i-1)} + {}^{n} \boldsymbol{\varepsilon}_{\mathbf{p}} \right), \tag{64c}$$

$$\bar{\boldsymbol{\alpha}} := \frac{1}{2} \left({}^{n+1} \boldsymbol{\alpha}^{(i-1)} + {}^{n} \boldsymbol{\alpha} \right). \tag{64d}$$

Further, $\mathbf{1}_{n \times m}$ and $\mathbf{0}_{n \times m}$ denote the identity and zero matrices of dimension $n \times m$, respectively.

Remark 4.1 It should be noted that the tangent stiffness operator in Eq. (61) is unsymmetric due to $\mathbb{B}(\bar{y}) \neq \mathbb{B}^{(n+1} y^{(i-1)})$, as we enforce the compatibility condition at time $t = t_{n+1}$ instead of $t = t_n + \frac{1}{2}\Delta t$ in Eq. (58b).

Remark 4.2 In Sect. B.2, we analytically verify the energy–momentum consistency of the present EMC scheme, which holds regardless of the element sizes in temporal and spatial domains.



5 Spatial discretization: an isogeometric approach

Next, we present the spatial discretization of the configuration and physical variables in the framework of isogeometric analysis (IGA). Especially, we show that the finite element approximation of the (total) director field leads to the objectivity of the approximated (discrete) beam strain. This basically necessitates an approximation of the initial director field.

5.1 NURBS curve

In the isogeometric approach, we employ the same spline basis functions in the Computer-Aided Design (CAD) model for the analysis. This enables (i) an exact representation of the geometry of the beam's initial center axis including conic sections like a circle, and (ii) higher-order inter-element continuity which leads to the superior per DOF accuracy, compared with the conventional C^0 -finite elements, see, e.g., Choi et al. [9]. Using NURBS (Non-Uniform Rational B-Spline), the beam's initial center axis can be represented by

$$\varphi_0(s(\xi)) \equiv X(\xi) = \sum_{I=1}^{n_{\text{cp}}} N_I^p(\xi) \mathbf{P}_I,$$
 (65)

where \mathbf{P}_I denotes the position of the Ith control point, $N_I^p(\xi)$ denotes the corresponding NURBS basis function of degree p, and $n_{\rm cp}$ denotes the total number of basis functions (control points) in the given patch. Here, $\xi \in \mathcal{Z} \subset \mathbb{R}$ denotes the parametric coordinate, and the parametric domain $\mathcal{Z} := \left[\xi_1, \xi_{n_{\rm cp}+p+1}\right]$ is associated with the so called *knot vector*, which is a given sequence of nondecreasing real numbers, $\widetilde{\mathcal{Z}} = \left\{\xi_1, \xi_2, \dots, \xi_{n_{\rm cp}+p+1}\right\}$, where $\xi_i \in \mathbb{R}$ denotes the ith knot. For a more detailed description on the NURBS basis functions and mesh refinements, one may refer to Piegl and Tiller [31] and Hughes et al. [19]. The arc-length coordinate along the initial center axis can be expressed by the mapping $s(\xi) : \mathcal{Z} \to [0, L]$, (Choi et al. [7])

$$s(\xi) := \int_{\xi_1}^{\eta = \xi} \| X_{,\eta}(\eta) \| \, \mathrm{d}\eta, \tag{66}$$

whose Jacobian is

$$\tilde{j} := \frac{\mathrm{d}s}{\mathrm{d}\xi} = \|X_{,\xi}(\xi)\|. \tag{67}$$

In the following section, we often use the notation $N_{L,s}^p$ for brevity, which is defined by

$$N_{I,s}^{p} := N_{I,\xi}^{p} \frac{\mathrm{d}\xi}{\mathrm{d}s} = \frac{1}{\tilde{j}} N_{I,\xi}^{p}, \tag{68}$$

where $N_{I,\xi}^p$ denotes the differentiation of the basis function $N_I^p(\xi)$ with respect to ξ .

5.1.1 Approximation of the initial director field

For a given continuous initial director field $D_{\alpha}(\xi)$, we approximate (reconstruct) it using the NURBS basis functions of degree p_d , as

$$\boldsymbol{D}_{\alpha}^{h}(\xi) = \sum_{I=1}^{n_{\text{cp}}^{d}} N_{I}^{p_{\text{d}}}(\xi) \, \mathbf{D}_{\alpha I}, \ \xi \in \Xi, \ \alpha \in \{1, 2\},$$
 (69)



where $(\bullet)^h$ denotes a finite element approximation in space. Here, the knot vector for the basis functions is given by $\widetilde{\Xi}^d = \left\{ \xi_1^d, \xi_2^d, \dots, \xi_{n_{cp}^d + p_d + 1}^d \right\}$. We determine the control coefficients $\left\{ \mathbf{D}_{\alpha 1}, \mathbf{D}_{\alpha 2}, \dots, \mathbf{D}_{\alpha n_{cp}^d} \right\}$ $(\alpha \in \{1, 2\})$ from the set of equations,

$$\boldsymbol{D}_{\alpha}^{h}(\bar{\xi}_{J}) = \boldsymbol{D}_{\alpha}(\bar{\xi}_{J}), \ J \in \left\{1, 2, \dots, n_{\text{cp}}^{d}\right\}, \tag{70}$$

where the coordinate $\bar{\xi}_J$ is given by the Greville-abscissae (a moving average of knots) (Farin and Hansford [14, Chap. 10])

$$\bar{\xi}_J = \frac{1}{p_d} \sum_{i=1}^{p_d} \xi_{J+i}^d, \ J \in \left\{ 1, 2, \dots, n_{cp}^d \right\}.$$
 (71)

Note that, here, the collocation points from the Greville-abscissae are merely introduced to approximate the initial director field. Further, their number is the same as that of the unknown control coefficient vectors, so that Eq. (70) can be rewritten as

$$\bar{\mathbf{N}} \begin{bmatrix} \mathbf{D}_{\alpha 1}^{\mathrm{T}} \\ \mathbf{D}_{\alpha 2}^{\mathrm{T}} \\ \vdots \\ \mathbf{D}_{\alpha n_{\mathrm{cp}}^{\mathrm{d}}}^{\mathrm{T}} \end{bmatrix} = \begin{bmatrix} \boldsymbol{D}_{\alpha}^{\mathrm{T}}(\bar{\xi}_{1}) \\ \boldsymbol{D}_{\alpha}^{\mathrm{T}}(\bar{\xi}_{2}) \\ \vdots \\ \boldsymbol{D}_{\alpha}^{\mathrm{T}}(\bar{\xi}_{n_{\mathrm{cp}}^{\mathrm{d}}}) \end{bmatrix}, \ \alpha \in \{1, 2\}, \tag{72}$$

with square matrix

$$\bar{\mathbf{N}} := \begin{bmatrix}
N_1^{pd}(\bar{\xi}_1) & N_2^{pd}(\bar{\xi}_1) & \cdots & N_{n_{\text{cp}}}^{pd}(\bar{\xi}_1) \\
N_1^{pd}(\bar{\xi}_2) & N_2^{pd}(\bar{\xi}_2) & \cdots & N_{n_{\text{cp}}}^{pd}(\bar{\xi}_2) \\
\vdots & \vdots & \ddots & \vdots \\
N_1^{pd}(\bar{\xi}_{n_{\text{cp}}}) & N_2^{pd}(\bar{\xi}_{n_{\text{cp}}}) & \cdots & N_{n_{\text{cp}}}^{pd}(\bar{\xi}_{n_{\text{cp}}})
\end{bmatrix}.$$
(73)

Note that the matrix $\bar{\mathbf{N}}$ is banded with band-width p_d+1 due to the local support property of NURBS basis functions, and invertible due to their linear independence. The control coefficients of the initial director field are determined by solving Eq. (72) in *pre-processing*. A continuous director field $D_{\alpha}(\xi)$, which is evaluated on the right-hand side of Eq. (72) at the discrete points of $\bar{\mathcal{E}}^d := \left\{\bar{\xi}_1, \bar{\xi}_2, \ldots, \bar{\xi}_{n_{\mathrm{cp}}^d}\right\}$, can be obtained in various ways like with the Frenet-Serret formula and the smallest rotation method, see, e.g., Meier et al. [26], Choi and Cho [6], and references therein. In the present approach, $D_{\alpha}^h(\xi) \neq D_{\alpha}(\xi)$ in general, except for the chosen points $\xi \in \bar{\mathcal{E}}^d$. This approximation error should decrease by increasing the number of basis functions (control coefficients) or elevating the degree p_d . One may also refer to the approach to enforce the condition $D_{\alpha}^h(\xi) = D_{\alpha}(\xi)$ at the Gauss quadrature points of Dornisch et al. [12], which requires an additional matrix multiplication to obtain a square (invertible) system matrix. However, note that, in the present formulation, the beam strain components in Eqs. (12a)–(12f) still vanish at the undeformed configuration, in spite of the approximation error in the initial director field.



5.2 Spatial discretization of the variational forms

5.2.1 Patch-wise approximation of kinematic variables

We approximate the current center axis position as

$$\boldsymbol{\varphi}^{h}(\xi) = \sum_{I=1}^{n_{\text{cp}}} N_{I}^{p}(\xi) \, \boldsymbol{\varphi}_{I}, \quad \xi \in \mathcal{Z}, \tag{74}$$

with the control coefficients $\varphi_I \in \mathbb{R}^3$. Similarly, we approximate the director field as

$$\boldsymbol{d}_{\alpha}^{h}(\xi) = \sum_{I=1}^{n_{\text{cp}}^{l}} N_{I}^{p_{\text{d}}}(\xi) \, \mathbf{d}_{\alpha I}, \quad \alpha \in \{1, 2\}, \ \xi \in \Xi.$$
 (75)

In IGA, we define an *element* by a nonzero knot span. Let $\mathcal{E}_e := \left[\xi_1^e, \xi_2^e \right) \ni \xi$ denote the eth element such that $\mathcal{E} = \mathcal{E}_1 \cup \mathcal{E}_2 \cup \cdots \cup \mathcal{E}_{n_{\rm el}}$ where $n_{\rm el}$ denotes the total number of elements. For the last element (i.e., $e = n_{\rm el}$) of an open curve, we have a closed interval $\mathcal{E}_{n_{\rm el}} := \left[\xi_1^{n_{\rm el}}, \xi_2^{n_{\rm el}} \right]$ to include the end point. Further, we use the same mesh along the initial center axis for both the current center axis position and director vectors. Then, in each element, from Eqs. (74) and (75), we have

$$\boldsymbol{\varphi}^{h}(\xi) = \sum_{I=1}^{n_e} N_I^p(\xi) \, \boldsymbol{\varphi}_I^e, \quad \xi \in \Xi_e, \tag{76}$$

and

$$\boldsymbol{d}_{\alpha}^{h}(\xi) = \sum_{I=1}^{n_{e}^{d}} N_{I}^{p_{d}}(\xi) \, \mathbf{d}_{\alpha I}^{e}, \quad \xi \in \Xi_{e}, \, \alpha \in \{1, 2\},$$
 (77)

respectively, where $n_e = p + 1$ and $n_e^d = p_d + 1$ denote the numbers of local support basis functions, and φ_I^e , $\mathbf{d}_{\alpha I}^e \in \mathbb{R}^3$ denote the *I*th control coefficients in the *e*th element. Combining Eqs. (76) and (77), we obtain

$$\mathbf{y}^{h} = \left\{ \begin{array}{l} \boldsymbol{\varphi}^{h} \\ \boldsymbol{d}_{1}^{h} \\ \boldsymbol{d}_{2}^{h} \end{array} \right\} = \mathbf{N}_{e}(\xi) \, \mathbf{y}^{e}, \quad \xi \in \Xi_{e}, \tag{78}$$

with

$$\mathbf{N}_{e} := \begin{bmatrix} N_{1}^{p} \mathbf{1}_{3 \times 3} & \cdots & N_{n_{e}}^{p} \mathbf{1}_{3 \times 3} \\ & \mathbf{0}_{6 \times 3n_{e}} & \end{pmatrix} \begin{bmatrix} \mathbf{0}_{3 \times 6n_{e}^{d}} \\ N_{1}^{p_{d}} \mathbf{1}_{6 \times 6} & \cdots & N_{n_{e}^{d}}^{p_{d}} \mathbf{1}_{6 \times 6} \end{bmatrix}, \tag{79}$$

and the column array of the control coefficients in each element,

$$\mathbf{y}^e := \left[\boldsymbol{\varphi}_1^{e \, \mathrm{T}}, \dots, \boldsymbol{\varphi}_{n_e}^{e \, \mathrm{T}}, \mathbf{d}_1^{e \, \mathrm{T}}, \dots, \mathbf{d}_{n_e^{\mathrm{d}}}^{e \, \mathrm{T}} \right]^{\mathrm{T}},$$

where we have defined $\mathbf{d}_{I}^{e} := \left[\mathbf{d}_{1I}^{eT}, \mathbf{d}_{2I}^{eT}\right]^{T} \in \mathbb{R}^{6}$. Further, using the same basis functions as in Eq. (78), we discretize the independent velocity field as

$$\mathbf{V}^h = \mathbf{N}_e(\xi) \, \mathbf{V}^e, \quad \xi \in \mathcal{Z}_e. \tag{80}$$



5.2.2 Patch-wise approximation of physical stress resultants and strains

It has been shown that the element-wise approximation of the physical stress resultant and strain in an isogeometric mixed finite element formulation may still suffer from numerical locking due to the parasitic terms from the higher-order inter-element continuity in the displacement field, see Kikis and Klinkel [21], Casquero and Golestanian [5], Choi et al. [9], Sauer et al. [35], and references therein. To circumvent this, we introduce the following patch-wise approximation of the physical variables.

$$\mathbf{r}_{p}^{h}(\xi) = \mathbf{B}_{e}(\xi) \,\mathbf{r}^{e},\tag{81a}$$

$$\boldsymbol{\varepsilon}_{\mathbf{p}}^{h}(\xi) = \mathbf{B}_{e}(\xi) \, \mathbf{e}^{e}, \tag{81b}$$

 $\xi \in \Xi_e$, with

$$\mathbf{B}_{e}(\xi) := \begin{bmatrix} B_{1}^{p_{p}}(\xi)\mathbf{1}_{d_{p}\times d_{p}} & \cdots & B_{n_{e}^{p}}^{p_{p}}(\xi)\mathbf{1}_{d_{p}\times d_{p}} \end{bmatrix}, \tag{82}$$

where $B_I^{p_p}(\xi)$ denotes the *I*th B-spline basis functions of degree p_p , and $n_e^p = p_p + 1$ denotes the number of basis functions per element. Here, we choose $p_p = p - 1$ for numerical stability.

Remark 5.1 Even though this *global* approach turns out to be numerically more stable compared to the *local* (element-wise) approach with further reduced degree p_p of Choi et al. [9], the condensation of those control coefficients of the physical solution fields is computationally prohibitive. Thus, here, we include those control coefficients in the global system of equations. However, compared to the local approaches, for the same number of elements, this approach still requires a much smaller number of control coefficients due to the higher-order continuity of the B-spline functions $B_I^{p_p}$.

In Sect. 6.3, we present a numerical example to compare DOF numbers between the global ('mix.glo') and local ('mix.loc-ur', 'mix.loc-sr') approaches.

5.2.3 Element-wise approximation of enhanced strains

It should be noted that, in contrast to the physical strain (ε_p) , the enhanced strain (α) has no associated geometric strain from the configuration variable. This may simplify the selection of approximate solution space for the enhanced strain field, without any special care about numerical locking. In order to facilitate an element-wise elimination (condensation) of the control coefficients, we allow inter-element discontinuity. We utilize Lagrange polynomial basis functions in the parametric domain $[-1,1) \ni \bar{\xi}$ which can be mapped to the interval $\Xi_e := [\xi_1^e, \xi_2^e) \ni \xi$ via

$$\bar{\xi} = 1 - 2\left(\frac{\xi_2^e - \xi}{\xi_2^e - \xi_1^e}\right). \tag{83}$$

Then, we approximate the enhanced strain as

$$\boldsymbol{\alpha}^{h}(\bar{\xi}) = \mathbf{L}_{e}(\bar{\xi}) \boldsymbol{\alpha}^{e}, \tag{84}$$



with

$$\mathbf{L}_{e}(\bar{\xi}) := \begin{bmatrix} L_{1}^{p_{a}}(\bar{\xi})\mathbf{1}_{d_{a}\times d_{a}} & \cdots & L_{n_{e}^{a}}^{p_{a}}(\bar{\xi})\mathbf{1}_{d_{a}\times d_{a}} \end{bmatrix}, \tag{85}$$

and the nodal coefficients α^e in the *e*th element. Here, $L_I^{p_a}$ denotes the *I*th basis function of degree p_a , and $n_a^e = p_a + 1$ denotes the number of basis functions per element. Here, without any special care about numerical instability, we simply choose the lowest degree of the basis functions, $p_a = 0$.

5.3 Discretization of linearized variational equation

We first discretize the external virtual work in Eq. (59), using Eq. (78), as

$$G_{\text{ext}}\left(\delta \mathbf{y}, t_n + \frac{1}{2}\Delta t\right) \approx \delta \mathbf{y}^{\text{T}} \mathbf{F}_{\text{ext}}\left(t_n + \frac{1}{2}\Delta t\right),$$
 (86)

where we have defined the global external load vector

$$\mathbf{F}_{\text{ext}}(t) := \mathbf{A}_{e=1}^{n_{\text{el}}} \mathbf{f}_{\text{ext}}^{e}(t) + \mathbf{A} \left[\bar{\mathbf{R}}_{0}(t) \right]_{s \in \Gamma_{\text{N}}}, \tag{87}$$

from the element external load vector due to the distributed load,

$$\mathbf{f}_{\text{ext}}^{e}(t) := \int_{\mathcal{Z}_{e}} \mathbf{N}_{e}^{\text{T}} \bar{\mathbf{R}}(t) \, \tilde{j} \, \mathrm{d}\xi. \tag{88}$$

Here, \mathbf{A} denotes the finite element assembly operator, and \mathbf{y} denotes the global array of the control coefficients for the approximated configuration variable \mathbf{y}^h . Then, the linearized equation (59) is discretized as

$$\sum_{e=1}^{n_{\text{el}}} \delta \mathbf{y}^{e\text{T}} \left\{ \left(\mathbf{k}_{yy}^{e} + \frac{4}{\Delta t^{2}} \mathbf{m}^{e} \right) \Delta \mathbf{y}^{e} + \mathbf{k}_{yr}^{e} \Delta \mathbf{r}^{e} \right\} = \delta \mathbf{y}^{\text{T}} 2 \mathbf{F}_{\text{ext}} \left(t_{n} + \frac{1}{2} \Delta t \right) - \sum_{n_{\text{el}}}^{n_{\text{el}}} \delta \mathbf{y}^{e\text{T}} 2 \left(\mathbf{f}_{y}^{e} + \frac{2}{\Delta t} \mathbf{f}_{\rho}^{e} \right),$$
(89a)

and

$$\sum_{e=1}^{n_{\text{el}}} \delta \mathbf{r}^{e\text{T}} \left(\mathbf{k}_{\text{ry}}^{e} \Delta \mathbf{y}^{e} + \mathbf{k}_{\text{re}}^{e} \Delta \mathbf{e}^{e} \right) = -\sum_{e=1}^{n_{\text{el}}} \delta \mathbf{r}^{e\text{T}} \mathbf{f}_{\text{r}}^{e}, \tag{89b}$$

$$\sum_{e=1}^{n_{\text{el}}} \delta \mathbf{e}^{e^{\mathsf{T}}} \left(\mathbf{k}_{r\varepsilon}^{e^{\mathsf{T}}} \Delta \mathbf{r}^{e} + \mathbf{k}_{\varepsilon\varepsilon}^{e} \Delta \mathbf{e}^{e} + \mathbf{k}_{a\varepsilon}^{e^{\mathsf{T}}} \Delta \boldsymbol{\alpha}^{e} \right) = -2 \sum_{e=1}^{n_{\text{el}}} \delta \mathbf{e}^{e^{\mathsf{T}}} \mathbf{f}_{\varepsilon}^{e}, \tag{89c}$$

$$\mathbf{k}_{\mathrm{a}\varepsilon}^{e} \Delta \mathbf{e}^{e} + \mathbf{k}_{\mathrm{a}a}^{e} \Delta \boldsymbol{\alpha}^{e} = -2 \,\mathbf{f}_{\mathrm{a}}^{e}, \text{ for } e \in \{1, \dots, n_{\mathrm{el}}\},$$
(89d)

where we have defined the following elemental matrices for the inertial contribution

$$\mathbf{m}^e := \int_{\mathcal{Z}_e} \mathbf{N}_e^{\mathrm{T}} \, \mathcal{M} \, \mathbf{N}_e \, \tilde{j} \, \mathrm{d}\xi \,, \tag{90}$$



$$\mathbf{f}_{\rho}^{e} := \int_{\mathcal{Z}_{e}} \mathbf{N}_{e}^{\mathsf{T}} \mathcal{M} \left\{ \frac{1}{\Delta t} \left({}^{n+1} \mathbf{y}^{h(i-1)} - {}^{n} \mathbf{y}^{h} \right) - {}^{n} \mathbf{V}^{h} \right\} \tilde{j} \, \mathrm{d}\xi . \tag{91}$$

We also have

$$\mathbf{k}_{yy}^{e} := \int_{\Xi_{e}} \mathbb{Y}_{e}^{\mathsf{T}} \mathbf{k}_{\mathsf{G}}(\bar{\mathbf{r}}_{\mathsf{p}}^{h}) \mathbb{Y}_{e} \, \tilde{j} \, \mathrm{d}\xi, \tag{92a}$$

$$\mathbf{k}_{\mathrm{yr}}^{e} := \int_{\Xi_{e}} \mathbb{B}_{e}^{\mathrm{T}} (\bar{\mathbf{y}}^{h}) \, \mathbf{B}_{e} \, \tilde{j} \, \mathrm{d}\xi, \tag{92b}$$

$$\mathbf{k}_{ry}^{e} := \int_{\Xi_{e}} \mathbf{B}_{e}^{T} \mathbb{B}_{e} \binom{n+1}{2} \mathbf{y}^{h(i-1)} \tilde{j} \, \mathrm{d}\xi, \tag{92c}$$

$$\mathbf{k}_{\mathrm{r}\varepsilon}^{e} := -\int_{\mathcal{Z}_{e}} \mathbf{B}_{e}^{\mathrm{T}} \mathbf{B}_{e} \, \tilde{j} \, \mathrm{d}\xi, \tag{92d}$$

$$\mathbf{k}_{\varepsilon\varepsilon}^{e} := \int_{\Xi_{e}} \mathbf{B}_{e}^{\mathrm{T}} \mathbb{C}_{p}^{\varepsilon\varepsilon} \left(\tilde{\boldsymbol{\varepsilon}}_{p}^{h}, \tilde{\boldsymbol{\alpha}}^{h} \right) \mathbf{B}_{e} \, \tilde{j} \, \mathrm{d}\boldsymbol{\xi}, \tag{92e}$$

$$\mathbf{k}_{\mathrm{a}\varepsilon}^{e} := \int_{\mathcal{E}_{e}} \mathbf{L}_{e}^{\mathrm{T}} \, \mathbb{C}_{\mathrm{p}}^{\mathrm{a}\varepsilon} \Big(\bar{\boldsymbol{e}}_{\mathrm{p}}^{h}, \bar{\boldsymbol{\alpha}}^{h} \Big) \, \mathbf{B}_{e} \, \tilde{j} \, \mathrm{d}\boldsymbol{\xi}, \tag{92f}$$

$$\mathbf{k}_{\mathrm{aa}}^{e} := \int_{\Xi_{e}} \mathbf{L}_{e}^{\mathrm{T}} \, \mathbb{C}_{\mathrm{p}}^{\mathrm{aa}} \Big(\bar{\boldsymbol{e}}_{\mathrm{p}}^{h}, \bar{\boldsymbol{\alpha}}^{h} \Big) \, \mathbf{L}_{e} \, \tilde{j} \, \mathrm{d}\boldsymbol{\xi}, \tag{92g}$$

and

$$\mathbf{f}_{\mathbf{y}}^{e} := \int_{\mathcal{Z}_{e}} \mathbb{B}_{e}^{\mathbf{T}} (\bar{\mathbf{y}}^{h}) \, \bar{\mathbf{r}}_{\mathbf{p}}^{h} \, \tilde{\mathbf{j}} \, \mathrm{d}\boldsymbol{\xi}, \tag{93a}$$

$$\mathbf{f}_{\mathbf{r}}^{e} := \int_{\mathcal{Z}_{e}} \mathbf{B}_{e}^{\mathsf{T}} \left\{ \boldsymbol{\varepsilon} \left({}^{n+1} \boldsymbol{y}^{h (i-1)} \right) - {}^{n+1} \boldsymbol{\varepsilon}_{\mathbf{p}}^{h (i-1)} \right\} \tilde{j} \, \mathrm{d} \boldsymbol{\xi}, \tag{93b}$$

$$\mathbf{f}_{\varepsilon}^{e} := \int_{\mathcal{E}_{e}} \mathbf{B}_{e}^{T} \left\{ \partial_{\boldsymbol{\varepsilon}_{p}} \psi \left(\bar{\boldsymbol{\varepsilon}}_{p}^{h}, \bar{\boldsymbol{\alpha}}^{h} \right) - \bar{\boldsymbol{r}}_{p}^{h} \right\} \tilde{j} \, \mathrm{d}\boldsymbol{\xi}, \tag{93c}$$

$$\mathbf{f}_{\mathbf{a}}^{e} := \int_{\Xi_{e}} \mathbf{L}_{e}^{T} \, \partial_{\boldsymbol{\alpha}} \psi \left(\bar{\boldsymbol{\varepsilon}}_{\mathbf{p}}^{h}, \bar{\boldsymbol{\alpha}}^{h} \right) \, \tilde{j} \, \mathrm{d}\boldsymbol{\xi} \,, \tag{93d}$$

where the matrices \mathbb{B}_e and \mathbb{Y}_e are defined by Eqs. (A.5.2) and (A.5.4) in Choi et al. [7], respectively. Here, from Eqs. (64a)–(64d), we also define

$$\begin{split} & \bar{\mathbf{y}}^h := \frac{1}{2} \begin{pmatrix} {^{n+1}}\mathbf{y}^{h(i-1)} + {^n}\mathbf{y}^h \end{pmatrix}, \\ & \bar{\mathbf{r}}_p^h := \frac{1}{2} \begin{pmatrix} {^{n+1}}\mathbf{r}_p^{h(i-1)} + {^n}\mathbf{r}_p^h \end{pmatrix}, \\ & \bar{\boldsymbol{\varepsilon}}_p^h := \frac{1}{2} \begin{pmatrix} {^{n+1}}\boldsymbol{\varepsilon}_p^{h(i-1)} + {^n}\boldsymbol{\varepsilon}_p^h \end{pmatrix}, \\ & \bar{\boldsymbol{\alpha}}^h := \frac{1}{2} \begin{pmatrix} {^{n+1}}\boldsymbol{\alpha}^{h(i-1)} + {^n}\boldsymbol{\alpha}^h \end{pmatrix}. \end{split}$$



Note that Eq. (89d) is given element-wisely, since we allow discontinuity in the enhanced strain field α^h . In contrast, Eqs. (89a)–(89c) are given by a global equation, due to the higher-order continuity in the solution fields y^h , r_p^h , and ε_p^h .

5.3.1 Condensation and solution update

The unknown nodal variables of the enhanced strain field α^e can be eliminated from the global finite element equation, via element-wise condensation. As the matrix \mathbf{k}_{aa}^e is invertible, from Eq. (89d), we obtain

$$\Delta \boldsymbol{\alpha}^{e} = -\mathbf{k}_{aa}^{e-1} \left(2 \, \mathbf{f}_{a}^{e} + \mathbf{k}_{a\varepsilon}^{e} \Delta \mathbf{e}^{e} \right), \text{ for } e \in \{1, \dots, n_{el}\}.$$
 (95a)

Substituting Eq. (95a) into Eq. (89c), we obtain

$$\sum_{e=1}^{n_{\text{el}}} \delta \mathbf{e}^{eT} \left(\mathbf{k}_{r_{\varepsilon}}^{eT} \Delta \mathbf{r}^{e} + \mathbb{k}_{\varepsilon \varepsilon}^{e} \Delta \mathbf{e}^{e} \right) = -2 \sum_{e=1}^{n_{\text{el}}} \delta \mathbf{e}^{eT} \mathbb{f}_{\varepsilon}^{e}, \tag{96}$$

with

$$\mathbf{f}_{\varepsilon}^{e} := \mathbf{f}_{\varepsilon}^{e} - \mathbf{k}_{a\varepsilon}^{eT} \mathbf{k}_{aa}^{e-1} \mathbf{f}_{a}^{e}, \tag{97}$$

$$\mathbf{k}_{\varepsilon\varepsilon}^{e} := \mathbf{k}_{\varepsilon\varepsilon}^{e} - \mathbf{k}_{a\varepsilon}^{e \, \mathrm{T}} \mathbf{k}_{aa}^{e \, -1} \mathbf{k}_{a\varepsilon}^{e}. \tag{98}$$

Combining Eqs. (89a), (89b), and (96), we obtain

$$\sum_{e=1}^{n_{\text{el}}} \left\{ \frac{\delta \mathbf{y}^{e}}{\delta \mathbf{r}^{e}} \right\}^{\text{T}} \mathbb{k}^{e} \left\{ \frac{\Delta \mathbf{y}^{e}}{\Delta \mathbf{r}^{e}} \right\} = \delta \mathbf{y}^{\text{T}} \left\{ 2 \mathbf{F}_{\text{ext}} \left(t_{n} + \frac{1}{2} \Delta t \right) \right\} - \sum_{e=1}^{n_{\text{el}}} \left\{ \frac{\delta \mathbf{y}^{e}}{\delta \mathbf{e}^{e}} \right\}^{\text{T}} (2 \, \mathbb{f}^{e}), \tag{99}$$

with

$$\mathbb{k}^{e} := \begin{bmatrix} \left(\mathbf{k}_{yy}^{e} + \frac{4}{\Delta t^{2}} \mathbf{m}^{e}\right) & \mathbf{k}_{yr}^{e} & \mathbf{0}_{m_{e} \times m_{e}^{p}} \\ \mathbf{k}_{ry}^{e} & \mathbf{0}_{m_{e}^{p} \times m_{e}^{p}} & \mathbf{k}_{r\varepsilon}^{e} \\ \mathbf{0}_{m_{e}^{p} \times m_{e}} & \mathbf{k}_{\varepsilon\varepsilon}^{eT} & \mathbb{k}_{\varepsilon\varepsilon}^{e} \end{bmatrix}$$
(100)

and

$$\mathbf{f}^e := \left\{ \begin{array}{l} \mathbf{f}_y^e + \frac{2}{\Delta t} \mathbf{f}_\rho^e \\ \frac{1}{2} \mathbf{f}_r^e \\ \mathbf{f}_\varepsilon^e \end{array} \right\},$$
(101)

where we have the DOF numbers $m_e = 3n_e + 6n_e^d$, and $m_e^p = d_p \cdot n_e^p$. Then, Eq. (99) can be rewritten as

$$\delta \mathbf{z}^{\mathrm{T}} \mathbf{K} \Delta \mathbf{z} = \delta \mathbf{z}^{\mathrm{T}} 2 \, \mathbf{R}, \tag{102}$$

where we define the *effective* global tangent stiffness matrix

$$\mathbf{K} := \mathbf{A}_{e=1}^{n_{\text{el}}} \mathbb{k}^e, \tag{103}$$



and the effective global residual vector

$$\mathbf{R} := \left\{ \mathbf{F}_{\text{ext}} \left(t_n + \frac{1}{2} \Delta t \right) \right\} - \mathbf{A}_{e=1}^{n_{\text{el}}} \mathbf{f}^e. \tag{104}$$

Here, $\mathbf{z} := [\mathbf{y}^{\mathrm{T}}, \mathbf{r}^{\mathrm{T}}, \mathbf{e}^{\mathrm{T}}]^{\mathrm{T}}$ denotes a column array of the unknown global control coefficients, which assembles the control coefficients for the configuration variable (\mathbf{y}) , the physical stress resultant (\mathbf{r}) , and the physical strain (\mathbf{e}) , where \mathbf{r} and \mathbf{e} have the same number of entries, $n_{\mathrm{dof}}^{\mathrm{p}}$. It should be noted that $\mathbf{k}_{\mathrm{yr}}^{\mathrm{eT}} \neq \mathbf{k}_{\mathrm{ry}}^{\mathrm{e}}$ in Eq. (100) (see Remark 4.1), which makes \mathbf{k}^{e} , and eventually \mathbf{K} unsymmetric. After applying the displacement boundary conditions to Eq. (102), we finally have the reduced system of linear equations at ith iteration in the (n+1)th time step,

$$\mathbf{K}_{\mathbf{r}} \, \Delta \mathbf{z}_{\mathbf{r}} = 2 \, \mathbf{R}_{\mathbf{r}}, \tag{105}$$

where $(\bullet)_r$ denotes the reduced matrix. Using Δz obtained from solving Eq. (105), we update the global control coefficients at the *i*th iteration in the (n+1)th time step, as

$${}^{n+1}\mathbf{y}^{(i)} = {}^{n+1}\mathbf{y}^{(i-1)} + \Delta\mathbf{y}, \ {}^{n+1}\mathbf{y}^{(0)} \equiv {}^{n}\mathbf{y},$$
 (106a)

$$^{n+1}\mathbf{r}^{(i)} = ^{n+1}\mathbf{r}^{(i-1)} + \Delta\mathbf{r}, \ ^{n+1}\mathbf{r}^{(0)} \equiv {}^{n}\mathbf{r},$$
 (106b)

$${}^{n+1}\mathbf{e}^{(i)} = {}^{n+1}\mathbf{e}^{(i-1)} + \Delta\mathbf{e}, \ {}^{n+1}\mathbf{e}^{(0)} \equiv {}^{n}\mathbf{e}.$$
 (106c)

Further, using the element-wise control coefficients $\Delta \mathbf{e}^e$ extracted from the global one $\Delta \mathbf{e}$, we calculate the increment $\Delta \boldsymbol{\alpha}^e$ using Eq. (95a), which is followed by the update,

$$^{n+1}\alpha^{e(i)} = ^{n+1}\alpha^{e(i-1)} + \Delta\alpha^e, \ ^{n+1}\alpha^{e(0)} \equiv {}^{n}\alpha^e, \ \text{for } e \in \{1, 2, \dots, n_{\text{el}}\}.$$
 (107)

After the iteration finishes, we need to update the array of global control coefficients V for the approximated independent velocity field in Eq. (80), using the following update formula from Eq. (56).

$${}^{n+1}\mathbf{V} = \frac{2}{\Delta t} \left({}^{n+1}\mathbf{y} - {}^{n}\mathbf{y} \right) - {}^{n}\mathbf{V}. \tag{108}$$

6 Numerical examples

We present six numerical examples whose objectives are as follows.

- Ex. 1. Rigid rotation of a stress-free rod (quasi-statics). We verify the invariance of the beam strain under a rigid rotation, for any degree p_d by employing the present discretization of the initial director field.
- Ex. 2. Rigid rotation of a bent rod (quasi-statics). We investigate the invariance of the beam strain under a rigid rotation superposed to a deformed configuration.
- Ex. 3. A straight beam under twisting moment (quasi-statics). We verify the present EAS method to correct the torsional stiffness by comparison with analytical and brick element solutions.
- Ex. 4. Twisting of an elastic ring (quasi-statics). We verify the present EAS method to correct stiffness for an initially curved beam undergoing a large rotational motion.



 Ex. 5. Flying beam (dynamics). We verify the superior numerical stability in timestepping due to the conservation of the total energy from using the EMC scheme, compared with the results from using standard schemes.

• Ex. 6. Slotted ring with a rectangular cross-section (quasi-statics and dynamics). We verify the efficiency and accuracy of the present beam solution by comparison with a brick element one. Further, we verify the energy–consistency of the present EMC scheme.

We consider two different ways of evaluating the initial director field:

- [D-cont.] A continuous (exact) representation of the initial director field.
- [*D*-disc.] A discrete (reconstructed) initial director field from the continuous one evaluated at chosen discrete points, see Sect. 5.1.1.

Further, we employ the following different finite element formulations for reference solutions:

- [disp.] The displacement-based isogeometric beam formulation with extensible directors in Choi et al. [7].
- [Δθ] The finite element formulation of geometrically exact beams from Simo and Vu-Quoc [37], with Lagrange polynomial basis functions and uniformly reduced integration.
- [Brick] An isogeometric brick element formulation, where we denote the degree of basis functions by 'deg. = $(p_{\rm L}, p_{\rm W}, p_{\rm H})$ ', and the number of elements by ' $n_{\rm el} = n_{\rm el}^{\rm L} \times n_{\rm el}^{\rm W} \times n_{\rm el}^{\rm H}$ ', where 'L', 'W', and 'H' represent the directions along the length, width, and height, respectively.

The present isogeometric mixed finite element formulation can be combined with the following approaches:

- [mix.loc-ur] $p_d = p 1$, and a uniformly reduced degree $p_p = 1$.
- [mix.loc-sr] $p_d = p 1$, and a selectively reduced degree p_p from Table A.1 in Choi et al. [9].
- [mix.glo] $p_d = p$, $p_p = p 1$, and the inter-element continuity of the physical stress resultant and strain is C^{p_p-1} , see Sect. 5.2.2.

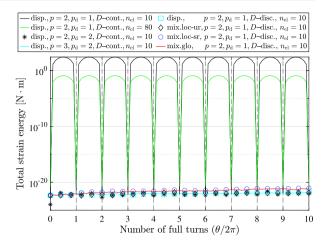
In both [mix.loc-ur] and [mix.loc-sr], the reduction of the degrees p_p and $p_d = p - 1$ compared to [mix.glo] is due to the parasitic strains arising from the higher-order inter-element continuity of the displacement field in IGA. Choi et al. [9] showed that a further (selective) reduction of p_p in [mix.loc-sr] could alleviate the parasitic strains more effectively in general cases than the uniform degree $p_p = 1$ in [mix.loc-ur] does. Although these local approaches allow element-wise condensation, we have observed numerical instability in some cases. In contrast, the approach [mix.glo] shows numerical stability in those cases. Further, due to the global patch-wise approximation, the number of DOFs for the physical stress resultants and strains in [mix.glo] is much lower than those in the local approaches, [mix.loc-ur] and [mix.loc-sr]. In all of those three approaches, we apply element-wise condensation of the enhanced strain field α^h . In the present mixed formulation, we use full Gauss integration for both longitudinal and transverse directions. In the center axis, we use p + 1 quadrature points per element, and in the cross-section, we use $(m_{\text{max}} + 1) \times (m_{\text{max}} + 1)$ quadrature points, where m_{max} denotes the largest value in $\{m_1, m_2, m_3, m_4\}$.

6.1 Rigid rotation of a stress-free rod: objectivity test 1

We consider a quasi-static beam problem where the initial center axis is placed on the XY-plane and represents a quarter circle of radius $R = 100 \,\mathrm{m}$. It has a square cross-section of



Fig. 3 Rigid rotation of a stress-free rod (objectivity test 1): Change of the total strain energy under a rigid rotation. The dashed vertical lines indicate full turns. Note that those markers are only plotted in every four rotational increments for better visibility (Color figure online)



dimension d=1 m. We constrain one end of the beam, at which a rotation of angle 20π [rad] (10 full turns) with respect to the X-axis is prescribed. We prescribe rotational motion by the two directors at the constrained end, but their extensions are not allowed. We incrementally apply the total rotation in 100 uniform increments ($n_{load} = 100$) and investigate the change of total strain energy. A St. Venant-Kirchhoff type isotropic hyperelastic material is considered with the Young's modulus E = 1 MPa and the Poisson's ratio $\nu = 0$. This example for testing the objectivity was proposed in Meier et al. [27, Sect. 11.1]. In Fig. 3, in cases of using the displacement-based beam formulation [disp.], and the continuous initial director field [D-cont.] with $p_d = 1$, it is seen that the strain energy does not vanish (black and green curves), which represents the inability to represent the rigid body rotation (= nonobjectivity). It is noticeable that the spurious strain energy vanishes in every full turn (= path-independence). It is also seen that the error decreases by increasing the number of elements (green curve). If we consider the continuous (exact) initial director field, we need at least quadratic NURBS basis functions in order to represent the rigid body rotation exactly due to the circular initial geometry (Choi et al. [9]). This can be seen in cases of using $p_{\rm d}=2$ and 3 (black star and cyan curve) where the strain energy vanishes up to machine precision. In contrast, employing the discrete initial director field [D-disc.] gives objectivity even with $p_d = 1$ (cyan square, black diamond, blue circle, and red curve).

6.2 Rigid rotation of a bent rod: objectivity test 2

We investigate the invariance of nonvanishing strain under rigid body rotation superposed onto a deformed configuration in quasi-statics. This test was presented in Romero and Armero [34, Sect. 5.1]. We consider a straight beam in its initial (undeformed) configuration with the position vectors of center axis' two end points A and B given by $\varphi_0^A = [0, 0, 2 \,\mathrm{m}]^T$ and $\varphi_0^B = [0, 0, 5 \,\mathrm{m}]^T$, respectively, in a global Cartesian coordinate system. The application of external load is divided into two phases, whose boundary conditions are as follows:

• **First phase.** The displacement of the end point B is prescribed by $\bar{u}_B = [1 \text{ m}, -1 \text{ m}, 0]^T$, with the directors at B free, and the cross-section at point A fixed. That is, we apply the boundary conditions

$$\varphi = \varphi_0, \ d_\alpha = D_\alpha \ (\alpha = 1, 2) \text{ at point A},$$
 (109a)



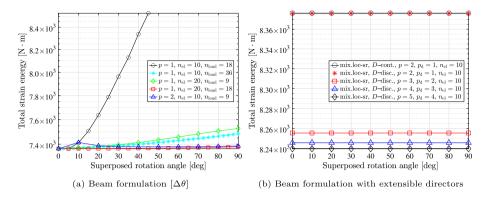


Fig. 4 Rigid rotation of a bent rod (objectivity test 2): Change of the total strain energy under a superposed rigid body rotation in the second phase. (a) Here, n_{load} represents the number of load steps in the second phase. (b) We use $n_{load} = 9$ for all cases (Color figure online)

$$\varphi = \varphi_0 + \bar{\mathbf{u}}_{\mathrm{B}} \text{ at point B},$$
 (109b)

and

$$\tilde{\tilde{m}}_0^{\alpha} = \mathbf{0}$$
, i.e., d_{α} ($\alpha = 1, 2$) are free at point B. (109c)

Second phase. In order to investigate the objectivity, a series of nine rigid body rotations
are quasi-statically superposed onto the deformed configuration after the first phase. We
apply the boundary conditions

$$\varphi = \varphi_0, \ d_\alpha = \bar{\Lambda}_i D_\alpha \ (\alpha = 1, 2) \text{ at point A},$$
 (110)

and

$$\varphi = \bar{\Lambda}_i \left(\varphi_0 + \bar{u}_B \right)$$
, with $d_\alpha \left(\alpha = 1, 2 \right)$ free at point B, (111)

for i = 1, 2, ..., 9, where

$$\bar{\mathbf{\Lambda}}_i = \exp\left[i\frac{\pi}{18}\mathbf{a}\right] \in \mathrm{SO}(3),\tag{112}$$

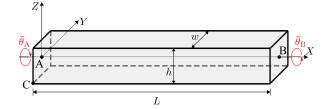
with the axis of rotation

$$a = \frac{1}{\sqrt{3}}[1, 1, 1]^{\mathrm{T}}.$$

We consider a St. Venant-Kirchhoff type isotropic hyperelastic material with Young's modulus $E=21\,\mathrm{MPa}$ and Poisson's ratio v=0.3, and a square cross-section of dimension $d=0.1\,\mathrm{m}$. In this example, since the beam's initial geometry is straight, the two approaches $[D\mathrm{-cont.}]$ and $[D\mathrm{-disc.}]$ yield the same initial director fields. In Fig. 4a, it is seen that the beam formulation $[\Delta\theta]$ suffers from spurious strain energy in superposed rigid rotation, that is, it is nonobjective. It is noticeable that the numerical error decreases by increasing number of load steps, n_{load} in the second phase, where we use a uniform load increment. The reason for this is that the error in the finite element approximation of an incremental rotation should decrease, as the magnitude of rotation decreases. This can be seen, in the much lower



Fig. 5 Straight beam under twisting moment: Initial geometry and boundary conditions. $\bar{\theta}_{A}$ and $\bar{\theta}_{B}$ denote the prescribed rotation angles of the cross-sections at the points A and B with respect to the center axis, respectively



spurious energy resulting from using doubled $n_{\rm load}$ (cyan curve with star), compared with that of using $n_{\rm load}=18$ (black curve with circle). This nonobjectivity can be also alleviated by mesh refinement, a degree elevation (blue curve with triangle) or a larger number of elements (red curve with square). In contrast, the beam formulation with a direct finite element approximation of the director vectors [mix.loc-sr] exhibits no spurious strain energy in the second phase. In Fig. 4b, both methods [D-cont.] and [D-disc.] give the same results and have constant strain energy under superposed rigid rotation (red star and black curve with circle). Figure 4b also verifies the objectivity in higher degrees p=3,4,5 and $p_{\rm d}=p-1$ (red, blue, and black curves).

6.3 Straight beam under twisting moment

We consider a straight beam along the X-axis that is subject to a prescribed twisting angle $\bar{\theta} = 2\pi$ [rad], which is applied quasi-statically, about its center axis at both ends A and B in opposite directions. The beam has an initial length L = 10 m, and a rectangular cross-section of width w and height h, see Fig. 5. For the following cases of w and h, we verify that the present beam formulation can represent the correct torsional stiffness without additional global DOFs.

- Case 1: w = 0.3 m, and h = 0.4 m,
- Case 2: w = 1/3 m, and h = 1 m.

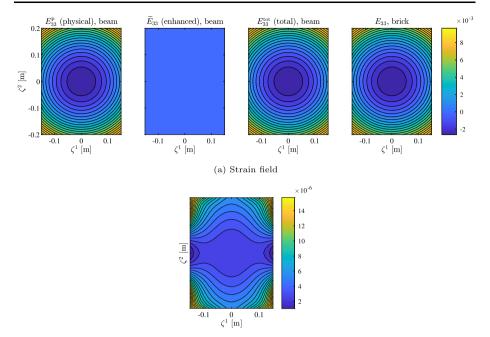
A St. Venant-Kirchhoff type isotropic hyperelastic material with Young's modulus $E = 210 \,\text{GPa}$ and Poisson's ratio $\nu = 0.3$ is considered. For comparison, we consider two different boundary conditions:

- [free-free] Prescribe the rotation angles $\bar{\theta}_A = -\bar{\theta}/2$ and $\bar{\theta}_B = \bar{\theta}/2$. Note that, here, out-of-plane cross-sectional warping is allowed at both ends, A and B. For the brick solution, the *X*-displacement is constrained at C in order to prevent a rigid-body translation.
- [fixed-free] Fix the cross-section at A ($\bar{\theta}_A = 0$), and prescribe the rotation angle $\bar{\theta}_B = \bar{\theta}$ at B. Note that we still allow for out-of-plane cross-sectional warping at the loaded end, B.

It should be noted that, in both cases of the boundary conditions, in-plane cross-sectional strains are not allowed at the ends. For beams, we use the [free-free] only, where the cross-sectional warping due to the enhanced strains is allowed at both ends. An application of boundary conditions to constrain the enhanced strain field remains future work. Figures 6–12 compare the cross-sectional strains at the middle of the center axis (s = L/2) between the present beam and brick solutions for case 1 (w = 0.3 m and h = 0.4 m). Here, for the beam solution, we use [mix.loc-sr] with p = 3, $m_1 = m_2 = m_3 = m_4 = 4$, and $n_{\rm el} = 40$. For the brick solution, we use deg. = (3, 3, 3) and $n_{\rm el} = 80 \times 10 \times 10$, and the [free-free] condition.

Observation 1 In Fig. 6, it is seen that the enhanced strain \widetilde{E}_{33} vanishes. This implies that the quadratic polynomial basis in $\mathbf{A}(\zeta^1, \zeta^2)$ in the third row of Eq. (11) is sufficient to represent the axial normal strain. It has also been discussed in Wackerfuß and Gruttmann





(b) Difference between the beam and brick solutions, $E_{33}^{\rm tot} - E_{33}$

Fig. 6 Straight beam under twisting moment for w = 0.3 m and h = 0.4 m: Comparison of the axial normal strain, E_{33} at s = L/2. For the beam solution, we enhance \widetilde{E}_{33} by the basis \mathbf{w}_3 with the maximum degree $m_3 = 4$ and the minimum degree $m_3 + 1 = 3$, which is orthogonal to polynomials up to degree $m_3 = 2$. The maximum value of $|\widetilde{E}_{33}|$ is 4.0×10^{-17} [-]. There is excellent agreement between the total strains for beam and brick (Color figure online)

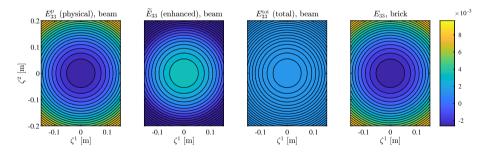
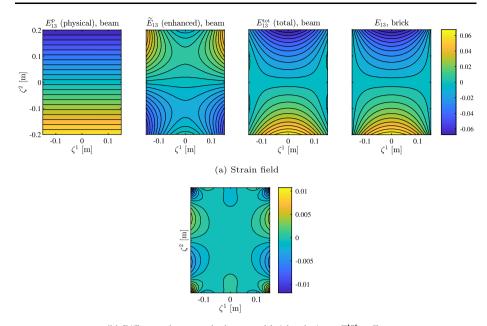


Fig. 7 Straight beam under twisting moment for w = 0.3 m and h = 0.4 m: Comparison of the axial normal strain, E_{33} at s = L/2. Brick solution is the same as in Fig. 6. For the beam solution, we enhance \widetilde{E}_{33} by the basis $\mathbf{w}_4 = \mathbf{w}_3$ with the maximum degree $m_3 = 4$ and the minimum degree $m_3 + 1 = 2$, which is orthogonal to polynomials up to degree $m_3 = 1$. The resulting enhanced strain \widetilde{E}_{33} leads to erroneous total strain (E_{33}^{tot}) (Color figure online)

[42] that the enhancement of E_{33} is not needed for a correct representation of the threedimensional strain state. Accordingly, they presented a so-called 'E2-model' which does not enrich the axial normal strain. This significantly reduces the number of internal DOFs for the enhanced strain field, which makes the formulation more efficient.





(b) Difference between the beam and brick solutions, $E_{13}^{\mathrm{tot}}-E_{13}$

Fig. 8 Straight beam under twisting moment for $w=0.3\,\mathrm{m}$ and $h=0.4\,\mathrm{m}$: Comparison of the transverse shear strain, E_{13} at s=L/2. For the beam solution, we enhance \widetilde{E}_{13} by the derivative $\mathbf{w}_{4,1}$, where the maximum and minimum degrees of \mathbf{w}_4 are $m_4=4$ and $\bar{m}_4+1=2$, respectively. There is excellent agreement between the total strains for beam and brick (Color figure online)

In Figs. 8 and 9, the physical transverse shear strains E_{13}^p and E_{23}^p are limited to linear distributions since it has polynomial basis up to degree 1. It is noticeable that, by employing the enhanced strains, the total strains show excellent agreement with the brick solution, in both cases, E_{13}^{tot} and E_{23}^{tot} . In Fig. 7, we show the beam solution using $\bar{m}_3 = 1$ instead of $\bar{m}_3 = 2$, with all other parameters the same as in the result of Fig. 6. This aims at showing that losing the orthogonality between E_{33}^p and \tilde{E}_{33} can lead to significant error. Selecting the lower degree $\bar{m}_3 = 1$ means that the basis \mathbf{w}_3 may have quadratic terms in spite of the existing quadratic terms in E_{33}^p . This eventually pollutes the physical strain field, which can be seen by the deviation of the resulting total strain from the brick solution.

In Figs. 10 and 11, the physical in-plane cross-sectional strains E_{11}^p and E_{22}^p , respectively, have negative constant values, which represents a uniform transversal contraction of the cross-section. Although the enhanced strains improve the degree of approximation, they are limited to a quadratic distribution due to the coupling with the axial normal strain E_{33}^p by the nonzero Poisson's ratio. Further, in the case of the in-plane shear strain, the enhanced strain \widetilde{E}_{12} turns out not to be activated, as Fig. 12 shows. Further investigation on the selection of the basis \mathbf{w}_1 and \mathbf{w}_2 for improving the enhancement of in-plane cross-sectional strain remains future work. In Fig. 13, we compare the applied moment for a given prescribed rotation angle $\bar{\theta}$ between the present beam element solution, the reference brick element solution, and the analytical solution in Eq. (C.1). In the first case (w = 0.3 m and h = 0.4 m), the beam solutions show excellent agreement with both analytic and brick element solutions. In Fig. 13a, the beam solution with higher degree $\bar{m}_4 = 2$ (cyan diamonds) shows overly stiff behavior due to the missing quadratic terms in the enriched basis \mathbf{w}_4 . Over-



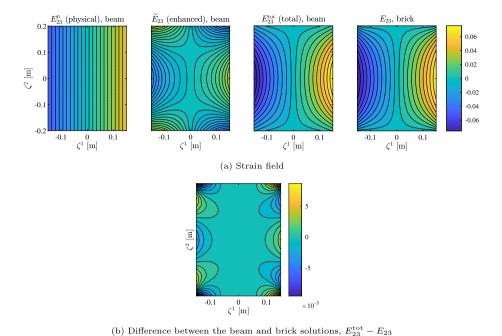


Fig. 9 Straight beam under twisting moment for w = 0.3 m and h = 0.4 m: Comparison of the transverse shear strain component, E_{23} at s = L/2. For the beam solution, we enhance \widetilde{E}_{23} by the derivative $\mathbf{w}_{4,2}$, where the maximum and minimum degrees of \mathbf{w}_4 are $m_4 = 4$ and $\overline{m}_4 + 1 = 2$. There is excellent agreement between the total strains for beam and brick (Color figure online)

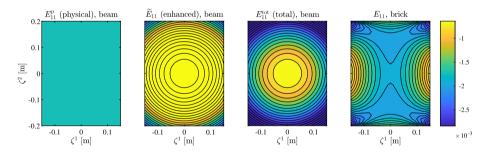


Fig. 10 Straight beam under twisting moment for w = 0.3 m and h = 0.4 m: Comparison of the transverse normal strain, E_{11} at s = L/2. For the beam solution, we enhance \widetilde{E}_{11} by \mathbf{w}_1 with the maximum and minimum degrees $m_1 = 4$ and $\bar{m}_1 + 1 = 1$, respectively (Color figure online)

all, the mixed finite element formulation enables much larger load increments compared to the displacement-based brick element formulation. In the second case (w = 1/3 m and h = 1 m), a nonlinear effect of increasing torsional stiffness along the prescribed rotation is clearly seen in both beam and brick solutions. In the latter range of the prescribed rotation, it is seen that the brick solution with the [fixed–free] boundary condition exhibits more rapidly increasing torsional stiffness compared to the solution using the [free-free] condition due to the constrained out-of-plane warping at s = 0. It is remarkable that the beam and brick solutions under the [free–free] condition show excellent agreement. In Table 1, we compare



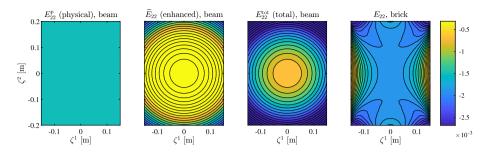


Fig. 11 Straight beam under twisting moment for $w=0.3\,\mathrm{m}$ and $h=0.4\,\mathrm{m}$: Comparison of the transverse normal strain, E_{22} at s=L/2. For the beam solution, we enhance \widetilde{E}_{22} by \mathbf{w}_1 with the maximum and minimum degrees $m_1=4$ and $\bar{m}_1+1=1$, respectively (Color figure online)

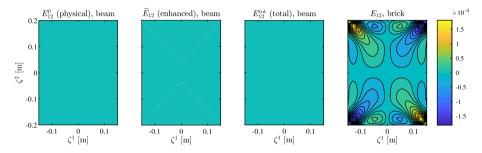


Fig. 12 Straight beam under twisting moment for w = 0.3 m and h = 0.4 m: Comparison of the in-plane shear cross-sectional strain component, E_{12} at s = L/2. For the beam solution, we enhance \widetilde{E}_{12} by \mathbf{w}_2 with the maximum and minimum degrees $m_2 = 4$ and $\overline{m}_2 + 1 = 1$, respectively (Color figure online)

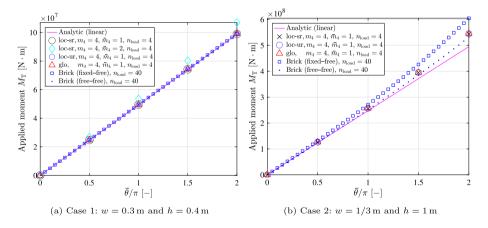


Fig. 13 Straight beam under twisting moment: Change of the applied moment along the prescribed rotation $\bar{\theta}$. In [mix.loc-ur] and [mix.loc-sr], we use p=3, $p_{\rm d}=2$. For [mix.glo], we use $p=p_{\rm d}=3$. All the beam solutions are from using $n_{\rm el}=40$, and $m_1=m_2=2$, without enriching the axial normal strain, E_{33} . For the brick solutions, we use deg. =(3,3,3), and $n_{\rm el}=80\times10\times10$ (Color figure online)



Table 1 Straight beam under twisting moment: Comparison of total number of DOFs for cases 1 and 2. For our beam formulation, we have DOFs for the configuration (y), physical stress resultant (r_p) , physical strain (ε_p) , and enhanced strain (α) . The gray cells represent the numbers of internal DOFs

	Beam [mix.loc-sr]	Beam [mix.loc-ur]	Beam [mix.glo]	Brick
Degree of basis	$p = 3, p_{\rm d} = 2$	$p = 3, p_{\rm d} = 2$	$p = p_{\rm d} = 3$	deg. = (3, 3, 3)
#elements	40	40	40	$80 \times 10 \times 10$
#DOFs(y)	381	381	387	42,081
#DOFs (r_p)	814	1200	630	_
#DOFs $(\boldsymbol{\varepsilon}_{\mathrm{p}})$	814	1200	630	_
#DOFs (α)	1080	1080	1080	_
#global DOFs	381	381	1647	42,081
#internal DOFs	2708	3480	1080	_

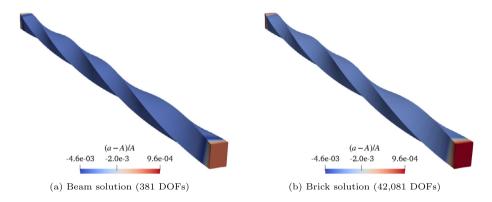


Fig. 14 Straight beam under twisting moment for $w=0.3\,\mathrm{m}$ and $h=0.4\,\mathrm{m}$ (case 1): Final deformed configuration at $\bar{\theta}_{\mathrm{A}}=-\pi$ and $\bar{\theta}_{\mathrm{B}}=\pi$. The colors represent the relative change of the cross-sectional area, where a and A denote the deformed and initial cross-sectional areas, respectively. For the beam solution, we use [mix.loc-sr] with p=3, $n_{\mathrm{el}}=40$, and $m_1=m_2=2$, $m_4=4$, where E_{33} has not been enriched. For the brick solution, we use deg. =(3,3,3), and $n_{\mathrm{el}}=80\times10\times10$. There is excellent agreement between beam and brick solutions apart from the two end faces (Color figure online)

the total number of global and internal DOFs between beam and brick solutions. It is noticeable that the beam solutions use much smaller number of DOFs than the brick solution does, since the DOFs of the enhanced strain (α) can be element-wisely condensed out. In the beam formulation [mix.glo], the DOFs of the physical stress resultants (\mathbf{r}) and physical strain (\mathbf{e}) are considered in the global system of equations, whose number is much smaller than those of the local approaches for the same number of elements, due to the higher-order continuity of B-spline basis functions, see Remark 5.1. Figures 14 and 15 compare the deformed configurations between the beam and brick solutions for the two different dimensions of the initial cross-section. It is noticeable that the same level of decrease in the cross-sectional area is observed in the beam and brick solutions. In the brick solution, it is seen that the cross-sectional area slightly increases at both ends due to the out-of-plane warping. This is not explicitly shown in the beam solution since we calculate the cross-sectional area using only two directors d_1 and d_2 , which can only represent planar cross-section deformations.



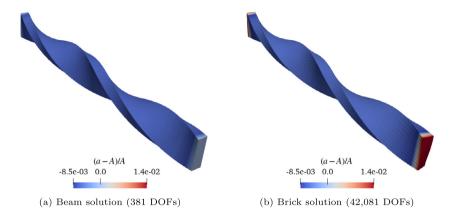
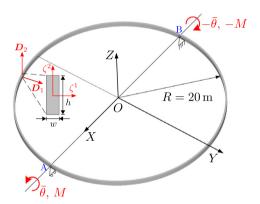


Fig. 15 Straight beam under twisting moment for w=1/3 m and h=1 m (case 2): Final deformed configuration at $\bar{\theta}_A=-\pi$ and $\bar{\theta}_B=\pi$. The colors represent the relative change of the cross-sectional area, where a and A denote the deformed and initial cross-sectional areas, respectively. For the beam solution, we use [mix.loc-sr] with p=3, $n_{\rm el}=40$, and $m_1=m_2=2$, $m_4=4$, where E_{33} has not been enriched. For the brick solution, we use deg. =(3,3,3), and $n_{\rm el}=80\times10\times10$. There is excellent agreement between beam and brick solutions apart from the two end faces (Color figure online)

Fig. 16 Twisting of an elastic ring: Undeformed configuration and boundary conditions



6.4 Twisting of an elastic ring

We consider a closed circular ring twisted by a prescribed rotation. Figure 16 shows the undeformed configuration and boundary conditions. The initial ring has a radius $R=20\,\mathrm{m}$, and a rectangular cross-section having width w and height h. We prescribe a rotation of the cross-sections at the material points of initial coordinates $[R,0,0]^\mathrm{T}$ (point A) and $[-R,0,0]^\mathrm{T}$ (point B) by the angle $\bar{\theta}$ with respect to the X-axis in opposite directions. We apply the prescribed rotation angle $\bar{\theta}=2\pi$ [rad] in total, such that the configuration finally goes back to the initial (undeformed) one if no impenetrability condition is imposed, see Fig. 17 for the deformed configurations at different values of $\bar{\theta}$. For $\bar{\theta}=\pi$, the ring is folded into a smaller circular ring. Without axial strain, the radius of the deformed ring is R/3, which was first theoretically investigated by Goto et al. [17]. With axial strain, the deformed configuration is still a folded ring but has a different radius r^* such that $2\pi \cdot r^* = \ell/3$, where ℓ denotes the current total length. In the present paper, this example aims at the following:



Verify the capability of the present beam formulation to follow the correct equilibrium
path of the deformation by correcting the torsional stiffness through the enrichment of
higher-order cross-sectional strains.

- Investigate the effect of axial strain coupled with bending strain in the present beam formulation. The coupled axial strain changes the total length ℓ , and subsequently the radius of the deformed ring becomes $r^* = \ell/(6\pi)$ at $\bar{\theta} = \pi$. This additional strain should quadratically decrease with increasing slenderness ratio (Choi et al. [7, Sect. 6.2]). In order to show this, we consider the following two different cases with cross-sectional dimensions:
 - Case 1: w = 1/3 m, and h = 1 m,
 - Case 2: $w = 1/30 \,\mathrm{m}$, and $h = 1/10 \,\mathrm{m}$.

Note that, here, we always have a ratio h/w=3, which is known to produce a deformed configuration of a smaller ring by the prescribe rotation $\bar{\theta}=\pi$ (Goto et al. [17]).

• Verify the path-independence of the present beam formulation.

A St. Venant-Kirchhoff type isotropic hyperelastic material is considered, with Young's modulus $E = 21 \,\text{MPa}$, and Poisson's ratio $\nu = 0.3$. In order to account for the decreased torsional stiffness due to the cross-sectional warping, we employ the following approaches:

- In the beam formulation $[\Delta\theta]$, we correct the torsional stiffness by replacing the polar moment of inertia with $I_p^* = K$, see Appendix C.1. Note that K is much smaller than the polar moment of inertia, $I_p = I^{11} + I^{22}$. For example, in case of w = 1/3 m and h = 1 m, we have $I^{11} \approx 3.086 \times 10^{-3}$ [m⁴] and $I^{22} \approx 2.778 \times 10^{-2}$ [m⁴], which gives the polar moment of inertia much larger than the corrected value $K = 9.753 \times 10^{-3}$ [m⁴]. This approach refers to Romero [33], Meier et al. [26], and references therein.
- For the present beam formulation, instead of directly adjusting the cross-sectional parameters, we enrich the polynomial basis functions up to degree m_4 for the transverse shear strains E_{13} and E_{23} , and $m_1 (= m_2)$ for the in-plane strains E_{11} , E_{22} , and E_{12} . Note that the axial normal strain component E_{33} is not enriched, see Observation 1.

In Fig. 18, we compare the equilibrium paths from using the formulation $[\Delta \theta]$ and the present one [mix.glo]. The black solid curves represent the result from $[\Delta \theta]$ using the corrected torsional stiffness with $I_p^* = K$. Note that, without this correction (i.e., with $I_p^* = I_p$), the torsional stiffness is significantly overestimated, see the much larger initial slope of the black solid curves compared to the others. This eventually leads to the divergence in the equilibrium iteration under the prescribed rotation. The black dashed curves represent the equilibrium paths with the corrected torsional stiffness. In the results from the present beam formulation [mix.glo], we first enrich the transverse normal strain components E_{11} and E_{22} by the maximum degree of basis $m_1 = 2$ in order to obtain the correct bending stiffness due to the nonzero Poisson's ratio. Then, as we increase the maximum degree of enriched basis for the transverse shear strain m_4 , the equilibrium path approaches the result from using $m_4 = 8$ (red curve). We have an analytical solution of the deformed center axis such that it becomes a folded circular ring of radius $r_{\rm ref} = R/3$ at $\bar{\theta} = \pi$ (Goto et al. [17]). However, it turns out that additional axial compression can be induced by bending (Choi et al. [7, Sect. 6.2.1]). In order to illustrate this, in Fig. 19, we compare the approximated total length of the center axis at $\bar{\theta} = \pi$ between the results from the formulations $[\Delta \theta]$ and [mix.glo]. It is noticeable that, in the present beam solution (red curve), the beam is axially compressed, in contrast to the result from using $[\Delta \theta]$. Further, it is seen that the amount of compressive strain decreases by 10⁻² times, with increasing the slenderness ratio by 10 times from Case 1 to Case 2. This means that, as expected, the decrease of the coupled axial strain is in



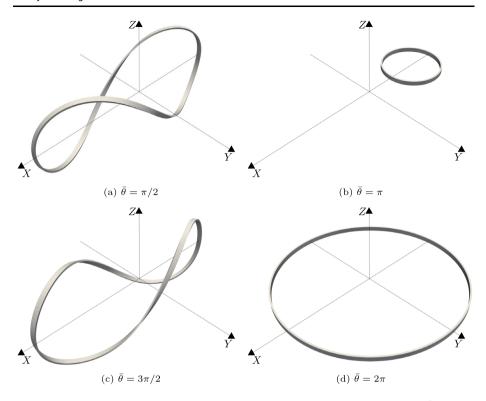


Fig. 17 Twisting of an elastic ring (case 1): Deformed configurations for four different values of $\bar{\theta}$

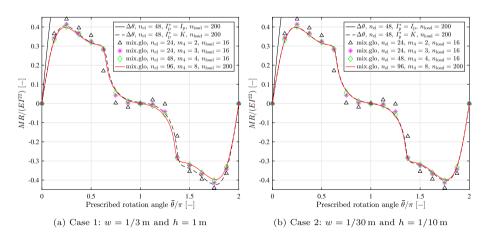


Fig. 18 Twisting of an elastic ring: Applied moment at point A due to the prescribed rotation $\bar{\theta}$. In both cases, we use p=1 and p=4 for the formulations $[\Delta\theta]$ and [mix.glo], respectively. Further, for [mix.glo], $m_1=m_2=2$, and E_{33} has not been enriched (Color figure online)

quadratic order with respect to the slenderness ratio. Based on this, we introduce a corrected reference solution for the deformed center axis at $\bar{\theta} = \pi$, such that it is still a folded circular



ring but with radius $r_{\text{ref}}^* = \ell^h/(6\pi)$, where the (approximated) total length ℓ^h is given by an overkill solution from using the present beam formulation, [mix.glo] with p=4. From this, we can define a relative L^2 -difference of the center axis displacement field as

$$e_{\varphi} := \sqrt{\frac{\int_{0}^{L} (r^{h} - r_{\text{ref}}^{*})^{2} ds}{\int_{0}^{L} r_{\text{ref}}^{*2} ds}}, \text{ with } r^{h} := \| \boldsymbol{\varphi}^{h} - \boldsymbol{c}_{\text{ref}}^{*} \|,$$
(113)

where the reference center of the circle is given by $c_{\text{ref}}^* := [-R + r_{\text{ref}}^*, 0, 0]^{\text{T}}$. Figure 20 shows the convergence of the relative differences from two different beam formulations, $[\Delta \theta]$ with the degree of basis p = 1, and the present one [mix.glo] with a degree of basis p = 2, 3, 4. It is seen that in both cases 1 and 2, the present beam solution from using [mix.glo] converges toward the reference one (a folded circular ring with radius r_{ref}^*), with increasing number of elements. In contrast, the beam formulation $[\Delta \theta]$ yields a different solution, which deviates from the reference one. This deviation is due to the inability to represent the axial strain coupled with bending, and it should decrease quadratically with increasing slenderness ratio. It is also noticeable that the results from [mix.glo] exhibit optimal convergence rates (p+1) or sometimes even faster, which can be clearly seen in both cases. In the results from using the beam formulation [mix.loc-sr], deteriorated convergence is observed due to numerical instability, except for the results from using p = 3. In Table C.1, we compared the total number of load steps and equilibrium iterations between [mix.glo] and [mix.loc-sr]. The present formulation [mix.glo] shows stability in all cases. In Goto et al. [17], it is shown that the deformed smaller ring at $\bar{\theta} = \pi$ can stay without any external load. Further, due to the symmetry, the configuration should go back to the initial (undeformed) one at $\bar{\theta} = 2\pi$ by either loading or unloading from the smaller ring at $\bar{\theta} = \pi$. That is, if a formulation is pathindependent, the applied moment M should vanish at both $\bar{\theta} = \pi$ and 2π . Figures 21a and 21b plot the convergence of the moments M at $\bar{\theta} = \pi$ and 2π , respectively, with increasing number of elements. It is seen that, for the formulation $[\Delta\theta]$, the moments do not vanish, which implies that the formulation suffers from path dependence. Note that this error decreases in quadratic order, which is consistent with the convergence rate of the displacement field in Fig. 20. In contrast, from using the present beam formulation, the moments vanish up to machine precision for any number of elements.

The next two examples deal with nonlinear transient dynamics to verify the present timestepping scheme.

6.5 Dynamics of a flying beam

We consider an initially straight beam of length $L=3\,\mathrm{m}$ which has a square cross-section of dimension $d=0.3\,\mathrm{m}$, see Fig. 22. A St. Venant-Kirchhoff type hyperelastic material with Young's modulus $E=21\,\mathrm{MPa}$ and Poisson's ratio v=0.3 is considered. The initial mass density is $\rho_0=1\,\mathrm{kg/m^3}$. We apply an initial velocity condition in X- and Y- directions along the initial center axis as

$$v_{\varphi 0}^{1} = -\frac{\bar{v}_{\varphi}^{1}}{L}(X - L), \qquad (114a)$$

$$v_{\varphi 0}^{2} = \bar{v}_{\varphi}^{2} \left\{ \frac{X}{L} \left(\frac{X}{L} - 1 \right) + \frac{1}{6} \right\} + \frac{\bar{v}_{\varphi}^{1}}{L} \left(X - \frac{L}{2} \right), \tag{114b}$$

respectively, with $\bar{v}_{\varphi}^1 = 1$ [m/s] and $\bar{v}_{\varphi}^2 = 150$ [m/s], see also Fig. 23. The directors are assumed to be initially stationary, which means that all the material points in each cross-section



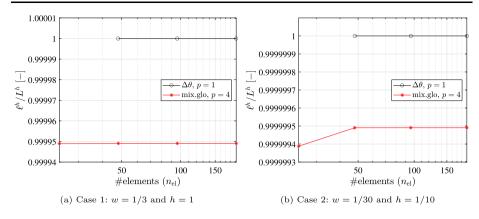


Fig. 19 Twisting of an elastic ring: Change of the ratio between the approximated total lengths of the center axis in the initial (L^h) and deformed configuration at $\bar{\theta} = \pi$ (ℓ^h) , for two different cases of (w, h). For [mix.glo], we use $m_1 = m_2 = 2$, $m_4 = 4$, and E_{33} has not been enriched

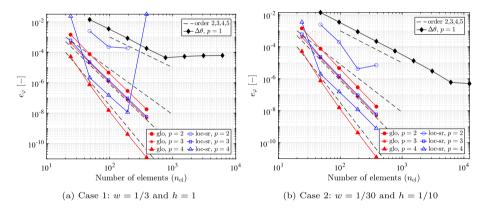


Fig. 20 Twisting of an elastic ring: Relative difference of the deformed radius from the reference solution r_{ref}^* at $\bar{\theta} = \pi$ for two different cases of (w, h). In both beam formulations [mix.glo] and [mix.loc-sr], we use $m_1 = m_2 = 2$, $m_4 = 4$, and E_{33} has not been enriched

have the same initial velocity. The initial velocity component in Eq. (114a) leads to a translation in X-direction. In Eq. (114b), the quadratic Y- directional velocity by the first term leads to the bending and transversal shear strains in the X-Y plane, as well as the transverse normal strain due to the nonzero Poisson's ratio. The second term in Eq. (114b) is introduced to have a rotational motion in the X-Y plane. Therefore, the total linear and angular momentum vectors may have nonzero components in X- and Z- directions, respectively. Figure 24 shows the deformed configurations at selected time instants. In Fig. 25, we investigate the conservation of energy and momentum. It is seen that the standard trapezoidal rule conserves the linear momentum but neither the angular momentum nor the total energy. For the given time step size $\Delta t = 0.1 \, \text{s}$, the total energy blows up. The total angular momentum initially has nonzero value due to the initial velocity condition and is maintained by both the standard mid-point rule and the present EMC schemes. Initially, the total energy only has a kinetic part, and it is exactly conserved by the present EMC scheme. This implies the symmetry between the strain energy (U) and the kinetic energy (K). In contrast, when



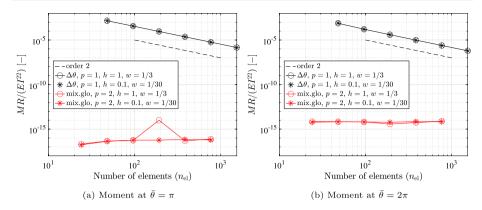


Fig. 21 Twisting of an elastic ring: path-independence test. Comparison of the applied moments at $\bar{\theta} = \pi$ and $\bar{\theta} = 2\pi$. In [mix.glo], we use $m_1 = m_2 = 2$, $m_4 = 4$, and E_{33} has not been enriched

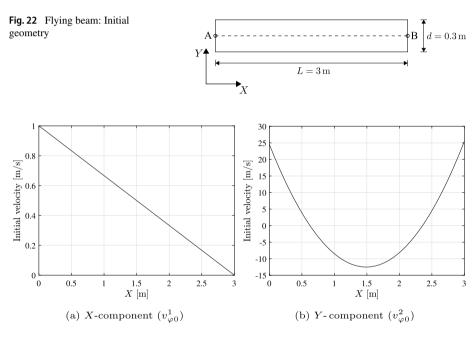


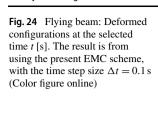
Fig. 23 Flying beam: Given initial velocity of the center axis, $v_{\varphi 0} = v_{\varphi 0}^i e_i$, where $v_{\varphi 0}^3 = 0$

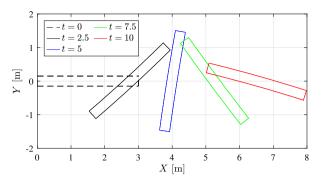
using the mid-point rule, the total energy is not conserved but blows up. Here, for all the beam solutions, we use [mix.glo] with p = 2, $n_{el} = 10$, and $m_1 = m_2 = 2$, $m_4 = 4$, with no enrichment for E_{33} .

6.6 Slotted ring with a rectangular cross-section

An initially circular ring of radius R = 1.3 m with a slot of angle 1° is considered. It has a rectangular cross-section of height h = 0.4 m and width w = 0.3 m, see Fig. 26. One end is fixed, where no cross-sectional deformation is allowed, and a distributed force is applied in







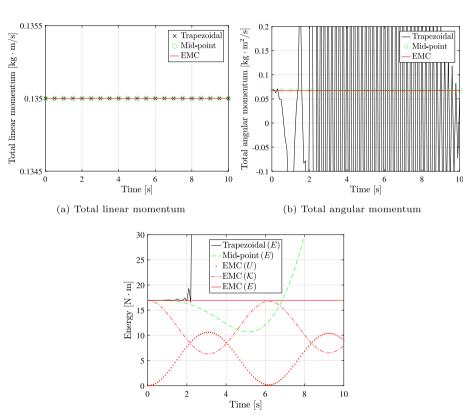


Fig. 25 Flying beam: Time history of the energy and momentum. In (a) and (b), the markers are plotted at every five-time steps for clearer visualization. In all the results, we use the same time step size, $\Delta t = 0.1$ s. In (c), the total energy E is obtained from the summation of the kinetic energy (K) and the strain energy (U)

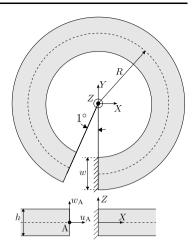
(c) Total energy

Z-direction on the other end (cross-section at point A). In this example, we investigate two different types of loading:

• Case 1 (elastostatics): A quasi-statically applied surface traction (force per unit undeformed area), $T = F_0 e_3$, where $F_0 = 5 \times 10^4 \,\text{N/m}^2$,



Fig. 26 Slotted ring: Undeformed configuration and boundary conditions. Dashed line represents the initial center axis



• Case 2 (elastodynamics): A (piece-wise linearly) time-dependent surface traction, $T(t) = F(t) e_3$, where F(t) (force per unit undeformed area) is given by

$$F(t) = \begin{cases} F_0 t / T_0 & \text{if } 0 \le t \le \frac{T}{2}, \\ F_0 (T - t) / T_0 & \text{if } \frac{T}{2} < t \le T. \end{cases}$$
 (115)

Here, $F_0 = 5 \times 10^4 \,\text{N/m}^2$, $T_0 = 1 \,\text{s}$ denotes a unit time, and T denotes the terminal time.

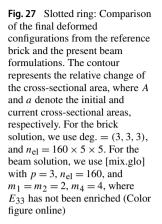
6.6.1 Case 1: comparison with the brick element solution

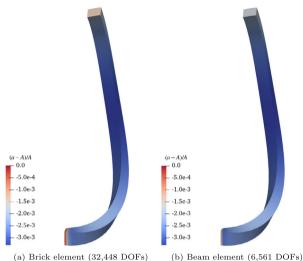
We consider a Neo-Hookean type isotropic hyperelastic material with Young's modulus $E = 1.12 \times 10^7$ Pa, and Poisson's ratio $\nu = 0.4$. For the given load, the ring undergoes a large deformation with torsion-induced cross-sectional warping. Here, we verify the corrected stiffness by using the present EAS method. In Fig. 27, we compare the deformed configurations between the present beam and the brick element solutions, where it is seen that the cross-section contracts due to the positive Poisson's ratio, except for the fixed end. It is seen that the change of cross-sectional area can be properly represented by only two directors in the present beam formulation. Figures 28a and 28b show the X- and Z-displacements of the ring at point A, which are denoted by u_A and w_A , respectively. In the beam formulation, we compare different degrees of the polynomial basis $m_4 = 2, 3, 4$ for the transverse shear strains, E_{13} and E_{23} . Note that we also use $m_1 = m_2 = 2$, and the axial normal strain E₃₃ has not been enriched. It is seen that without the enrichment of the transverse shear strain, the beam solution (blue curve) deviates significantly from the brick solution (black curve). It is noticeable that with increasing degree m_4 , the solution approaches the brick solution (magenta and red marks). Further, a much larger load step is allowed in the present beam formulation, compared to the displacement-based brick formulation, which can be seen by the results of the beam element using six uniform load increments (cyan mark).

6.6.2 Case 2: verification of the consistency in the EMC scheme

In the second case, we investigate a transient dynamic response of the slotted ring, under the time-dependent load given by Eq. (115). The total simulation time is T = 20 s. Figures 29







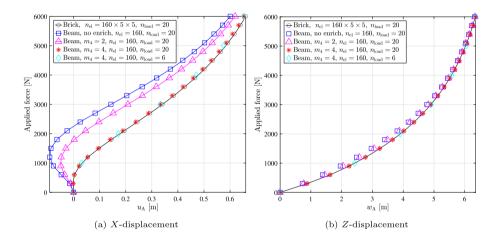


Fig. 28 Slotted ring: Comparison of the *X* and *Z*-directional displacements at the tip (point A) with increasing applied force. In all the beam solutions, we use [mix.glo] with p = 3, $m_1 = m_2 = 2$, and E_{33} has not been enriched (Color figure online)

and 30 investigate whether a time-stepping scheme satisfies the following identity

$$^{n+1}E = ^{n+1}E^*, n = 0, 1, \dots,$$
 (116)

which is recalled from Eq. (B.22). For constitutive laws, we consider two different models, St. Venant-Kirchhoff (linear) and Neo-Hookean (nonlinear) types, with the same values of the parameters E and ν as in case 1. The initial mass density is $\rho_0 = 1 \, \text{kg/m}^3$. From using the standard mid-point rule, it is seen that the total energy ^{n+1}E deviates from $^{n+1}E^*$ for both linear and nonlinear constitutive laws. In contrast, the EMC scheme satisfies the identity exactly (up to machine precision) for the linear constitutive law, see Fig. 29b. For the nonlinear constitutive law, the EMC scheme has also lost consistency, but the relative difference is much smaller than that from the mid-point rule, see Fig. 30b.



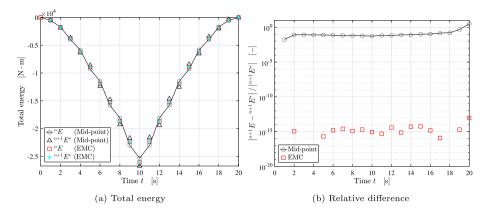


Fig. 29 Slotted ring: Verification of the energy—consistency for a linear constitutive law (St. Venant-Kirchhoff material). Missing data points (rectangles) in (b) are due to an exact zero difference in the numerator

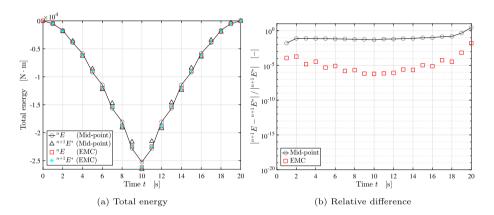


Fig. 30 Slotted ring: Verification of the energy-consistency for a nonlinear constitutive law (Neo-Hookean material)

7 Conclusions

We present a stable and efficient implicit isogeometric finite element method for geometrically and materially nonlinear beams in transient dynamics. The superior numerical stability of the developed method comes from the following aspects:

- Objectivity and path-independence of the isogeometric finite element formulation that directly approximates the director field, which turns out to hold for any degree p_d , which is verified in several numerical examples. This formulation is based on the approximation of the initial director field, which is, in the present paper, denoted by [D-disc.].
- An energy-momentum consistent (EMC) time-stepping scheme, which gives exact energy conservation property for linear constitutive laws.
- Employing the independent DOFs for the stress resultants in the mixed finite element
 formulation based on the Hu-Washizu variational principle alleviates the overestimation
 of the internal force, which improves the stability of the equilibrium iteration for larger
 time (load) steps in the Newton-Raphson method.



• In the isogeometric approach, we have observed numerical instabilities of the local approach [mix.loc-sr] in some cases. The present global approach [mix.glo] shows numerical stability in all those cases. Further, compared to the local approach, the present global approach requires much fewer DOFs for the physical stress resultants and strains due to the higher-order continuity of the B-spline basis functions.

Further, the developed method provides superior computational efficiency, compared to brick element formulations, due to the following aspects:

- The present 9-DOF formulation with extensible (unconstrained) directors allows for an
 efficient representation of a constant in-plane cross-sectional strain, e.g., the change of
 cross-sectional area.
- The higher-order cross-sectional strains like torsion-induced warping have been further
 enriched by an EAS method. Due to the allowed inter-element discontinuity in the enhanced strain field, one can eliminate those additional DOFs from the global system of
 linear equations.

One can extend the present work in the following aspects:

- In the inertia terms, we have considered the cross-sectional deformation only up to the constant strains, represented by two extensible directors.
- We have limited our scope of investigation to convex-shaped cross-sections like a rectangular one.
- The present EAS method corrects stiffness very effectively; however, it is seen that the
 enrichment of in-plane cross-sectional strains is limited.

Appendix A: Beam formulation

A.1 Objectivity of the geometric strain in continuous form

The *invariance* of the beam strain components under a superposed rigid body motion can be analytically verified. We superpose an arbitrary constant rigid body rotation $\Lambda_{\theta} \in SO(3)$ and translation $c_{\varphi} \in \mathbb{R}^3$ to the current configuration y as (Crisfield and Jelenić [11])

$$\boldsymbol{\varphi}^* = \boldsymbol{\Lambda}_{\theta} \left(\boldsymbol{\varphi} + \boldsymbol{c}_{\varphi} \right), \tag{A.1a}$$

$$\boldsymbol{d}_{\alpha}^* = \boldsymbol{\Lambda}_{\theta} \, \boldsymbol{d}_{\alpha}, \ \alpha \in \{1, 2\}, \tag{A.1b}$$

where SO(3) denotes a three-dimensional rotation group, defined as

$$SO(3) := \left\{ \Lambda \in \mathbb{R}^{3 \times 3} \middle| \Lambda^{T} \Lambda = 1, \text{ and det } \Lambda = 1 \right\}.$$
 (A.2)

By inserting Eqs. (A.1a) and (A.1b) into Eqs. (12a)–(12f), and using $\Lambda_{\theta,s} = 0$, $c_{\varphi,s} = 0$, and $\Lambda_{\theta}^{T} \Lambda_{\theta} = 1$, we have

$$\boldsymbol{\varepsilon}(\mathbf{y}^*) = \boldsymbol{\varepsilon}(\mathbf{y}). \tag{A.3}$$

A.2 Objectivity of the geometric strain in discrete form

We investigate the invariance of the strain measures approximated by Eqs. (76) and (77) under superposed rigid body translation and rotation. We superpose an arbitrary constant



rigid body rotation $\Lambda_{\theta} \in SO(3)$ and translation $c_{\varphi I} \in \mathbb{R}^3$ to the current control coefficients φ_I and $d_{\alpha I}$, as

$$\boldsymbol{\varphi}^{h*}(\xi) = \sum_{I=1}^{n_{\text{cp}}} N_I^p(\xi) \boldsymbol{\Lambda}_{\theta} \left(\boldsymbol{\varphi}_I + \boldsymbol{c}_{\varphi} \right)$$

$$= \boldsymbol{\Lambda}_{\theta} \left(\sum_{I=1}^{n_{\text{cp}}} N_I^p(\xi) \boldsymbol{\varphi}_I + \boldsymbol{c}_{\varphi} \right)$$

$$= \boldsymbol{\Lambda}_{\theta} \left(\boldsymbol{\varphi}^h(\xi) + \boldsymbol{c}_{\varphi} \right), \tag{A.4a}$$

and

$$d_{\alpha}^{h*}(\xi) = \sum_{I=1}^{n_{\text{cp}}^{d}} N_{I}^{p_{\text{d}}}(\xi) \mathbf{\Lambda}_{\theta} \mathbf{d}_{\alpha I}$$

$$= \mathbf{\Lambda}_{\theta} \sum_{I=1}^{n_{\text{cp}}^{d}} N_{I}^{p_{\text{d}}}(\xi) \mathbf{d}_{\alpha I}$$

$$= \mathbf{\Lambda}_{\theta} d_{\alpha}^{h}(\xi), \tag{A.4b}$$

 $\alpha \in \{1, 2\}$, where we utilize the partition of unity property of the NURBS basis functions. Therefore, substituting Eqs. (A.4a) and (A.4b) into Eqs. (12a)–(12f), and using $\Lambda_{\theta,s} = \mathbf{0}$, $c_{\varphi,s} = \mathbf{0}$, and $\Lambda_{\theta}^{T} \Lambda_{\theta} = \mathbf{1}$, the beam strain at the superposed configuration y^{h*} , we have

$$\varepsilon\left(\mathbf{y}^{h*}\right) = \varepsilon\left(\mathbf{y}^{h}\right),$$
 (A.5)

which represents the frame-invariance (objectivity) of the approximated beam strain.

A.3 Construction of warping basis functions

Here, we discuss the determination of the polynomial basis \mathbf{w}_1 , \mathbf{w}_2 , \mathbf{w}_3 , and \mathbf{w}_4 in Eq. (21). Note that this is a generalization of the formulation of Wackerfuß and Gruttmann [41] in order to account for the higher order strains in Eq. (10) from the beam kinematics.

A.3.1 Orthogonality condition

Here, we construct a complete set of polynomials from degree $\bar{m} + 1$ to M that are orthogonal to an arbitrary polynomial of degree \bar{m} . We first introduce the following arrays (Wackerfuß and Gruttmann [42])

$$\pi_{0} := [P_{(0,0)}],$$

$$\pi_{1} := [P_{(1,0)}, P_{(0,1)}],$$

$$\vdots$$

$$\pi_{M} := [P_{(M,0)}, P_{(M-1,1)}, \dots, P_{(1,M-1)}, P_{(0,M)}],$$
(A.6)



with $P_{(n,m)} := (\zeta^1)^n (\zeta^2)^m$, $(n,m) \in \mathbb{S}_M := \mathbb{P}_0 \cup \mathbb{P}_1 \cdots \cup \mathbb{P}_M$, where M denotes the maximum degree, and we define

$$\mathbb{P}_M := \left\{ (n, m) \in \mathbb{N}_0 \times \mathbb{N}_0 \middle| n + m = M \right\}.$$

Here, \mathbb{N}_0 denotes the set of positive integers, and the number of elements in \mathbb{P}_M is $|\mathbb{P}_M| = M + 1$. Further, we introduce a modified polynomial basis from degree $\bar{m} + 1$ as

$$\pi_{\bar{m}+1}^* := \left[P_{(\bar{m}+1,0)}^*, P_{(\bar{m},1)}^*, \dots, P_{(1,\bar{m})}^*, P_{(0,\bar{m}+1)}^* \right],
\pi_{\bar{m}+2}^* := \left[P_{(\bar{m}+2,0)}^*, P_{(\bar{m}+1,1)}^*, \dots, P_{(1,\bar{m}+1)}^*, P_{(0,\bar{m}+2)}^* \right],
\vdots
\pi_{M}^* := \left[P_{(M,0)}^*, P_{(M-1,1)}^*, \dots, P_{(1,M-1)}^*, P_{(0,M)}^* \right].$$
(A.7)

Here, we define, for each $(n, m) \in \mathbb{S}_M - \mathbb{S}_{\bar{m}} = \mathbb{P}_{\bar{m}+1} \cup \mathbb{P}_{\bar{m}+2} \cdots \cup \mathbb{P}_M$,

$$P_{(n,m)}^* := P_{(n,m)} + \sum_{k=0}^{\bar{m}} \pi_k \boldsymbol{\beta}_{(n,m)}^k, \tag{A.8}$$

with the coefficients $\boldsymbol{\beta}_{(n,m)}^k \in \mathbb{R}^{|\mathbb{P}_k|}$,

$$\boldsymbol{\beta}_{(n,m)}^{0} := \left[\beta_{(n,m)}^{(0,0)}\right]^{T},$$

$$\boldsymbol{\beta}_{(n,m)}^{1} := \left[\beta_{(n,m)}^{(1,0)}, \beta_{(n,m)}^{(0,1)}\right]^{T},$$

$$\vdots$$

$$\boldsymbol{\beta}_{(n,m)}^{\bar{m}} := \left[\beta_{(n,m)}^{(\bar{m},0)}, \beta_{(n,m)}^{(\bar{m}-1,1)}, \dots, \beta_{(n,m)}^{(0,\bar{m})}\right]^{T},$$
(A.9)

such that the following orthogonality to arbitrary polynomials up to degree \bar{m} is satisfied

$$\int_{\mathcal{A}} P_{(p,q)} P_{(n,m)}^* \, \mathrm{d}\mathcal{A} = 0, \quad \forall (p,q) \in \mathbb{S}_{\bar{m}}, \tag{A.10}$$

where $\mathbb{S}_{\bar{m}} := \mathbb{P}_0 \cup \mathbb{P}_1 \cdots \cup \mathbb{P}_{\bar{m}}$ whose number of elements is $|\mathbb{S}_{\bar{m}}| = (\bar{m} + 1)(\bar{m} + 2)/2$. Equation (A.8) can be rewritten in the compact form

$$P_{(n,m)}^* = P_{(n,m)} + \bar{\mathbf{w}} \boldsymbol{\beta}_{(n,m)}, \tag{A.11}$$

with $\bar{\mathbf{w}} := [\boldsymbol{\pi}_0, \boldsymbol{\pi}_1, \dots, \boldsymbol{\pi}_{\bar{m}}] \in \mathbb{R}^{1 \times |\mathbb{S}_{\bar{m}}|}$ and $\boldsymbol{\beta}_{(n,m)} := [\boldsymbol{\beta}_{(n,m)}^{0T}, \boldsymbol{\beta}_{(n,m)}^{1T}, \dots, \boldsymbol{\beta}_{(n,m)}^{\bar{m}T}]^T \in \mathbb{R}^{|\mathbb{S}_{\bar{m}}|}$. Substituting Eq. (A.11) into Eq. (A.10), we obtain the following system of linear equations to determine $\boldsymbol{\beta}_{(n,m)}$ for each $(n,m) \in \mathbb{S}_M - \mathbb{S}_{\bar{m}}$

$$\bar{\mathbb{W}}\boldsymbol{\beta}_{(n,m)} = -\int_{\mathcal{A}} \bar{\mathbf{w}}^{\mathrm{T}} P_{(n,m)} \,\mathrm{d}\mathcal{A},\tag{A.12}$$

where the system matrix is given by

$$\bar{\mathbb{W}} := \int_{A} \bar{\mathbf{w}}^{\mathrm{T}} \bar{\mathbf{w}} \, \mathrm{d}\mathcal{A} \in \mathbb{R}^{|\mathbb{S}_{\bar{m}}| \times |\mathbb{S}_{\bar{m}}|}. \tag{A.13}$$

Remark A.1 Since $\bar{\mathbb{W}}$ is invertible, the coefficients $\boldsymbol{\beta}_{(n,m)}$ for each $(n,m) \in \mathbb{S}_M - \mathbb{S}_{\bar{m}}$ can be uniquely determined by solving Eq. (A.12). It should be noted that Eq. (A.12) depends solely on the cross-section's initial shape and dimension. Therefore, this process to construct a complete set of polynomial basis functions from degree $\bar{m}+1$ to M, $\mathbf{w}:=\left[\boldsymbol{\pi}_{\bar{m}+1}^*,\boldsymbol{\pi}_{\bar{m}+2}^*,\ldots,\boldsymbol{\pi}_{M}^*\right]$ can be done in *pre-processing*. For a given (fixed) value of \bar{m} , the total time for determining the whole coefficients $\boldsymbol{\beta}_{(n,m)}$, $(n,m) \in \mathbb{S}_M - \mathbb{S}_{\bar{m}}$ is linearly proportional to the number of elements in $\mathbb{S}_M - \mathbb{S}_{\bar{m}}$, $|\mathbb{S}_M| - |\mathbb{S}_{\bar{m}}| = (M+1)(M+2)/2 - (\bar{m}+1)(\bar{m}+2)/2$, i.e., the time complexity of constructing \mathbf{w} is $O(M^2)$.

A.3.2 Polynomial basis for enhanced strains

Here, we show the construction of polynomial basis \mathbf{w}_1 , \mathbf{w}_2 , \mathbf{w}_3 , and \mathbf{w}_4 in Eq. (21), based on the method in Sect. A.3.1. The enhanced basis can be expressed by

$$\mathbf{w}_{i} = \left[\boldsymbol{\pi}_{\bar{m}_{i}+1}^{*}, \boldsymbol{\pi}_{\bar{m}_{i}+2}^{*}, \dots, \boldsymbol{\pi}_{m_{i}}^{*}\right], \quad i \in \{1, 2, 3, 4\},$$
(A.14)

where m_i denotes the maximum degree of the polynomial, and $\bar{m}_i < m_i$ denotes the maximum degree in the corresponding compatible strain component, i.e., from Eq. (11), we have $\bar{m}_1 = \bar{m}_2 = 0$, $\bar{m}_3 = 2$, and $\bar{m}_4 = 1$. We first show that the orthogonality condition in Eq. (18) can be simplified to the form of Eq. (A.10). Since the orthogonality condition in Eq. (18) should hold for all $\alpha \in \mathbb{R}^{d_a}$, we obtain

$$\int_{\mathcal{A}} \mathbf{\Gamma}(\zeta^{1}, \zeta^{2})^{\mathrm{T}} \underline{\mathbf{S}_{\mathrm{p}}} j_{0} \, \mathrm{d}\mathcal{A} = 0, \tag{A.15}$$

where $\underline{S_p} := \left[S_p^{11}, S_p^{22}, S_p^{33}, S_p^{12}, S_p^{13}, S_p^{23} \right]^T$, such that $S_p = S_p^{ij} G_i \otimes G_j$. Further, we make the following assumptions in the evaluation of the orthogonality condition in Eq. (A.15):

- S_p^{ij} has the same polynomial order in ζ^1 and ζ^2 as the conjugate strain component E_p^{ij} in Eq. (19).
- j₀ ≈ 1, which holds exactly only for initially straight beams or linear finite elements, see Remark 2.1. This makes the orthogonality condition solely depend on the cross-section's initial shape.

Then, we obtain the following orthogonality condition: For each $(n, m) \in \mathbb{S}_{m_i} - \mathbb{S}_{\bar{m}_i}$, we have

$$\int_{A} P_{(p,q)} P_{(n,m)}^{*} d\mathcal{A} = \mathbf{0}, \quad \forall (p,q) \in \mathbb{S}_{\tilde{m}_{i}}, \ i \in \{1, 2, 3, 4\},$$
(A.16)

which has the same form as Eq. (A.10). Subsequently, from Eq. (A.12), we obtain

$$\bar{\mathbb{W}}_{i} \boldsymbol{\beta}_{(n,m)} = -\int_{\mathcal{A}} \bar{\mathbf{w}}_{i}^{\mathrm{T}} P_{(n,m)} \, \mathrm{d}\mathcal{A}$$
 (A.17)



to determine the unknown coefficients $\boldsymbol{\beta}_{(n,m)} := \left[\boldsymbol{\beta}_{(n,m)}^{0\,\mathrm{T}}, \ldots, \boldsymbol{\beta}_{(n,m)}^{\bar{m}_i\,\mathrm{T}}\right]^{\mathrm{T}} \in \mathbb{R}^{\left|\mathbb{S}_{\bar{m}_i}\right|}$ in

$$P_{(n,m)}^* = P_{(n,m)} + \bar{\mathbf{w}}_i \, \boldsymbol{\beta}_{(n,m)}, \ (n,m) \in \mathbb{S}_{m_i} - \mathbb{S}_{\bar{m}_i}, \tag{A.18}$$

in each $\mathbf{w}_i \in \mathbb{R}^{1 \times d_i}$, $i \in \{1, 2, 3, 4\}$, where $\bar{\mathbf{w}}_i := \left[\boldsymbol{\pi}_0, \dots, \boldsymbol{\pi}_{\bar{m}_i}\right] \in \mathbb{R}^{1 \times \left|\mathbb{S}_{\bar{m}_i}\right|}$, and the system matrix is given by

$$\bar{\mathbb{W}}_i := \int_{\mathcal{A}} \bar{\mathbf{w}}_i^{\mathrm{T}} \bar{\mathbf{w}}_i \, \mathrm{d}\mathcal{A} \in \mathbb{R}^{\left|\mathbb{S}_{\bar{m}_i}\right| \times \left|\mathbb{S}_{\bar{m}_i}\right|}. \tag{A.19}$$

Here, d_i denotes the number of elements in the row array \mathbf{w}_i , and it is obtained by

$$d_i := \left| \mathbb{S}_{m_i} \right| - \left| \mathbb{S}_{\bar{m}_i} \right| = \frac{1}{2} (m_i + 1)(m_i + 2) - \frac{1}{2} (\bar{m}_i + 1)(\bar{m}_i + 2). \tag{A.20}$$

Therefore, from Remark A.1, the time complexity to construct \mathbf{w}_i is $O(m_i^2)$, $i \in \{1, 2, 3, 4\}$.

Appendix B: Time-stepping scheme

B.1 Verification of the energy-momentum conservation for continuous time

Here, we verify that energy–momentum conservation laws can be deduced from Eqs. (49a)–(49d) under appropriate boundary conditions.

B.1.1 Total linear momentum

Let $\delta y = [c_{\varphi}^{\mathrm{T}}, \mathbf{0}^{\mathrm{T}}, \mathbf{0}^{\mathrm{T}}]^{\mathrm{T}} \in \bar{\mathcal{V}}$ with a constant vector $c_{\varphi} \in \mathbb{R}^3$, which describes a superposed rigid body translation. Here, we assume no displacement boundary conditions, i.e., $\Gamma_{\mathrm{D}} = \emptyset$, so that $c_{\varphi} \neq \mathbf{0}$. Then, the internal virtual work vanishes, and from Eq. (49a), we obtain

$$c_{\varphi} \cdot \left\{ \frac{\mathrm{d}}{\mathrm{d}t} L(V) - f_{\mathrm{ext}}(t) \right\} = 0, \ \forall c_{\varphi} \in \mathbb{R}^3,$$
 (B.1)

with the time-dependent total external force,

$$\boldsymbol{f}_{\text{ext}}(t) := \int_0^L \bar{\boldsymbol{n}} \, \mathrm{d}s + \bar{\boldsymbol{n}}_0 \,. \tag{B.2}$$

Therefore, without any displacement boundary conditions, we have the balance of total linear momentum

$$\frac{\mathrm{d}}{\mathrm{d}t}L(V) = f_{\mathrm{ext}}(t). \tag{B.3}$$

This implies that the total linear momentum is conserved if no external force and displacement boundary condition are applied.



B.1.2 Total angular momentum

Let $\delta y := [(c_{\theta} \times \varphi)^{\mathrm{T}}, (c_{\theta} \times d_1)^{\mathrm{T}}, (c_{\theta} \times d_2)^{\mathrm{T}}]^{\mathrm{T}} \in \bar{\mathcal{V}}$ with a constant vector $c_{\theta} \in \mathbb{R}^3$, which represents a superposed (infinitesimal) rigid body rotation, and c_{θ} represents the rotation axis vector. Here, we also assume no displacement boundary conditions, i.e., $\Gamma_{\mathrm{D}} = \emptyset$, so that $c_{\theta} \neq 0$. Then, the internal virtual work vanishes, and from Eq. (49a), we obtain

$$c_{\theta} \cdot \left[\frac{\mathrm{d}}{\mathrm{d}t} J(y, V) - m_{\text{ext}}(y, t) \right] = 0, \ \forall c_{\theta} \in \mathbb{R}^{3},$$
 (B.4)

with the total external moment defined by

$$\boldsymbol{m}_{\mathrm{ext}}(\boldsymbol{y},t) := \int_{0}^{L} \left(\boldsymbol{\varphi} \times \bar{\boldsymbol{n}} + \boldsymbol{d}_{\alpha} \times \bar{\tilde{\boldsymbol{m}}}^{\alpha} \right) \mathrm{d}s + \left[\boldsymbol{\varphi} \times \bar{\boldsymbol{n}}_{0} + \boldsymbol{d}_{\alpha} \times \bar{\tilde{\boldsymbol{m}}}^{\alpha}_{0} \right]_{s \in \Gamma_{\mathrm{N}}}. \tag{B.5}$$

Therefore, without any displacement boundary conditions, we have the balance of total angular momentum

$$\frac{\mathrm{d}}{\mathrm{d}t}J(y,V) = m_{\mathrm{ext}}(y,t). \tag{B.6}$$

B.1.3 Total energy

We define the total energy at an equilibrium solution of Eqs. (49a)–(49d), by

$$E(\mathbf{y}, \mathbf{V}, \boldsymbol{\varepsilon}_{p}, \boldsymbol{\alpha}, t) := \mathcal{K}(\mathbf{V}) + U(\boldsymbol{\varepsilon}_{p}, \boldsymbol{\alpha}) - W_{\text{ext}}(\mathbf{y}, t), \tag{B.7}$$

where the internal energy can be evaluated, from Eq. (22), by

$$U(\boldsymbol{\varepsilon}_{\mathbf{p}}, \boldsymbol{\alpha}) := \int_{0}^{L} \psi(\boldsymbol{\varepsilon}_{\mathbf{p}}, \boldsymbol{\alpha}) \, \mathrm{d}s. \tag{B.8}$$

We choose $\delta y = \dot{y} \in \bar{\mathcal{V}}$, and have the compatibility condition $V = \dot{y}$ from Eq. (50). Then, from Eq. (49a), we obtain

$$\frac{\mathrm{d}}{\mathrm{d}t}E(\mathbf{y}, \mathbf{V}, \boldsymbol{\varepsilon}_{\mathrm{p}}, \boldsymbol{\alpha}, t) = -\frac{\partial}{\partial t}W_{\mathrm{ext}}(\mathbf{y}, t). \tag{B.9}$$

This means that the total energy is conserved unless the external load explicitly depends on time. That is, for a conservative system, we have

$$\frac{\mathrm{d}}{\mathrm{d}t}E(\mathbf{y}, \mathbf{V}, \boldsymbol{\varepsilon}_{\mathrm{p}}, \boldsymbol{\alpha}) = 0, \tag{B.10}$$

where the total energy is

$$E(\mathbf{y}, \mathbf{V}, \boldsymbol{\varepsilon}_{p}, \boldsymbol{\alpha}) = \mathcal{K}(\mathbf{V}) + U(\boldsymbol{\varepsilon}_{p}, \boldsymbol{\alpha}) + U_{\text{ext}}(\mathbf{y}).$$
 (B.11)

B.2 Verification of the energy-momentum conservation in discrete-time intervals

We show that the time-stepping scheme in Eqs. (58a)–(58d) preserves the conservation properties of the continuous form in Sect. B.1.



B.2.1 Total linear momentum

Let $\delta y = \begin{bmatrix} c_{\varphi}^T, \mathbf{0}^T, \mathbf{0}^T \end{bmatrix}^T \in \bar{\mathcal{V}}$ with a constant vector $\mathbf{c}_{\varphi} \in \mathbb{R}^3$. Here, we assume no displacement boundary condition, i.e., $\Gamma_D = \emptyset$, so that $\mathbf{c}_{\varphi} \neq \mathbf{0}$. Then, we obtain the discrete form of the linear momentum balance from Eq. (58a) as

$$\frac{1}{\Delta t} \left\{ L\left(^{n+1}V\right) - L\left(^{n}V\right) \right\} = f_{\text{ext}}\left(t_{n} + \frac{1}{2}\Delta t\right), \tag{B.12}$$

which is consistent with Eq. (B.1). Therefore, the total linear momentum is conserved if no external force and displacement boundary conditions are applied.

B.2.2 Total angular momentum

We choose

$$\delta \mathbf{y} = \begin{cases} \mathbf{c}_{\theta} \times {}^{n+\frac{1}{2}} \mathbf{\varphi} \\ \mathbf{c}_{\theta} \times {}^{n+\frac{1}{2}} \mathbf{d}_{1} \\ \mathbf{c}_{\theta} \times {}^{n+\frac{1}{2}} \mathbf{d}_{2} \end{cases} \in \bar{\mathcal{V}}, \tag{B.13}$$

with a constant vector $c_{\theta} \in \mathbb{R}^3$, and assume no displacement boundary conditions, i.e., $\Gamma_D = \emptyset$, so that $c_{\theta} \neq 0$. Then, from Eq. (58a), we obtain

$$\frac{1}{\Delta t} \left\{ \boldsymbol{J} \left({}^{n+1}\boldsymbol{y}, {}^{n+1}\boldsymbol{V} \right) - \boldsymbol{J} \left({}^{n}\boldsymbol{y}, {}^{n}\boldsymbol{V} \right) \right\} = \boldsymbol{m}_{\text{ext}} \left({}^{n+\frac{1}{2}}\boldsymbol{y}, t_{n} + \frac{1}{2}\Delta t \right), \tag{B.14}$$

which is consistent with Eq. (B.6). Therefore, the angular momentum is conserved if no external loads and boundary conditions are applied. Note that this holds for both the conventional mid-point rule and the present EMC scheme.

B.2.3 Total energy

We choose $\delta y = {}^{n+1}y - {}^ny \in \bar{\mathcal{V}}$. Then, in Eq. (58a), using Eq. (56), we have

$$\frac{2}{\Delta t}G_{\text{iner}}\left(\frac{1}{\Delta t} \binom{n+1}{y} - {}^{n}y\right) - {}^{n}V, {}^{n+1}y - {}^{n}y\right)$$

$$= \frac{1}{2} \int_{0}^{L} \binom{n+1}{y} + {}^{n}V \cdot \mathcal{M} \binom{n+1}{y} - {}^{n}V \cdot ds$$

$$= \mathcal{K}\binom{n+1}{y} - \mathcal{K}\binom{n}{y}.$$
(B.15)

Further, due to the linearity of the operator $\mathbb{B}(y)$ with respect to y, we can obtain

$$\mathbb{B}\left(^{n+\frac{1}{2}}y\right)\delta y = \varepsilon\left(^{n+1}y\right) - \varepsilon\left(^{n}y\right). \tag{B.16}$$

Then, from Eq. (26a), we have

$$G_{\text{int}}^{\mathbf{y}}\left(^{n+\frac{1}{2}}\mathbf{y},^{n+\frac{1}{2}}\mathbf{r}_{\mathbf{p}},^{n+1}\mathbf{y}-^{n}\mathbf{y}\right)=\int_{0}^{L}\left\{\boldsymbol{\varepsilon}\left(^{n+1}\mathbf{y}\right)-\boldsymbol{\varepsilon}\left(^{n}\mathbf{y}\right)\right\}\cdot^{n+\frac{1}{2}}\mathbf{r}_{\mathbf{p}}\,\mathrm{d}s.\tag{B.17}$$



By choosing $\delta \boldsymbol{\varepsilon}_{p} = {}^{n+1}\boldsymbol{\varepsilon}_{p} - {}^{n}\boldsymbol{\varepsilon}_{p} \in \mathcal{V}_{p}$ and $\delta \boldsymbol{\alpha} = {}^{n+1}\boldsymbol{\alpha} - {}^{n}\boldsymbol{\alpha} \in \mathcal{V}_{a}$ in Eqs. (58c) and (58d), respectively, and combining the resulting equations, we obtain the identity,

$$\int_{0}^{L} \left\{ \psi \left({^{n+1}\boldsymbol{\varepsilon}_{\mathbf{p}}}, {^{n+1}\boldsymbol{\alpha}} \right) - \psi \left({^{n}\boldsymbol{\varepsilon}_{\mathbf{p}}}, {^{n}\boldsymbol{\alpha}} \right) \right\} \, \mathrm{d}s = \int_{0}^{L} \left({^{n+1}\boldsymbol{\varepsilon}_{\mathbf{p}}} - {^{n}\boldsymbol{\varepsilon}_{\mathbf{p}}} \right) \cdot {^{n+\frac{1}{2}}\boldsymbol{r}_{\mathbf{p}}} \, \mathrm{d}s. \tag{B.18}$$

Note that, here, we assume linear constitutive laws. Evaluating Eq. (58b) at time $t = t_{n+1}$ and $t = t_n$ with $\delta \mathbf{r}_p = {}^{n+\frac{1}{2}}\mathbf{r}_p \in \mathcal{V}_p$, we have

$$\int_{0}^{L} {n+\frac{1}{2}\mathbf{r}_{p}} \cdot \left\{ \boldsymbol{\varepsilon} \left({n+1} \, \boldsymbol{y} \right) - {n+1} \, \boldsymbol{\varepsilon}_{p} \right\} ds = 0, \tag{B.19a}$$

$$\int_{0}^{L} {n + \frac{1}{2} \mathbf{r}_{p} \cdot \left\{ \boldsymbol{\varepsilon} \left({}^{n} \mathbf{y} \right) - {}^{n} \boldsymbol{\varepsilon}_{p} \right\} ds} = 0, \tag{B.19b}$$

respectively. Equations (B.19a) and (B.19b) imply that Eqs. (B.17) and (B.18) are the same so that we obtain the identity

$$G_{\text{int}}^{\mathbf{y}}\left(^{n+\frac{1}{2}}\mathbf{y},^{n+\frac{1}{2}}\mathbf{r}_{\mathbf{p}},^{n+1}\mathbf{y} - {}^{n}\mathbf{y}\right) = \int_{0}^{L} \left\{ \psi\left(^{n+1}\boldsymbol{\varepsilon}_{\mathbf{p}},^{n+1}\boldsymbol{\alpha}\right) - \psi\left(^{n}\boldsymbol{\varepsilon}_{\mathbf{p}},^{n}\boldsymbol{\alpha}\right) \right\} \, \mathrm{d}s$$
$$= U(^{n+1}\boldsymbol{\varepsilon}_{\mathbf{p}},^{n+1}\boldsymbol{\alpha}) - U(^{n}\boldsymbol{\varepsilon}_{\mathbf{p}},^{n}\boldsymbol{\alpha}). \tag{B.20}$$

Substituting Eqs. (B.15) and (B.20) into Eq. (58a), we have

$$\mathcal{K}(^{n+1}V) + U(^{n+1}\varepsilon_{p}, ^{n+1}\alpha) - W_{\text{ext}}\left(^{n+1}y, t_{n} + \frac{1}{2}\Delta t\right)$$

$$= \mathcal{K}(^{n}V) + U(^{n}\varepsilon_{p}, ^{n}\alpha) - W_{\text{ext}}\left(^{n}y, t_{n} + \frac{1}{2}\Delta t\right). \tag{B.21}$$

This implies the following:

 If the applied load is (piece-wise) linearly proportional to time, we obtain the following identity

$$n+1E = n+1E^*, \quad n = 0, 1, \dots,$$
 (B.22)

where we define

$$^{n+1}E := E\left(^{n+1}y,^{n+1}V,^{n+1}\boldsymbol{\varepsilon}_{p},^{n+1}\boldsymbol{\alpha},t_{n+1}\right),$$

$$^{n+1}E^* := {^n}E - \overline{\Delta_t W_{\text{ext}}}$$

with

$$^{n}E := E(^{n}y, ^{n}V, ^{n}\varepsilon_{p}, ^{n}\alpha, t_{n}),$$

and

$$\overline{\Delta_t W_{\text{ext}}} := \frac{1}{2} \left(\left. \frac{\partial W_{\text{ext}}}{\partial t} \right|_{\mathbf{y} = {}^{n+1}\mathbf{y}, \, t = t_{n+1}} + \left. \frac{\partial W_{\text{ext}}}{\partial t} \right|_{\mathbf{y} = {}^n\mathbf{y}, \, t = t_n} \right) \Delta t.$$



p	$n_{ m el}$	Case 1: $w = 1/3$, $h = 1$				Case 2: $w = 1/30$, $h = 1/10$			
		[mix.loc-sr]		[mix.glo]		[mix.loc-sr]		[mix.glo]	
		n_{load}	n _{iter}	n_{load}	n _{iter}	n_{load}	n _{iter}	n_{load}	$n_{ m iter}$
2	24	96	×	16	105	96	×	16	103
	48	96	641	16	103	96	523	16	101
	96	96	673	16	103	192	973	16	100
	192	16	224	16	102	16	183	16	100
	384	384	×	20	161	16	208	16	100
3	24	16	105	16	105	16	105	16	103
	48	16	104	16	103	16	103	16	101
	96	16	104	16	103	16	103	16	100
	192	16	116	16	114	16	103	16	100
	384	16	144	20	175	16	103	16	100
4	24	16	150	16	105	16	109	16	103

Table C.1 Twisting of an elastic ring: Total number of load steps (n_{load}) with uniform increments, and total number of iterations (n_{iter}) , for $\bar{\theta}=2\pi$ in the results of Fig. 20. Here, the symbol \times represents a failure in convergence of the equilibrium iteration

• If the external load has no explicit time-dependence, the total energy is conserved, i.e., ${}^{n+1}E = {}^{n}E$.

Appendix C: Numerical examples

C.1 An analytical solution for a pure torsion problem

We consider a pure torsion of a straight bar of length L and a shear modulus $G = E/(2(1 + \nu))$. In the isotropic linear elasticity, the torsional angle (radians) of the bar under a twisting moment M_T can be expressed by (Budynas and Sadegh [4, page 350])

$$\theta = \frac{M_{\rm T}L}{KG},\tag{C.1}$$

where K is a factor that depends on the cross-sectional shape and dimensions. For a circular cross-section, the cross-section remains plane under the moment load, and K is equal to the polar moment of inertia, I_p . For other cross-sections, the cross-section warps, and K is less than I_p . For a (solid) rectangular cross-section of dimensions 2a and 2b ($a \ge b$), we have a solution for K in Budynas and Sadegh [4, page 366, Table 10.7],

$$K = ab^{3} \left\{ \frac{16}{3} - 3.36 \frac{b}{a} \left(1 - \frac{b^{4}}{12a^{4}} \right) \right\}, \tag{C.2}$$

obtained from a truncation of an infinite series solution of the St. Venant torsion problem, which can be found in Timoshenko and Goodier [40, page 288]. From Eq. (C.1),



 $KG/L = M_T/\theta$ represents the torsional stiffness of the bar. Note that Eq. (C.2) is based on the assumptions (Budynas and Sadegh [4, page 349]): (i) The bar is straight, (ii) the material is homogeneous, isotropic, and linearly elastic, (iii) the deformation is purely torsional under equal and opposite twisting moments at the ends.

C.2 Total number of load steps in example 4

Here, we present total number of load steps and iterations in the results of Fig. 20. For the formulation $[\Delta\theta]$, we use in total 200 and 2000 load steps with uniform increments in cases 1 and 2, respectively. Table C.1 shows the total number of load steps and iterations in the results from using [mix.glo] and [mix.loc-sr]. We have observed that, using [mix.loc-sr] with p=2 and $n_{\rm el}=24$, the ring does not deform into a smaller one, and the equilibrium iteration eventually diverges.

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Declarations

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