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# Numerical study of time-dependent deformation and cracking in brittle rocks with phase-field method and application to slope instability analysis

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## Abstract

Most rocks exhibit time-dependent deformation and failure. Two main mechanisms are generally considered, the progressive growth of cracks and viscoelastic and/or viscoplastic deformation. In this study, cracking process is described by a viscous phase-field method which is coupled with a viscoplastic model. The evolution of crack field is controlled by both elastic and viscoplastic tensile volumetric and deviatoric strains. The threshold of viscoplastic deformation is weakened by the growth of cracks. The efficiency of the proposed model is first assessed by comparing numerical predictions with experimental data in triaxial compression and creep tests. Then, the proposed model is applied to modeling time-dependent deformation and failure process of a high slope section in the left bank of Jinping-I hydropower station in China. Numerical predictions are compared with field measurements.

*Keywords:* Damage, Cracking, viscoplastic deformation, Brittle rocks Phase-field method, Slope instability

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## 1. Introduction

- Damage due to initiation and propagation of cracks is commonly observed in brittle rocks.
- Macroscopic failure of these materials is generally driven by the transition from micro-cracks

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4 to macro-cracks or fractures [1, 2]. Further, the so-called sub-critical slow propagation of  
5 micro-cracks and their coalescence [3, 4] can lead to time-dependent deformation and failure.  
6 Different types of experimental studies have been reported. For instance, in the framework  
7 of linear fracture mechanics, double torsion tests were performed to identify the sub-critical  
8 growth kinetics of cracks in various rocks [5, 6, 7, 8].

9 On the other hand, most rocks also exhibits viscoelastic and viscoplastic deformation  
10 under different stress states and environmental conditions [9, 10, 11, 12, 13]. According  
11 to previous studies [10, 14], the irrecoverable viscoplastic deformation should be the main  
12 responsible for the time-dependent delayed failure of rock engineering structures. Moreover,  
13 in many rocks, the viscoplastic deformation is intimately coupled with the growth of cracks.  
14 The macroscopic plastic and viscoplastic deformation can be partially related to the fric-  
15 tional sliding along micro-crack surfaces [15, 16, 17]. The viscoplastic deformation can be  
16 enhanced by the induced growth of cracks while the cracking process is affected by the creep  
17 deformation of materials. Therefore, it is primordial to take into account such a coupling  
18 phenomenon between cracking and viscoplastic deformation.

19 From the point of view of engineering application, instability and failure of engineered  
20 rock structures and natural systems can also be caused by time-dependent crack growth  
21 and viscoplastic deformation. For instance, many catastrophic landslides were induced by  
22 the accumulation of creep deformation and slow crack propagation and localization [18, 19].  
23 Before the final failure, a gradual increase of deformation over time was monitored in a  
24 number of typical landslides [20, 21]. For instance, landslide of a natural slope caused  
25 by tunnel excavation was reported in [22]. In that accident, the slope failure occurred  
26 a few days after an arriving entrance of the tunnel was cut through. Progressive failure  
27 was clearly observed while no evidence of rainfall or underground water movement could  
28 be monitored [23]. Some experimental investigations were performed to understand the  
29 physical mechanisms involved in rainfall induced landslides [24, 25]. In many cases, natural  
30 hazards should be analyzed as evolutionary processes by taking into account time-dependent  
31 mechanical behavior of geological formations [26], in particular creep deformation and slow  
32 growth of cracks. The widely used limit equilibrium methods mainly focused on ultimate

33 instability mechanisms and do not considered progressive failure processes involved.

34 In order to complete the limit analysis methods and provide a finer description of natu-  
35 ral hazards, different types of numerical methods have been developed. Briefly, two families  
36 of methods are available. In the first family, continuum mechanics based approaches are  
37 developed by using suitable constitutive models. Some of them were applied to analyzing  
38 progressive deformation and stability of rainfall-induced landslides [27, 28, 29, 30]. Viscoelas-  
39 tic and viscoplastic models were widely used to describe the time-dependent deformation  
40 of rocks and soils. For instance, the extended finite element method (XFEM) was used  
41 in [31, 32]. The mechanical response of materials has been studied by incorporating basis  
42 functions extracted from viscoelastic asymptotic fields. However, special treatments are nec-  
43 essary in simulating the complex crack topology, branching and multi-cracks configurations  
44 [33]. In the second family, one can find various types of discrete methods considering rock  
45 formations as an assembly of particles and blocks [34]. In particular, the discrete element  
46 model (DEM) has been employed to study the creep behavior of rocks, such as in [35, 36].  
47 Fractures are created when the bonds between particles or blocks are broken, resulting that  
48 fracturing paths are not arbitrary but depend on the initial arrangement of particles and  
49 blocks [37].

50 During the recent years, the so-called phase-method has attracted more and more inter-  
51 est in view of cracking modeling in various engineering fields including geological materials  
52 [38]. Inspired by the variational principle for linear fracture mechanics [39], this method  
53 was initially formulated for studying brittle fractures in elastic materials [40, 41]. Based on  
54 approximation methods of functions with jumps [42, 43, 44], the evolution of sharp crack  
55 surface area is approximated by that of a regularized crack surface density function, which  
56 depends on an auxiliary phase-field (damage) variable and its gradient. Consequently, the  
57 phase-field method provides a non-local approach to describe the continuous transition from  
58 diffuse damage (micro-cracks) to macro-cracks in the form of localized damage bands. Fur-  
59 thermore, the phase-field method is particularly suitable to deal with merging and branching  
60 of multiple cracks with no additional effort or heuristics [45]. For geological materials un-  
61 der compressive stresses, particular efforts were made on the description of shear and mixed

62 cracks [46, 47, 48], and for coupled thermo-hydronechanical problems [49, 50, 51]. Moreover,  
63 the application of phase-field method has also been extended from fractures in solids with  
64 infinitesimal strains to plates, shells and composites under finite strains [52]. In view of the  
65 dynamic cracking analysis, explicit phase field models were also reported [53]. In addition,  
66 a deep neural network (DNN) based solution for solving phase-field fracture was proposed  
67 by [54], which totally avoids a classical discretization to solve the underlying coupled par-  
68 tial differential equations. Despite all these advances, only few studies have been devoted  
69 to time-dependent fracturing in viscoplastic rocks [55]. However, as the main limitation  
70 of phase-field method, the propagation of sharp cracks is approximated by the growth of  
71 damage field and onset of localized damage bands. Consequently, the time-dependent crack-  
72 ing process is described by the time-dependent evolution of damage field coupled with the  
73 viscoplastic flow.

74 In the present work, we shall develop an efficient numerical modeling method of progres-  
75 sive cracking process in rocks exhibiting an elastic-viscoplastic behavior. The initiation and  
76 propagation of cracks is described by a time-dependent phase field method while the creep  
77 deformation of rocks by a viscoplastic model. The growth of phase field is affected by the  
78 viscoplastic deformation and inversely the threshold of viscoplastic flow is degraded by the  
79 induced cracks. The proposed numerical method is implemented in a standard finite element  
80 framework. Its efficiency is assessed through comparisons with both laboratory tests and  
81 field observations.

82

## 83 **2. Time-dependent phase-field model for viscoplastic rocks**

### 84 *2.1. Regularized crack surface description*

85 Let  $\Omega \subset \mathbb{R}^3$  be a volumetric domain of solid material with an external boundary  $\partial\Omega$ ,  
86 which contains a number of cracks  $\Gamma_i, i = 1, \dots, N$ . It is generally very delicate to trace  
87 the evolution of each individual crack during loading history. In order to overcome such  
88 difficulty, a regularized description of cracks has been introduced in the so-called phase-field  
89 method [40, 45, 56, 57]. The exact description of discontinuous sharp cracks is approximated

90 by a continuous volumetric distribution of crack surface density. More precisely, the total  
 91 area of sharp crack surfaces  $A_\Gamma$  is approximated a regularized quantity  $A_{\Gamma_d}$  as follows:

$$A_\Gamma = \sum_{i=1}^N A_{\Gamma_i} \cong A_{\Gamma_d} = \int_{\Omega} \Gamma_d(d, \nabla d) dV \quad (1)$$

92 The function  $\Gamma_d(d, \nabla d)$  denotes the crack surface area per unit initial volume ( $\text{m}^2/\text{m}^3$ ).  
 93 The auxiliary scalar variable  $d(\mathbf{x}) \in [0, 1]$  defines an undamaged state by 0 and a fully  
 94 damaged state by 1. It is usually called phase-field variable and physically equivalent to  
 95 that used in continuum damage mechanics. It is noticed that the crack density function  
 96  $\Gamma_d(d, \nabla d)$  is not only dependent on the local damage variable  $d$  but also on its gradient  
 97  $\nabla d$ . Therefore, the crack density is a non-local function of damage variable. This is an  
 98 essential feature allowing the regularization of damage localization process and avoiding the  
 99 related pathological mesh dependency. Among various choices for the crack surface density  
 100 function, the following widely used form introduced in [43] is adopted here:

$$\Gamma_d(d, \nabla d) = \frac{d^2}{2l_d} + \frac{l_d}{2} |\nabla d|^2 \quad (2)$$

101 The parameter  $l_d > 0$  is a length scale which defines the width of smeared cracks represented  
 102 by localized damage zones. Further, the cracking is an irreversible process. The evolution  
 103 rate of crack surface area should be not negative and it is usually defined as follows:

$$\dot{A}_{\Gamma_d} = \frac{d}{dt} \int_{\Omega} \Gamma_d(d, \nabla d) dV = \int_{\Omega} \dot{\Gamma}_d dV \geq 0 \quad (3)$$

104 with:

$$\dot{\Gamma}_d = \left(\frac{d}{l_d}\right) \dot{d} + l_d \nabla d \cdot \nabla \dot{d} \geq 0 \quad (4)$$

## 105 2.2. Energy functional

106 In this study, we consider an isotropic rock material exhibiting an elastic-viscoplastic  
 107 behavior and the transition from diffuse damage to localized cracking which is described by  
 108 the phase-field method. With the assumption of small strains, the total strain tensor  $\boldsymbol{\varepsilon}$  is  
 109 conventionally divided into an elastic part  $\boldsymbol{\varepsilon}^e$  and a viscoplastic part  $\boldsymbol{\varepsilon}^{vp}$  such as  $\boldsymbol{\varepsilon} = \boldsymbol{\varepsilon}^e + \boldsymbol{\varepsilon}^{vp}$ .  
 110 The total energy functional can be expressed in the following general form:

$$E(\boldsymbol{\varepsilon}^e, \boldsymbol{\varepsilon}^{vp}, d) = \int_{\Omega} [w^e(\boldsymbol{\varepsilon}^e, d) + w^{vp}(\boldsymbol{\varepsilon}^{vp}, d)] dv + E^{crack} \quad (5)$$

111  $w^e$  is the elastic strain energy density per initial unit volume of damaged material.  $w^{vp}$   
 112 denotes the viscoplastic strain energy density (dissipated and stored). The third term  $E^{crack}$   
 113 represents the energy needed to create cracks in  $\Omega$ .

### 114 2.2.1. Elastic strain energy

115 The elastic strain energy of material is affected by cracking process and it is a function  
 116 of the damage variable  $d$ . Inversely, the damage evolution is driven by the growth of elastic  
 117 strain energy. In this work, it is assumed that the damage evolution is motivated only the  
 118 tensile (positive) volumetric strain and deviatoric (or shear) strains. For this purpose, the  
 119 elastic strain energy of undamaged material is conveniently decomposed as follows:

$$\begin{cases} w_0^e(\boldsymbol{\varepsilon}^e) = w_{0-}^e(\boldsymbol{\varepsilon}^e) + w_{0+}^e(\boldsymbol{\varepsilon}^e) \\ w_{0-}^e = \frac{k_0}{2} [\langle \text{tr}(\boldsymbol{\varepsilon}^e) \rangle_-]^2 \\ w_{0+}^e = \frac{k_0}{2} [\langle \text{tr}(\boldsymbol{\varepsilon}^e) \rangle_+]^2 + \mu_0 \boldsymbol{\varepsilon}_{dev}^e : \boldsymbol{\varepsilon}_{dev}^e \end{cases}, \boldsymbol{\varepsilon}_{dev}^e = \boldsymbol{\varepsilon}^e - \frac{\text{tr}(\boldsymbol{\varepsilon}^e)}{3} \mathbf{I} \quad (6)$$

120  $k_0$  and  $\mu_0$  are respectively the elastic bulk and shear modulus of undamaged material.  $\langle x \rangle_+ =$   
 121  $x$  if  $x > 0$  and  $\langle x \rangle_+ = 0$  if  $x \leq 0$ , inversely for  $\langle x \rangle_-$ .  $\mathbf{I}$  is the second-order identity tensor.  
 122 Accordingly, the elastic strain density function of damaged material is written as:

$$w^e(\boldsymbol{\varepsilon}^e, d) = w_{0-}^e(\boldsymbol{\varepsilon}^e) + g(d)w_{0+}^e(\boldsymbol{\varepsilon}^e, d) \quad (7)$$

123 The so-called degradation function  $g(d)$  defines the coupling between the elastic strain energy  
 124 and damage evolution. The elastic stress-strain relations of damaged material are obtained  
 125 from the standard derivation of  $w^e$ :

$$\boldsymbol{\sigma} = \frac{\partial w^e}{\partial \boldsymbol{\varepsilon}^e} = \mathbb{C}(d) : \boldsymbol{\varepsilon}^e \quad (8)$$

126  $\boldsymbol{\sigma}$  is Cauchy stress tensor.  $\mathbb{C}(d)$  denotes the elastic stiffness tensor of damaged material and  
 127 it is here given by:

$$\mathbb{C}(d) = r_e^- k_0 \mathbf{I} \otimes \mathbf{I} + g(d)[r_e^+ k_0 \mathbf{I} \otimes \mathbf{I} + 2\mu_0(\mathbb{I} - \frac{1}{3} \mathbf{I} \otimes \mathbf{I})] \quad (9)$$

128  $\mathbb{I}$  is the fourth-order symmetric identity tensor. The coefficients  $r_e^\pm = \frac{1}{2} \{ \text{sign}[\pm \text{tr}(\boldsymbol{\varepsilon}^e)] + 1 \}$   
 129 describe the fact that the bulk modulus  $k$  is affected by open cracks ( $r_e^+ = 1$  and  $r_e^- = 0$ ) but  
 130 not affected by closed shear cracks ( $r_e^+ = 0$  and  $r_e^- = 1$ ).

131 It is noticed that the choice of  $g(d)$  leads to different types of elastic properties degrada-  
 132 tion. A review on the choice of degradation function and its consequences can be found in  
 133 [58]. Among various forms of  $g(d)$  proposed in previous studies [40, 57, 49], the most widely  
 134 used one is  $g(d) = (1 - d)^2$  by satisfying the condition  $g'(d = 1) = 0$ . It is adopted in this  
 135 study.

### 136 2.2.2. Viscoplastic strain energy

137 Furthermore, since the concept of elasto-viscoplasticity was firstly introduced by [59],  
 138 the elasto-viscoplastic behavior of rocks has been widely discussed and some significant  
 139 progresses have been reported in [60, 11]. Moreover, An elasto-viscoplastic model based on  
 140 Perzyna's general theory was proved to be in good agreement with both laboratory and in  
 141 situ data by [61]. Consequently, the classical Perzyna's formulation [59, 62] will be coupled  
 142 with the phase-field method to describe the time-dependent behavior of brittle rocks in this  
 143 paper.

144 Time-dependent deformation of rocks can be related to various mechanisms such as  
 145 viscoplastic flow, sub-critical growth of micro-cracks etc. In this work, for the sake of  
 146 simplicity, it is assumed that the time-dependent deformation and cracking is mainly driven  
 147 by the viscoplastic flow. Further, by putting the emphasis on cracking modeling by the  
 148 phase-field method, the viscoplastic deformation is described by using the classical Perzyna's  
 149 formulation [59, 62] and a simple constitutive model.

150 The accumulated viscoplastic strain  $\boldsymbol{\varepsilon}^{vp}$  is calculated by time integration of its flow rate  
 151  $\dot{\boldsymbol{\varepsilon}}^{vp}$ :

$$\boldsymbol{\varepsilon}^{vp}(t) = \int_0^t \dot{\boldsymbol{\varepsilon}}^{vp}(\tau) d\tau \quad (10)$$

152 The viscoplastic strain rate is defined by the following widely used exponential form:

$$\dot{\boldsymbol{\varepsilon}}^{vp} = \eta \langle f^{vp}(\boldsymbol{\sigma}) \rangle_+^n \dot{\lambda} \frac{\partial G^{vp}}{\partial \boldsymbol{\sigma}} \quad (11)$$

153  $\eta$  is a viscosity parameter controlling the initial rate of viscoplastic flow.  $\langle \rangle_+$  are the Macaulay  
 154 brackets.  $f^{vp}$  represents the viscoplastic loading function. Similarly to some previous studies  
 155 [63], it is assumed that the viscoplastic flow of rocks can be enhanced by the induced damage.

156 Therefore, the following damage-coupled form is adopted [64]:

$$f^{vp}(\boldsymbol{\sigma}) = \frac{q - g_{vp}(d)\sigma_s}{p_r} \quad (12)$$

157  $q = \sqrt{3\mathbf{s} : \mathbf{s}/2}$  is the conventional generalized deviatoric stress, with  $\mathbf{s} = \boldsymbol{\sigma} - \frac{\text{tr}(\boldsymbol{\sigma})}{3}\mathbf{I}$  being  
 158 the deviatoric stress tensor. The parameter  $\sigma_s$  denotes the initial threshold for viscoplastic  
 159 flow and  $p_r = 1$  MPa is a fixed normalizing stress.  $g_{vp}(d)$  is the degradation function for  
 160 viscoplastic flow. The use of product form in  $g_{vp}(d)\sigma_s$  means that the viscoplastic threshold  
 161 is progressively reduced by the growth of damage. For the sake of simplicity, the same form  
 162 as the elastic degradation function is here used:  $g_{vp}(d) = g(d)$ .

163 The coefficient  $\dot{\lambda}$  defines the current amplitude of the viscoplastic strain rate, which is a  
 164 function of the accumulated viscoplastic strain:

$$\dot{\lambda} = (1 - \varepsilon_{vp}^{eq})^m, \text{ with } \varepsilon_{vp}^{eq} = \int_0^t \sqrt{\frac{2}{3}\dot{\boldsymbol{\varepsilon}}_{dev}^{vp}(\tau) : \dot{\boldsymbol{\varepsilon}}_{dev}^{vp}(\tau)} d\tau \quad (13)$$

165 in which  $\dot{\boldsymbol{\varepsilon}}_{dev}^{vp} = \dot{\boldsymbol{\varepsilon}}^{vp} - \frac{\text{tr}(\dot{\boldsymbol{\varepsilon}}^{vp})}{3}\mathbf{I}$  is the viscoplastic deviatoric strain tensor.  $n$  and  $m$  are two  
 166 material parameters.

167 In the viscoplastic flow rule (11), the scalar-valued function  $G^{vp}$  is a potential which  
 168 defines the components of viscoplastic strain rate. Different forms are available in literature.  
 169 For hard rocks, it was widely assumed that the viscoplastic flow produced the time-dependent  
 170 deviatoric strains only. Therefore, a very simply viscoplastic flow potential  $G^{vp} = q$  was used  
 171 in some previous studies [65, 64]. However, according to laboratory creep tests on Jinping  
 172 marble diabase [13], large volumetric creep strain was also obtained. For this reason, the  
 173 following viscoplastic flow potential is proposed in this work:

$$G^{vp} = q + c_v p \quad (14)$$

174 where  $p = \text{tr}(\boldsymbol{\sigma})/3$  is the mean stress. The parameter  $c_v$  defines the volumetric viscoplastic  
 175 strain rate.

176 In consistency with the elastic strain energy decomposition (6), the viscoplastic strain  
 177 energy of undamaged material is also decomposed into two parts: respectively coupled

178 and not with the induced damage  $d$ . Therefore, the viscoplastic strain energy of damaged  
 179 materials is given by:

$$\begin{cases} w^{vp}(\boldsymbol{\varepsilon}^{vp}, d) = w_{0-}^{vp}(\boldsymbol{\varepsilon}^{vp}) + g(d)w_{0+}^{vp}(\boldsymbol{\varepsilon}^{vp}) \\ w_{0-}^{vp} = \int_0^t \boldsymbol{\sigma}(\tau) : \dot{\boldsymbol{\varepsilon}}_{sph-}^{vp}(\tau) d\tau \\ w_{0+}^{vp} = \int_0^t \boldsymbol{\sigma}(\tau) : [\dot{\boldsymbol{\varepsilon}}_{sph+}^{vp}(\tau) + \dot{\boldsymbol{\varepsilon}}_{dev}^{vp}(\tau)] d\tau \end{cases}, \begin{cases} \dot{\boldsymbol{\varepsilon}}_{sph-}^{vp} = \frac{\langle \text{tr}(\dot{\boldsymbol{\varepsilon}}^{vp}) \rangle_-}{3} \mathbf{I} \\ \dot{\boldsymbol{\varepsilon}}_{sph+}^{vp} = \frac{\langle \text{tr}(\dot{\boldsymbol{\varepsilon}}^{vp}) \rangle_+}{3} \mathbf{I} \\ \dot{\boldsymbol{\varepsilon}}_{dev}^{vp} = \dot{\boldsymbol{\varepsilon}}^{vp} - \frac{\text{tr}(\dot{\boldsymbol{\varepsilon}}^{vp})}{3} \mathbf{I} \end{cases} \quad (15)$$

### 180 2.2.3. Crack growth related energy

181 In the variational framework for fracture mechanics, the crack growth or the evolution  
 182 of damage variable is seen as an energy minimization problem [39, 66]. As in previous  
 183 studies [67, 57], the energy related to fracture growth is here assumed to be fully dissipated.  
 184 Moreover, the classical Griffith-type critical energy release rate is here adopted to quantify  
 185 the work needed for an instantaneous crack surface creation [68]. Therefore, using the crack  
 186 surface density defined in (2), the total energy dissipated in instantaneous crack creation  
 187 can be expressed as follows:

$$E^{crack-ins} = \int_{\Omega} g_c \Gamma_d dV \quad (16)$$

188 The parameter  $g_c$  denotes the energy needed for the creation of unit crack surface in unit  
 189 initial volume ( $N/m$ ). By assuming that the value of  $g_c$  is time-independent, the crack  
 190 energy dissipation rate can be calculated as:

$$\dot{E}^{crack-ins} = \int_{\Omega} g_c \dot{\Gamma}_d dV = \int_{\Omega} g_c \left[ \left( \frac{d}{l_d} \right) \dot{d} + l_d \nabla d \cdot \nabla \dot{d} \right] dV \quad (17)$$

191 In some situations, it is important to take into account a time-dependent or rate-dependent  
 192 growth of cracks. Therefore, we shall also quantify the energy needed for such a crack growth  
 193 process. Based on previous studies [69, 70], the following specific form is adopted:

$$\dot{E}^{crack-vis} = \int_{\Omega} \eta_d \dot{d}^2 dV \quad (18)$$

194 in which  $\eta_d \geq 0$  ( $N.s/m^2$ ) is a viscosity parameter. Accordingly, the total dissipation rate  
 195 crack growth is given by:

$$\dot{E} = \dot{E}^{crack-ins} + \dot{E}^{crack-vis} \quad (19)$$

196 *2.3. Governing equations*

197 The mechanical boundary values problem with crack growth (or phase-field evolution)  
 198 can be solved by the minimization of the following potential function:

$$\dot{\Pi}(\dot{\mathbf{u}}, \dot{d}) = \dot{E}(\dot{\mathbf{u}}, \dot{d}) - \dot{P}(\dot{\mathbf{u}}) = 0 \quad (20)$$

199 By using the elastic strain energy, viscoplastic strain energy and crack growth energy defined  
 200 above, the rate of the total energy functional  $\dot{E}$  is given by:

$$\begin{aligned} \dot{E}(\dot{\boldsymbol{\varepsilon}}^e, \dot{\boldsymbol{\varepsilon}}^{vp}, \dot{d}) &= \int_{\Omega} \dot{w}^e + \dot{w}^{vp} \, dV = \int_{\Omega} \left[ \frac{\partial(w^e + w^{vp})}{\partial d} \dot{d} + \boldsymbol{\sigma} : (\dot{\boldsymbol{\varepsilon}}^e + \dot{\boldsymbol{\varepsilon}}^{vp}) \right] dV \\ &+ \int_{\Omega} \left\{ g_c \left[ \left( \frac{d}{l_d} \right) \dot{d} + l_d \nabla d \cdot \nabla \dot{d} \right] + \eta_d \dot{d}^2 \right\} dV \end{aligned} \quad (21)$$

201 As the evolution of phase-field variable  $d$  is driven by elastic and viscoplastic strains (or  
 202 equivalently by displacement field  $\mathbf{u}$ ) [40, 56], the external applied work rate is simply  
 203 expressed as:

$$\dot{P} = \int_{\Omega} \mathbf{b} \cdot \dot{\mathbf{u}} \, dV + \int_{\partial\Omega_t} \mathbf{t} \cdot \dot{\mathbf{u}} \, dA \quad (22)$$

204 where  $\mathbf{b}$  is a given body force field per unit volume ( $N/m^3$ ) and  $\mathbf{t}$  a prescribed surface  
 205 traction ( $N/m^2$ ) on the external boundary  $\partial\Omega_t$ .

206 The energy balance condition  $\dot{\Pi} = 0$  leads to the following integral equation:

$$\begin{aligned} &\int_{\Omega} -[\text{div}(\boldsymbol{\sigma}) + \mathbf{b}] \cdot \dot{\mathbf{u}} \, dV + \int_{\partial\Omega} [\boldsymbol{\sigma} \cdot \mathbf{n} - \mathbf{t}] \cdot \dot{\mathbf{u}} \, dA \\ &+ \int_{\Omega} \left[ \frac{\partial(w^e + w^{vp})}{\partial d} + \frac{g_c}{l_d} d - g_c l_d \Delta d + \eta_d \dot{d} \right] \cdot \dot{d} \, dV + \int_{\partial\Omega} [g_c l_d \nabla d \cdot \mathbf{n}] \cdot \dot{d} \, dA = 0, \quad \forall \Omega \end{aligned} \quad (23)$$

207 in which  $\Delta d = \text{div}(\nabla d)$ . Accordingly, one gets easily the following local governing equations  
 208 for the mechanical (or displacement) problem:

$$\text{div}(\boldsymbol{\sigma}(t)) + \mathbf{b} = 0, \quad \text{with } \boldsymbol{\sigma}(t) = \mathbb{C}(d) : [\boldsymbol{\varepsilon} - \boldsymbol{\varepsilon}^{vp}(t)] \quad (24)$$

209 It is worth noticing that as the viscoplastic strains evolve with time, the current stress  
 210 tensor is also a function of time. However, for the sake of simplicity, it is assumed that the  
 211 time-dependent progressive deformation and cracking in rock structures can be seen as a

212 quasi-static process. Therefore, the inertia terms are not taken into account in the balance  
 213 equations. The governing equation for the phase-field evolution is given by:

$$\eta_d \dot{d} = 2(1-d)(w_{0+}^e + w_{0+}^{vp}) - g_c \left[ \frac{d}{l_d} - l_d \Delta d \right] \quad (25)$$

214 In order to verify the irreversible condition of crack growth  $\dot{\Gamma}_d \geq 0$  and  $\dot{d} \geq 0$ , the concept  
 215 of energy history functional introduced in [40] is here adopted and expressed by:

$$\mathcal{H}(t) = \max_{\tau \in [0, t]} [w_{0+}^e(\tau) + w_{0+}^{vp}(\tau)] \quad (26)$$

216 Consequently, the governing equation of phase-field evolution problem is rewritten as follows:

$$\eta_d \dot{d} = 2(1-d)\mathcal{H} - g_c \left[ \frac{d}{l_d} - l_d \Delta d \right] \quad (27)$$

### 217 3. Numerical implementation in finite element framework

218 The time-dependent phase-field evolution is coupled with the displacement (stress) field  
 219 solution. Both problems are here solved by using the classical finite element method. The  
 220 main steps of numerical implementation are outlined in this section.

#### 221 3.1. Weak forms

222 In view of finite element discretization, the governing equations (24) and (27) are trans-  
 223 formed into the corresponding weak forms by using the standard Galerkin procedure with  
 224 the test functions  $\delta \mathbf{u}$  and  $\delta d$  respectively. For instance, one gets the weak form integral for  
 225 the displacement problem as:

$$\int_{\Omega} \boldsymbol{\sigma} : \nabla(\delta \mathbf{u}) dV = \int_{\Omega} \mathbf{b} \cdot \delta \mathbf{u} dV + \int_{\partial\Omega} \mathbf{t} \cdot \delta \mathbf{u} dS \quad (28)$$

226 And that for the phase-field evolution is written as [56]:

$$\int_{\Omega} \left[ (2\mathcal{H} + \frac{g_c}{l_d})d + \eta_d \dot{d} \right] \delta d dV + \int_{\Omega} g_c l_d \nabla d \cdot \nabla(\delta d) dV = \int_{\Omega} 2\mathcal{H} \delta d dV \quad (29)$$

227 *3.2. Time discretization*

228 The total loading history (designed by the time  $T$ ) is divided into a number of incremental  
 229 steps such as  $t = 0, t_1, t_2, \dots, t_n, t_{n+1}, \dots, T$ . For the current loading step  $n + 1$ , the time  
 230 increment is denoted as  $\Delta t = t_{n+1} - t_n$ . For the sake of simplicity, the evolution rate of  $d$  is  
 231 assumed to be constant in the time increment under consideration, and it is calculated by:

$$\dot{d} = \frac{d_{n+1} - d_n}{\Delta t} \quad (30)$$

232 where  $d_n$  and  $d_{n+1}$  represent the damage values at the loading steps  $t_n$  and  $t_{n+1}$ , respectively.

233 Further, starting from the initial state, the value at the end of previous step  $t_n$  is known,  
 234 the current viscoplastic strain is calculated by:

$$\boldsymbol{\varepsilon}_{n+1}^{vp} = \boldsymbol{\varepsilon}_n^{vp} + \int_{t_n}^{t_{n+1}} \dot{\boldsymbol{\varepsilon}}^{vp}(\tau) d\tau \quad (31)$$

235 where  $\boldsymbol{\varepsilon}_n^{vp}$  and  $\boldsymbol{\varepsilon}_{n+1}^{vp}$  are the accumulated viscoplastic strains at the time steps  $t_n$  and  $t_{n+1}$   
 236 respectively. Again for the simplicity, the viscoplastic strain rate is considered to be con-  
 237 stant and determined by the state at the loading step  $t_n$  with Equation (11). Thus, the  
 238 accumulated viscoplastic strain for the current step is given by:

$$\boldsymbol{\varepsilon}_{n+1}^{vp} = \boldsymbol{\varepsilon}_n^{vp} + \Delta t \dot{\boldsymbol{\varepsilon}}^{vp}(t_n) \quad (32)$$

239 It is worth noticing that due to the simplified explicit time integration scheme adopted  
 240 here, a small time increment should be chosen for a stable numerical solution. In general,  
 241 preliminary calculations are performed in order to choose a suitable time increment.

242 With the time discretization scheme presented above, the weak form integrals for the  
 243 current loading step  $t_{n+1}$  are rewritten as:

$$\int_{\Omega} \{ \mathbb{C}(d_{n+1}) : [\boldsymbol{\varepsilon}_{n+1} - \boldsymbol{\varepsilon}_{n+1}^{vp}] \} : \nabla(\delta \mathbf{u}) dV = \int_{\Omega} \mathbf{b} \cdot \delta \mathbf{u} dV + \int_{\partial\Omega} \mathbf{t} \cdot \delta \mathbf{u} dS \quad (33)$$

244 and

$$\int_{\Omega} \left[ (2\mathcal{H}_{n+1} + \frac{g_c}{l_d}) d_{n+1} + \eta_d \frac{d_{n+1} - d_n}{\Delta t} \right] \delta d dV + \int_{\Omega} g_c l_d \nabla d_{n+1} \cdot \nabla(\delta d) dV = \int_{\Omega} 2\mathcal{H}_{n+1} \delta d dV \quad (34)$$

245 It is noticed that  $H_{n+1}$  is calculated from the values of  $\boldsymbol{\varepsilon}_{n+1}$  and  $\boldsymbol{\varepsilon}_{n+1}^{vp}$ .

246 *3.3. Geometrical discretization*

247 The computational structure domain  $\Omega$  is divided into  $n_e$  elements which are defined  
 248 by  $n_p$  nodes. The target here is to determine, at each loading step, the nodal values of  
 249 displacement components denoted by the vector  $\mathbf{U}$  and those of phase-field variable by  $\mathbf{d}$ .  
 250 Classically, the displacement and damage fields are approximated in terms of their nodal  
 251 values by using suitable shape functions. Their gradient functions are accordingly calculated:

$$\begin{aligned} \mathbf{u}(\mathbf{x}) &= \mathbf{N}_u(\mathbf{x})\mathbf{U}^e & d(\mathbf{x}) &= \mathbf{N}_d(\mathbf{x})\mathbf{d}^e \\ \boldsymbol{\varepsilon}(\mathbf{x}) &= \mathbf{B}_u(\mathbf{x})\mathbf{U}^e & \nabla d(\mathbf{x}) &= \mathbf{B}_d(\mathbf{x})\mathbf{d}^e \end{aligned} \quad (35)$$

252 where  $\mathbf{N}_u(\mathbf{x})$ ,  $\mathbf{B}_u(\mathbf{x})$  and  $\mathbf{N}_d(\mathbf{x})$ ,  $\mathbf{B}_d(\mathbf{x})$  are the matrices of shape functions and their  
 253 derivatives for the displacement and damage fields, respectively. For the sake of simplicity,  
 254 the same shape functions are usually used for both fields and their test ones. More details can  
 255 be found in [67, 56, 71]. With the geometrical discretization and elementary approximation  
 256 presented above, the weak form integrals (33) and (34) are transformed into two systems of  
 257 discrete equations:

$$\begin{cases} \mathbf{K}_u \mathbf{U}_{n+1} = \mathbf{F}_{u,n+1}^{ext} + \mathbf{F}_{n+1}^{vp} \\ \mathbf{K}_d \mathbf{d}_{n+1} = \mathbf{F}_{d,n+1} \end{cases} \quad (36)$$

258  $\mathbf{K}_u$  and  $\mathbf{K}_d$  are the global stiffness matrices,  $\mathbf{F}_{u,n+1}^{ext}$ ,  $\mathbf{F}_{d,n+1}$  the global nodal forces vectors,  
 259 respectively for the displacement and damage fields, and  $\mathbf{F}_{n+1}^{vp}$  the global nodal force vector  
 260 related to viscoplastic flow:

$$\begin{cases} \mathbf{K}_u = \int_{\Omega} \left[ \mathbf{B}_u^T \mathbb{C}(d_{n+1}) \mathbf{B}_u \right] dV \\ \mathbf{F}_{u,n+1}^{ext} = \int_{\Omega} [\mathbf{N}_u^T \cdot \mathbf{b}] dV + \int_{\partial\Omega} [\mathbf{N}_u^T \cdot \mathbf{t}] dS \\ \mathbf{F}_{n+1}^{vp} = \int_{\Omega} \left[ \mathbf{B}_u^T \mathbb{C}(d_{n+1}) : \boldsymbol{\varepsilon}_{n+1}^{vp} \right] dV \end{cases} \quad (37)$$

261 and

$$\begin{cases} \mathbf{K}_d = \int_{\Omega} \left[ \left( \frac{g_c}{l_d} + 2\mathcal{H}_{n+1} + \frac{\eta_d}{\Delta t} \right) \mathbf{N}_d^T \mathbf{N}_d + g_c l_d \mathbf{B}_d^T \mathbf{B}_d \right] dV \\ \mathbf{F}_{d,n+1} = \int_{\Omega} \left( 2\mathcal{H}_{n+1} + \frac{\eta_d}{\Delta t} d_n \right) \mathbf{N}_d^T dV \end{cases} \quad (38)$$

262 *3.4. Staged coupling algorithm*

263 It is worth noticing that the displacement (and stress) and cracking evolution problems  
 264 are strongly coupled. Indeed, the displacement (stress) evolution is affected by the damage  
 265 process. For instance, the elastic stiffness tensor and viscoplastic threshold are functions of  
 266 damage variable. Inversely, the damage evolution is driven by the mechanical field through  
 267 the stored elastic and viscoplastic energy. Therefore, both problems should be solved in a  
 268 coupled way. However, it is demonstrated that the energy history functional  $\mathcal{H}(t)$  given in  
 269 (38) is non-convex with respect to the couple of unknowns  $(\mathbf{u}, d)$  [44, 66]. As a consequence,  
 270 it is difficult to solve the systems of coupled equations (36) by using the conventional Newton-  
 271 Raphson procedure. However, the functional  $\mathcal{H}(t)$  is convex with respect to one of two  
 272 unknowns  $(\mathbf{u}$  or  $d)$ , if the other one is fixed. In general, the so-called alternating minimization  
 273 (AM) solver proposed in [44, 66] is widely used and adopted here. In the AM solver, two  
 274 solution stages are successively performed at the  $k^{th}$  iteration of a specific loading step  $n+1$ .  
 275 This is summarized as follows:

*Stage-1:* Solving the displacement problem with the damage field fixed at  $\mathbf{d}_{n+1}^{k-1}$  obtained from  
 the last iteration:

$$\mathbf{U}_{n+1}^k = [\mathbf{K}_u^{k-1}(\mathbb{C})]^{-1} \cdot [\mathbf{F}_{u,n+1}^{ext} + \mathbf{F}_{n+1}^{vp}] , \text{ with } \mathbb{C} = \mathbb{C}(\mathbf{d}_{n+1}^{k-1})$$

276

*Stage-2:* Solving the phase-field problem using the updated displacement field and history en-  
 ergy functional:

$$\mathbf{d}_{n+1}^k = [\mathbf{K}_d^k]^{-1} \cdot \mathbf{F}_{d,n+1}^k(\mathcal{H}) , \text{ with } \mathcal{H}_{n+1}^k = \mathcal{H}(\mathbf{U}_{n+1}^k)$$

277

278 These two stages are repeated until the convergence criterion  $\|\mathbf{d}_{n+1}^k - \mathbf{d}_{n+1}^{k-1}\| \leq \epsilon$  is verified.  
 279 According to [38], a convergence tolerance is taken as  $\epsilon = 1 \times 10^{-5}$ .

## 280 4. Simulation of laboratory tests on Jinping marble

281 In view of assessing the efficiency of time-dependent phase-field model, representative  
282 laboratory tests are investigated in this section. It is worth noticing that the emphasis of  
283 phase-field model is to capture the progressive damage toward cracking process. For the  
284 sake of simplicity, instantaneous plastic deformation is not taken into account. Inelastic  
285 strains, failure and post-failure softening are entirely attributed to damage evolution and  
286 localization as well as viscoplastic flow. Therefore, the proposed phase-field model is mainly  
287 suitable for brittle rocks under low confining pressure. For this reason, a typical brittle rock,  
288 Jinping marble, is here selected. This rock has been widely investigated in the context of  
289 stability analysis of high slopes around the Jinping I hydraulic power plant.

290 Short term triaxial compression tests and long term creep tests are considered. Repre-  
291 sentative experimental data are selected from previous studies [12, 13]. The laboratory tests  
292 were performed on cylindrical samples of 50 mm in diameter and 100 mm in height. For the  
293 sake of simplicity, two-dimensional plane strain simulations are performed. The specimen  
294 geometrical domain is divided into  $200 \times 400$  (80 000) elements. According to experimental  
295 observations [12, 13], failure patterns of samples are mainly characterized by localized frac-  
296 tures even if the samples are subjected to uniform macroscopic stresses and displacements.  
297 The onset of localized fractures or non-uniform stress and strain fields is generally related  
298 to material heterogeneity. Without detailed descriptions of such heterogeneity of tested  
299 samples and in order to enhance the onset of localized fractures, a small weak region (one  
300 element) is placed at the sample center where the value of  $g_c$  is reduced by 1%. The bound-  
301 ary conditions are composed of a uniform normal stress on the lateral surfaces (denoted as  
302  $\sigma_3$ ) and a uniform vertical displacement on the upper boundary surface, which is used to  
303 calculate the average axial strain  $\epsilon_1$ . As output results, one obtains the average lateral strain  
304  $\epsilon_3$  and axial stress  $\sigma_1$ .

### 305 4.1. Short term triaxial tests

306 For modeling of short term mechanical behavior, two elastic parameters and the critical  
307 energy release rate  $g_c$  should be identified. The elastic parameters of Jinping marble are

308 directly taken from the previous experimental studies [12] and the following values are used:  
 309 Young's modulus  $E = 45000 \text{ MPa}$  and Poisson's ratio  $\nu = 0.15$ . The critical fracture  
 310 energy (or energy release rate)  $g_c$  is a key parameter in the phase-field model. Its value  
 311 is generally identified from uniaxial tension or bending tests, which are not available for  
 312 the studied rock. Further, it is found that the value of  $g_c$  identified from tensile test could  
 313 not be suitable for cracking under compressive stresses [50, 51]. However, the value of  $g_c$   
 314 can be indirectly estimated from the peak values of deviatoric stress obtained in triaxial  
 315 compression tests. Moreover, it is found that for most rocks, the values of  $g_c$  can vary with  
 316 confining stress. Inspired by the previous studies [50, 51], the following empirical relation  
 317 is here adopted:  $g_c = g_{c0} e^{b(p_c/p_r)}$ , with  $p_c$  being the confining stress and  $p_r = 1 \text{ MPa}$  the  
 318 reference normalizing pressure.  $g_{c0}$  denotes the value of  $g_c$  for uniaxial compression with  
 319  $p_c = 0$  and the parameter  $b$  controls its variation with confining stress. For the case of  
 320 Jinping marble studied here, by using the peak deviatoric stresses obtained in three triaxial  
 321 compression tests respectively with 0, 10 and 20 MPa confining stress, we have obtained  
 322  $g_{c0} = 2.0937 \text{ N/mm}$  and  $b = 0.0466$ . Finally, the length scale parameter  $l_d$  is generally taken  
 323 as 1 to 3 times of the smallest element size. It is set here to 0.25 mm.

324 Comparisons between experimental data and simulation results are presented in Figure  
 325 1 for three values of confining pressure. The main features of mechanical behavior of Jinping  
 326 marble are correctly reproduced by the relatively simple phase-field model. The values of  
 327 peak deviatoric stress are well predicted. For low confining pressures (0 and 10 MPa), the  
 328 post-peak brittle failure process is also properly described. For the test with a high confining  
 329 pressure (20 MPa), it seems that the marble exhibits a brittle-ductile transition. Quite large  
 330 plastic strains, in particular in the radial direction, are observed. There is a residual strength  
 331 in the post-peak regime due to the frictional effect along closed cracks. All these features  
 332 are not properly captured by the simple elastic-brittle model as it is not the emphasis of the  
 333 present work. However, these features can be improved by incorporating an instantaneous  
 334 plastic deformation mechanism in the phase-field model or/and taking into account the  
 335 friction effect along localized cracks. This will be performed in our future studies.

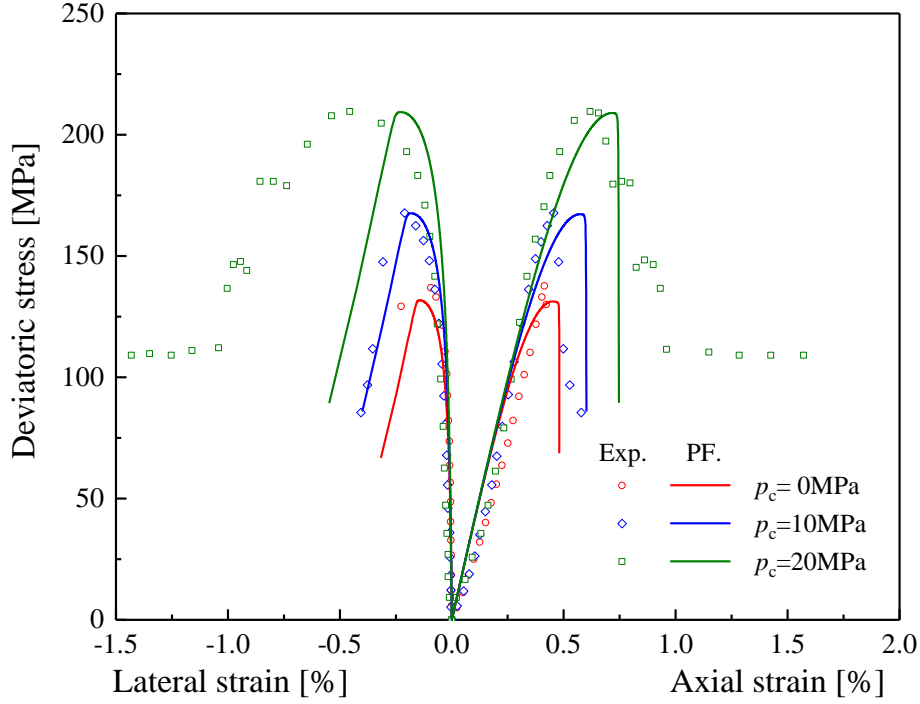


Figure 1: Axial and radial strains versus deviatoric stress of Jinping marble in triaxial compression tests with different pressures: comparisons between numerical results and experimental data [12]

#### 336 4.2. Uniaxial and triaxial creep tests

337 At first, an uniaxial compression creep test reported in [12] is considered. Five additional  
338 parameters involved in the viscoplastic model should be identified, say  $\eta$ ,  $\sigma_s$ ,  $c_v$ ,  $n$  and  $m$ .  
339 These parameters cannot be directly calculated from experimental measurements and they  
340 are generally calibrated from the optimal numerical fitting of creep tests. Basically, the  
341 parameter  $\sigma_s$  defines the threshold of viscoplastic flow and can be determined at the end  
342 of elastic stage in a triaxial creep test.  $\eta$  mainly controls the initial creep rate during  
343 the primary creep stage while  $n$  affects the creep rate evolution.  $m$  describe the viscoplastic  
344 hardening effect on the transition from the primary to secondly creep stages. The viscoplastic  
345 potential parameter  $c_v$  controls the volumetric viscoplastic strain and can be identified from  
346 the lateral creep strain curve. For the marble samples tested in [12], the following values of  
347 viscoplastic parameters are obtained:  $\eta = 1 \times 10^{-7} s^{-1}$ ,  $\sigma_s = 100$  MPa,  $c_v = 0.01$ ,  $n = 3.2$   
348 and  $m = 7980$ . In Figure 2, we show the axial and radial strains as functions of time under

349 an axial stress of 113 MPa. The evolution of damage is also reported. It is found that  
 350 the numerical predictions obtained using the elastic-viscoplastic and phase-field model are  
 351 in good agreement with the experimental data. The main features of creep deformation  
 352 behavior of Jinping marble are well described by the proposed model. In particular, typical  
 353 three stages of creep deformation are found in this test, namely an increasing primary creep  
 354 stage, an quasi-stationary stage and an accelerating one leading to the sample failure. It is  
 355 further to observe the creep deformation rate is clearly linked to the damage evolution which  
 356 also exhibits three distinct stages. During the first stage, the damage increases quickly due  
 357 to the applied axial stress. Then it evolves slowly with the viscoplastic strain. Finally, the  
 358 maximum value in the sample increases very rapidly and reaches 1 causing the macroscopic  
 359 failure of sample.

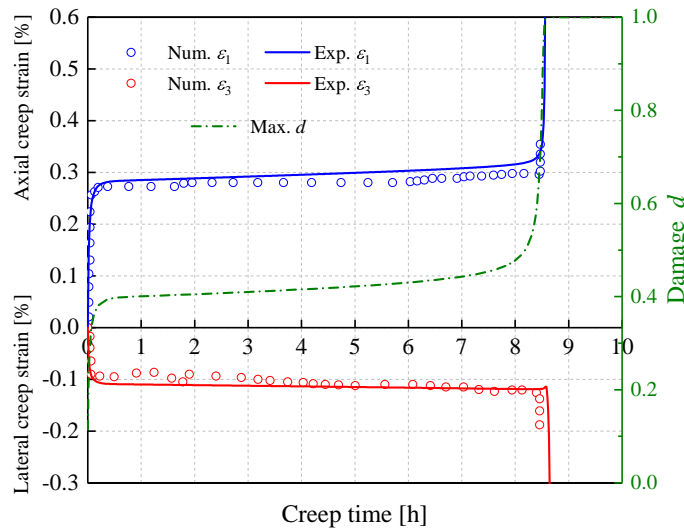


Figure 2: Evolution of axial and lateral strains as well as damage variable with time in an uniaxial compression creep test under an axial stress of 113 MPa: comparisons between numerical results and experimental data [12]

360 In Figure 3, we show the damage variable distributions inside the sample obtained at  
 361 different instances during the uniaxial compression creep test. Due to the presence of weak  
 362 element, the damage localization starts from the center of sample. Two quasi symmetric  
 363 localized damage bands are obtained. This configuration is close to fracture modes widely

364 observed in rock samples [12]. It is worth noticing that the use of a weak element is an  
 365 artificial technique to facilitate the onset of damage localization. In future studies, this  
 366 point can be addressed by considering the spatial variability of micro-structure of rocks such  
 367 as porosity and mineral composition.

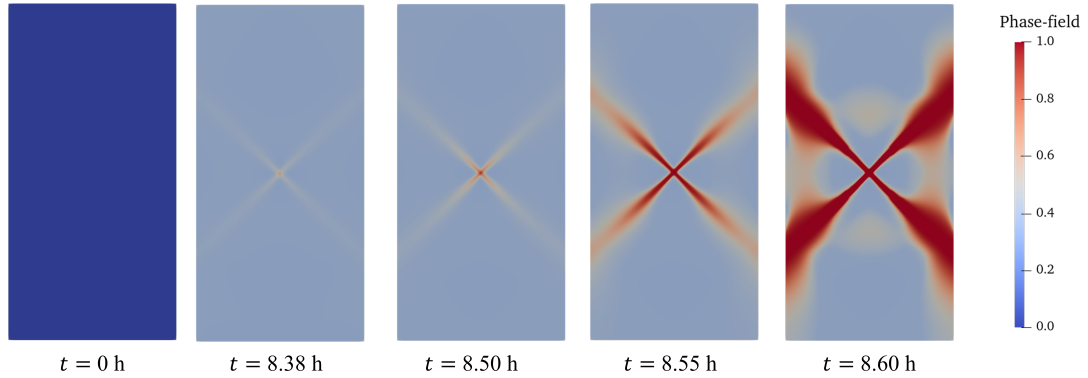


Figure 3: Evolution of damage distribution of uniaxial compression creep test

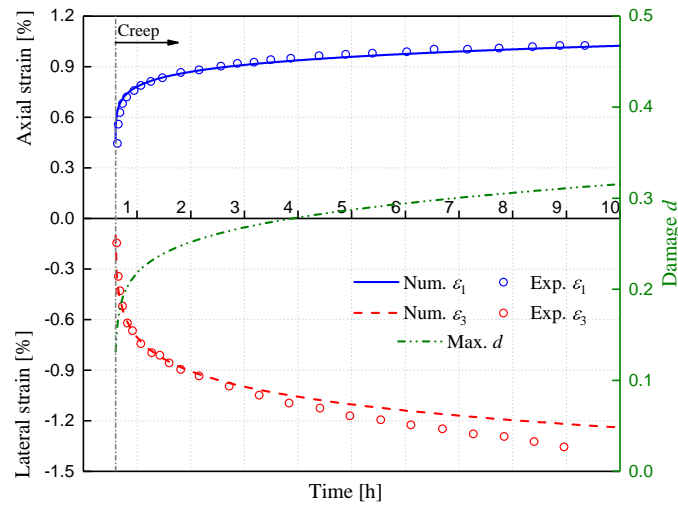
368 Moreover, two triaxial compression creep tests under a confining pressure of 30MPa,  
 369 also performed on Jinping marble respectively with a single and multiple deviatoric stress  
 370 steps, are investigated. The experimental data are taken from [13]. The samples used  
 371 in these tests and those used in [12] were drilled at different zones. Therefore, there are  
 372 some quantitative differences on mechanical parameters. For instance, the elastic properties  
 373 measured in triaxial compression tests [13] are as follow: Young's modulus  $E = 30000 MPa$   
 374 and Poisson's ratio  $\nu = 0.15$ . The critical fracture energy  $g_c$  is again fitted from the peak  
 375 deviatoric stress and one gets  $g_c = 1.2775 N/mm$ . The parameters for viscoplastic flow are  
 376 as follows:  $\eta_d = 1 \times 10^{-10} Ns/mm^2$ ,  $\eta = 1 \times 10^{-7} s^{-1}$ ,  $\sigma_s = 70MPa$ ,  $c_v = 0.26$ ,  $n = 2.4$  and  
 377  $m = 1080$ . However, the same finite element mesh and the same value of length scale  $l_d$  as  
 378 those in the uniaxial creep test are used.

Table 1: Reference set of model's parameters for rock samples investigated in [13]

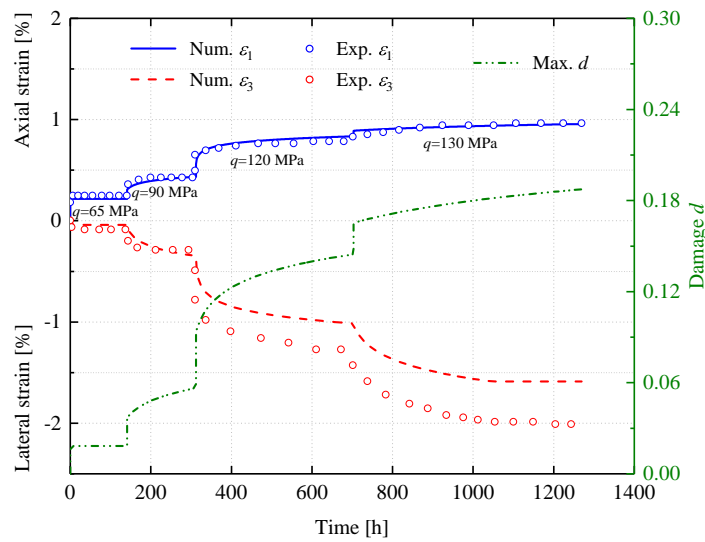
Elastic and phase-field parameters	$E$ ( <i>MPa</i> )	$\nu$ (-)	$g_c$ ( <i>N/mm</i> )	$p_c$ ( <i>MPa</i> )	$l_d$ ( <i>mm</i> )
Values	30000	0.15	1.2775	30	0.25
Viscoplastic parameters	$\eta$ ( $s^{-1}$ )	$\sigma_s$ ( <i>MPa</i> )	$n$ (-)	$m$ (-)	$c_v$ (-)
Values	$1 \times 10^{-7}$	70	2.4	1080	0.26

379 The creep test with a single loading step was performed under a deviatoric stress of  
 380 143 MPa. Numerical predictions are compared with experimental data in Figure 4(a),  
 381 together with the damage variable evolution. A good concordance is obtained. In this test,  
 382 one obtains mainly a primary creep phase. Both axial and radial strains evolve towards  
 383 stationary values and creep strain rates decrease with time. Another important feature  
 384 is that the radial creep strain is clearly larger than the axial one. This means that the  
 385 viscoplastic deformation induces an important volumetric dilatance. In the second test, four  
 386 loading steps were realized with increasing values of deviatoric stress from 65 MPa to 130  
 387 MPa. Comparisons between numerical results and experimental data of axial and radial  
 388 strains are shown in Figure 4(b). It is interesting to observe that the creep strain rate  
 389 increases with the rise of deviatoric stress. When this one is high, a small increment of  
 390 deviatoric stress can generate large creep strains, for instance during the last loading step  
 391 with an increment of 10 MPa. The damage variable evolution trend is fully consistent with  
 392 that of strains. Again, the radial creep strain is clearly larger than the axial one. There is a  
 393 good agreement between numerical and experimental results for the axial strain as well as  
 394 for the radial one when the deviatoric stress is low (the first two loading steps). But more  
 395 and more scatters are observed when the deviatoric stress is high (the last two steps). This  
 396 indicates that the volumetric dilation is enhanced during creep process by material damage.

397 As a possible improvement of the viscoplastic model, the volumetric dilation parameter  $c_v$   
 398 should evolve with damage.



(a) creep test with a single loading step ( $q = 143\text{MPa}$ )



(b) creep test with multiple loading steps

Figure 4: Variations of axial and lateral strains in triaxial compression creep tests under a confining pressure of 30 MPa respectively with a single and multiple loading steps, together with damage variable evolution

400 *4.3. Sensitive analysis*

401 Some comparative calculations are presented here in order to investigate effects of some  
 402 key parameters on short and long-term mechanical responses of rocks.

403 Firstly, the influences of geometry discretization as well as the characteristic length are  
 404 studied. For this purpose, six cases with different minimum element size  $h$  and length scale  
 405  $l_d$  are considered and detailed in Table 2. These case are chosen for the following strategy.  
 406 In the cases 1 to 4, the characteristic length is equal to or higher than the minimum element  
 407 size,  $l_d \geq h$ , while in the cases 5 and 6, it is smaller than the minimum element size  
 408  $l_d < h$ . Moreover, in order to facilitate the comparison and interpretation, the value of  $g_c$   
 409 is accordingly reset with that of the length scale  $l_d$ . The objective is to obtain a very close  
 410 value of uniaxial compression strength for all the cases considered to that reported in [12].

Table 2: Different sets of considered parameters for sensitivity study of uniaxial compression test reported in [12]

Case No.	$l_d$ (mm)	$h$ (mm)	$g_c$ (N/mm)	Case No.	$l_d$ (mm)	$h$ (mm)	$g_c$ (N/mm)
1	1.0	1.0	2.08	4	1.0	0.5	2.08
2	0.5	0.5	1.04	5	0.5	1.0	1.04
3	0.25	0.25	0.52	6	0.25	0.5	0.52

411 In Figure 5, we first show the overall strain-stress curves of uniaxial compression test  
 412 for six cases considered in Table 2. It is confirmed that all six cases give a quasi identical  
 413 uniaxial compression strength. However, the responses are slightly different in the post-  
 414 peak regime. For those cases where  $l_d \geq h$  (cases 1-4), the mechanical behavior is almost  
 415 the same while the curves of the cases where  $l_d < h$  (cases 5, 6) are lower than the other  
 416 cases. However, significant differences are observed in the uniaxial creep test as shown in  
 417 the same figure. Although the similar primary creep stages are obtained, the period of quasi  
 418 constant creep rate for the cases with  $l_d < h$  (cases 5, 6) is clearly shorter than those with  
 419  $l_d \geq h$  (cases 1-4). In other words, the accelerated creep stage occurs earlier. According

420 these results, it is recommended to ensure the condition of  $l_d \geq h$  in order to obtain the  
 421 stable numerical results in both pre- and post-peak regimes.

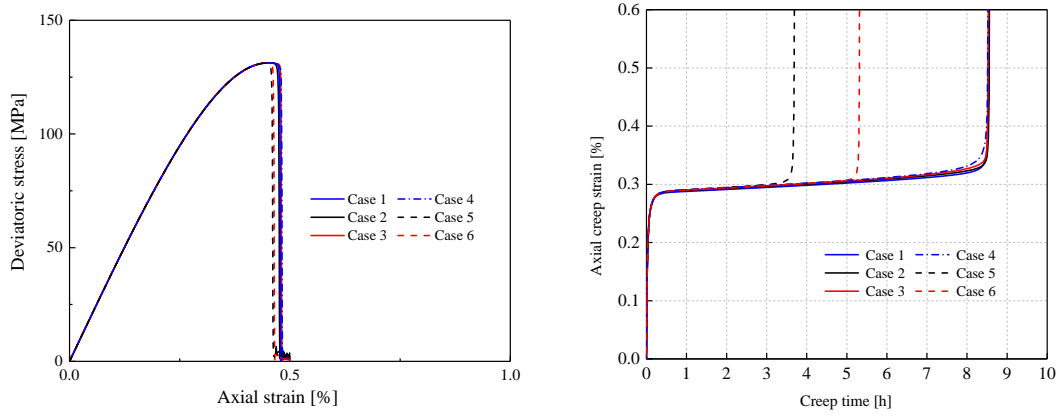


Figure 5: Influence of characteristic length and mesh size on overall strain-stress curve in uniaxial compression test (left) and on strain evolution in uniaxial compression creep test (right)

422 On the other hand, the influence of length scale parameter  $l_d$  and mesh size on cracking  
 423 pattern is also investigated and presented in Figure 6. One can clearly see that it is the  
 424 length parameter  $l_d$  that controls the width of localized cracks (damaged zones) rather than  
 425 the mesh size  $h$ . For instance, both the cases 1 and 4 show a very similar distribution  
 426 of damage variable though they have different element size. At the same time, with the  
 427 decrease of  $l_d$ , the width of localized cracked zones becomes finer from the case 1 to 3.

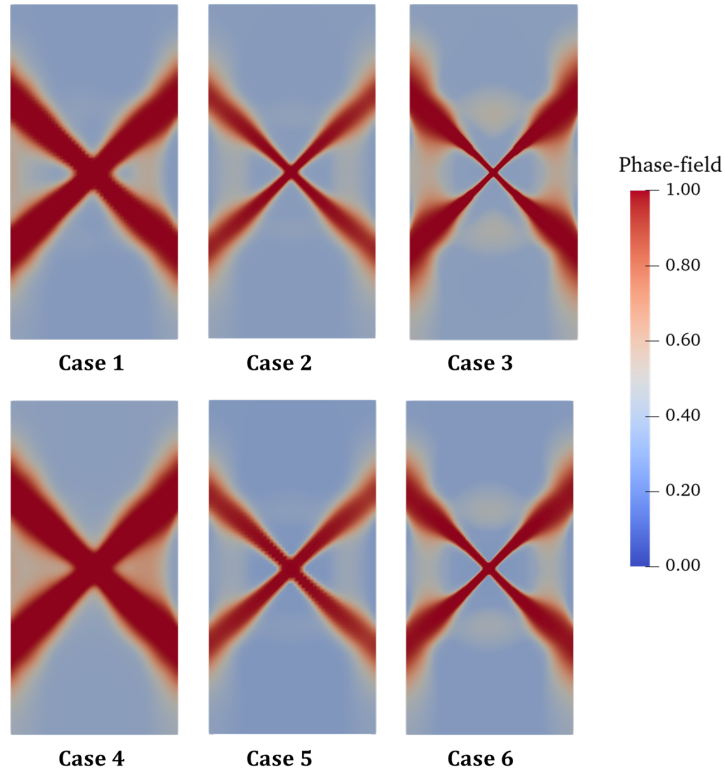


Figure 6: Distribution of damage in uniaxial compression creep test in different cases

428 The coupling effect of viscoplastic flow and damage evolution is considered. Indeed,  
 429 in the proposed viscoplastic model, the viscoplastic threshold is weakened by the damage  
 430 growth though the degradation function  $g_{vp}(d)$  introduced in the loading function  $f^{vp}$  (12).  
 431 Two calculations are performed on the uniaxial creep test by respectively considering that  
 432 the viscoplastic threshold is affected  $g_{vp}(d) = (1 - d)^2$  or not affected  $g_{vp}(d) = 1$  by the  
 433 damage evolution. Comparisons between two cases are presented in Figure 7. One can see  
 434 that in the case without degradation, an primary creep is obtained. However, due to the  
 435 degradation of viscoplastic threshold by damage, three creep stages are observed. The creep  
 436 strains during the primary stage are much larger than those in the case without degradation.  
 437 After a period of quasi constant creep rate, an accelerated creep stage occurs and leads to a  
 438 rapid failure. These results show that the damage degradation effect can play an essential  
 439 role in time-dependent deformation of rocks.

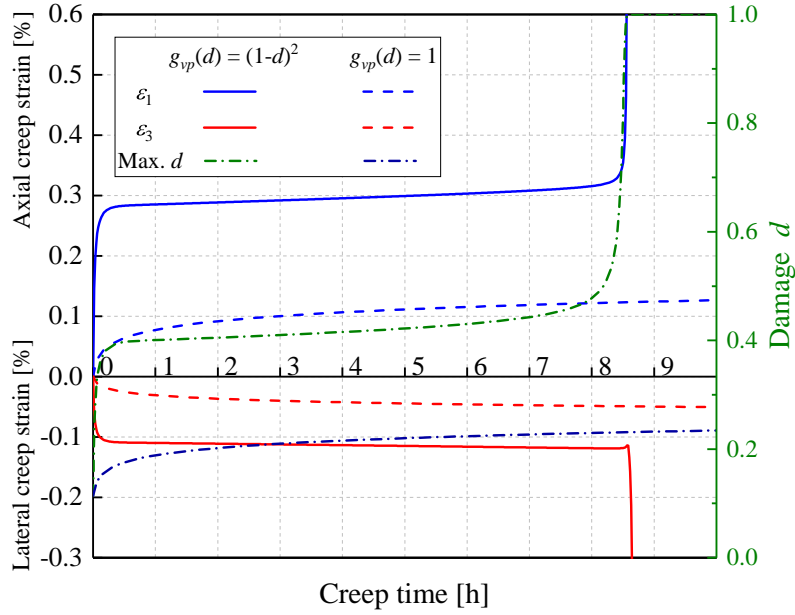


Figure 7: Influence of damage degradation effect on viscoplastic flow in uniaxial creep test

440 In the second example, the influence of critical fracture energy  $g_c$  on creep deforma-  
441 tion is studied. For this purpose, three different values of  $g_c$  ( $1.225N/mm$ ,  $1.2375N/mm$ ,  
442  $1.2775N/mm$ ) are used in simulations of the triaxial creep test with one-step axial loading.  
443 Comparisons between three calculations are presented in Figure 8 in terms of variations of  
444 strains and damage variable. It is obvious that the damage evolution rate is higher when the  
445 value of  $g_c$  is lower. Accordingly, the rate of creep strains is also enhanced by the decrease  
446 of critical fracture energy. In the case with  $g_c = 1.225N/mm$ , it seems that the damage  
447 variable evolves towards a stationary value. In the case with  $g_c = 1.2375N/mm$ , the damage  
448 evolution rate evolves to a constant value. Finally, In the case of  $g_c = 1.2775N/mm$ , the  
449 damage evolution rate is accelerated after a period of time and evolves towards to unit,  
450 producing an accelerated creep phase. There is a close connection between time-dependent  
451 deformation and damage evolution.

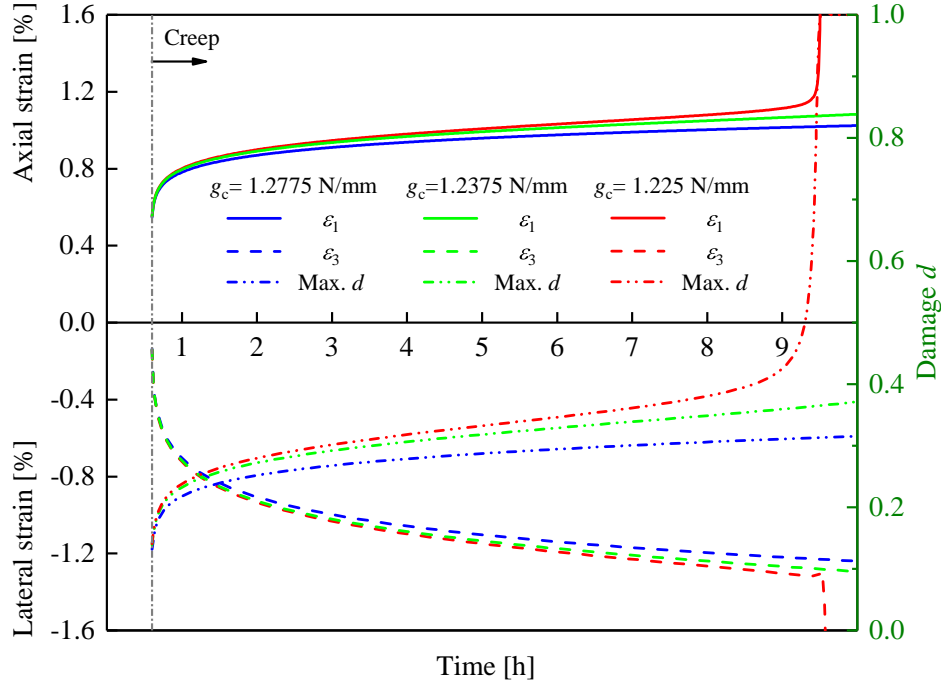
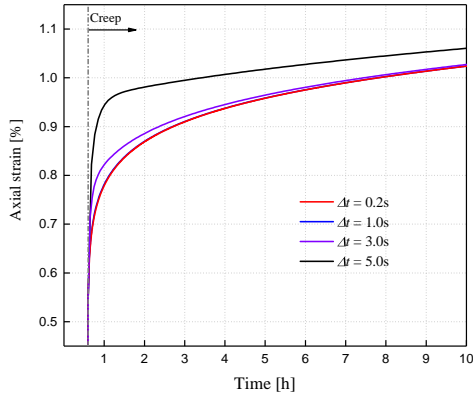
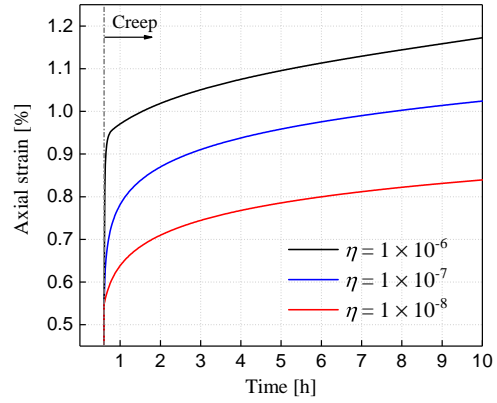


Figure 8: Influences of critical fracture energy  $g_c$  on time-dependent strains and damage evolution in a triaxial compression creep test with a confining pressure of 30 MPa

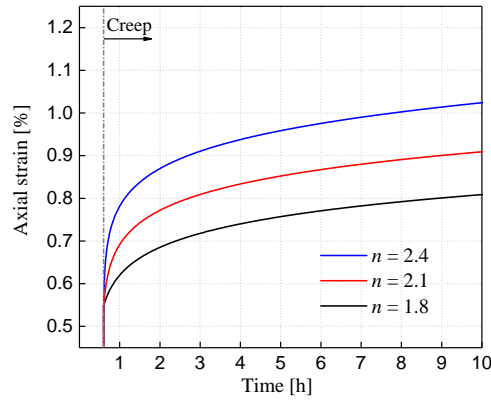
452 Moreover, as a time-dependent viscoplastic phase-field model is used in this study, the  
453 time-discretization scheme can affect the obtained numerical results. For this purpose, the  
454 sensitivity of creep deformation to the time increment size is here investigated. Four different  
455 values of  $\Delta t$  are chosen and compared. The obtained results are presented in Figure 9(a). It  
456 can be seen that the creep strain curves become almost unchanged when the time increment  
457 is less than 1.0s. As a consequent, the this value of  $\Delta t = 1.0s$  is chosen in all calculations  
458 of this section. In addition, the influences of two viscoplastic parameters  $\eta$  and  $n$  on creep  
459 strain are also studied. It is shown that  $\eta$  mainly controls the initial creep rate during the  
460 primary stage. As presented in Figure 9(b), higher the value of parameter  $\eta$  is, bigger the  
461 slope of primary creep curve is. The parameter  $n$  affects the creep rate evolution. On other  
462 word, the creep rate reduces quickly when a small value of  $n$  is used, as shown in Figure  
463 9(c).



(a) Effect of  $\Delta t$



(b) Effect of  $\eta$



(c) Effect of  $n$

Figure 9: Sensitive analysis of time increment and two viscoplastic parameters in a triaxial compression creep test with a confining pressure of 30 MPa

464 **5. Time-dependent behavior of left bank high slope of Jinping I hydropower**  
 465 **station**

466 As an example of engineering application, time-dependent mechanical responses of the  
 467 left bank high slope of Jinping I hydropower plant are not investigated by using the proposed  
 468 phase-field model.

469 *5.1. Summary of engineering geological conditions*

470 The Jinping I hydropower station, one of the highest and largest ones all over the world,  
 471 is built at the Pusiluogou in the west of Great River Bend of Yalong River which is adjacent  
 472 to Muli and Yanyuan counties in Liangshan Yi Autonomous Region. It is near Xichang

473 (27°32'-28°10'N, 101°46'-102°25'E), about 500 km southwest of Chengdu, Sichuan Province,  
474 PR China. The double-curvature arch dam, of 305  $m$  in height with a crest at 1885  $m$  above  
475 the sea level (asl), is constructed in a typical V-shaped valley [72]. Due to the height of the  
476 dam, the retained reservoir level will reach 1880  $m$  asl, with a total reservoir capacity 7.76  
477 billion  $m^3$  and the regulated storage 4.91 billion  $m^3$ .

478 The valley is cut through the Triassic Zagunao Formation which consists of sandstone,  
479 slate and marble. In view of its location between the Tibetan Plateau and Sichuan Basin, the  
480 area is characterized by complex geological conditions [73], especially on the left bank of the  
481 valley. A typical geological profile (L9-L9) of the left bank was reported in [17]. Accordingly  
482 the slope rises at an angle between 70°-80° below 1900  $m$  asl and 40°-50° above. In addition,  
483 the bedrock of slope is composed of metamorphic rocks belonging to the Zagunao group of  
484 the middle and upper Triassic ( $T_{2-3Z}$ ) which can be divided into three members [74]: a green  
485 schist member ( $T_{2-3Z}^1$ ), a marble member ( $T_{2-3Z}^2$ ) and a slate and low-grade metamorphic  
486 sandstone member ( $T_{2-3Z}^3$ ). Moreover, a large number of faults and fractures are observed  
487 in this area [74], for instance, the fault referred as  $f_2$  is shown in Figure 10(a). To sum up,  
488 the geological conditions at the dam site are extremely complex and request an in-depth  
489 investigation of stability conditions of the left abutment slope. In this context, the present  
490 study shall provide a numerical analysis of time-dependent deformation and damage by  
491 using an efficient phase-field method.

## 492 5.2. Description of numerical model

493 Here we consider the typical two-dimensional section L9-L9 of the left bank high slope of  
494 Jinping I hydro-power station. In Figure 10(a), the geometrical domain and main geological  
495 layers of the section are presented. The studied domain is of 370  $m$  in width and of 450  $m$   
496 in height. The slope section composition is simplified to be composed of three principal  
497 rock layers respectively referred to as II, III and IV, and a major fault zone named as  $f_2$ .  
498 Based on the studies related to the laboratory tests presented above and field investigations  
499 [73, 17], the basic model's parameters of each layers are given in Table 3. As the fault zone  
500  $f_2$  is mainly constituted with filled materials, it is here considered as a thin layer with a

501 small critical fracture energy  $g_c$ . In Figure 10(b), one also shows the finite element mesh  
 502 adopted in this study. It is composed of 12523 elements and 12729 nodes in total. The  
 503 length scale parameter for the regularized crack description is taken as  $l_d = 0.25 m$ , which  
 504 is equal to the minimum size of element.

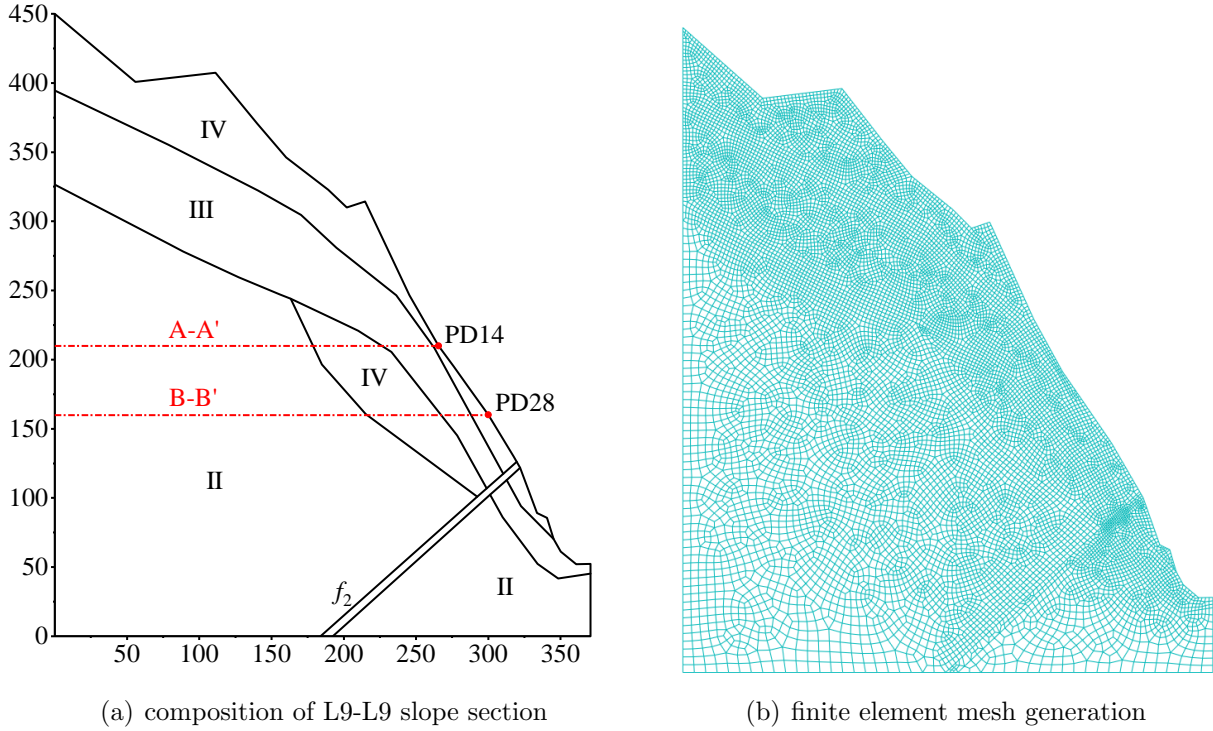


Figure 10: Geometrical domain of a selected slope section and finite element mesh generation

Table 3: Reference set of physical, elastic and phase-field parameters of different rock layers

	Rock layers			
	Layer II	Layer III	Layer IV	Fault $f_2$
Density $\rho$ ( $kg/m^3$ )	2800	2700	2700	2600
Young's modulus $E$ ( $MPa$ )	25000	8500	2500	3000
Poisson's ratio $\nu$	0.23	0.26	0.3	0.28
Critical energy $g_c$ ( $N/mm$ )	900	104	28.3	1.08

Table 4: Reference set of viscoplastic parameters

Parameters	$\eta$ ( $s^{-1}$ )	$\sigma_s$ ( $MPa$ )	$n$ (-)	$m$ (-)	$c_v$ (-)
Values	$1 \times 10^{-9}$	5.8	2.8	1030	0.24

505 *5.3. Main results of reference case*

506 Main results obtained from a reference calculation using the parameters given in Tables  
507 3 and 4 are first presented and discussed.

508 First of all, in Figure 11, the distributions of instantaneous displacement and damage  
509 induced by self gravity of slope are given. One can see that the maximum instantaneous  
510 displacement is obtained in the top zone of the slope. The fault layer is highly damaged. A  
511 small but not negligible damage distribution is also found inside the weak zone IV as shown  
512 in the Figure 11(b).

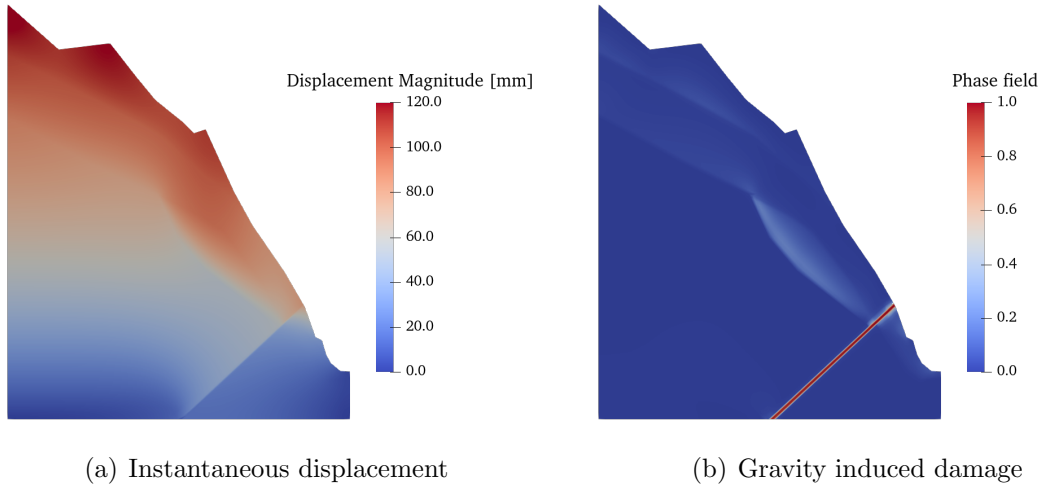
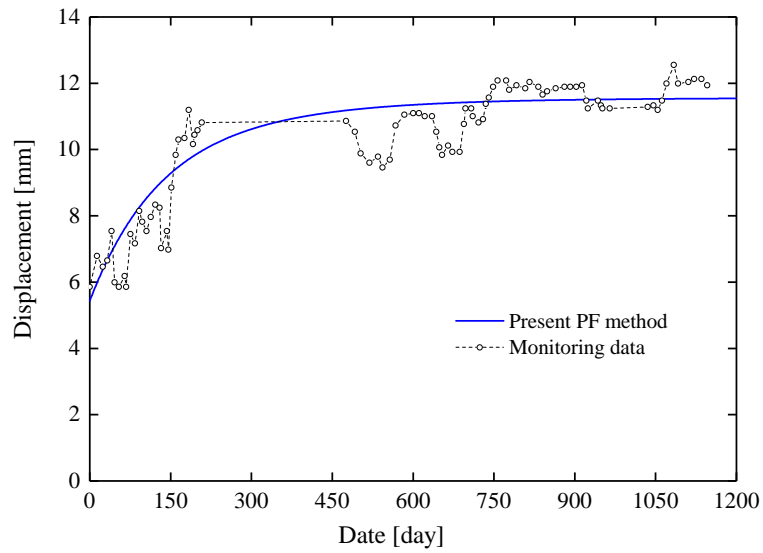


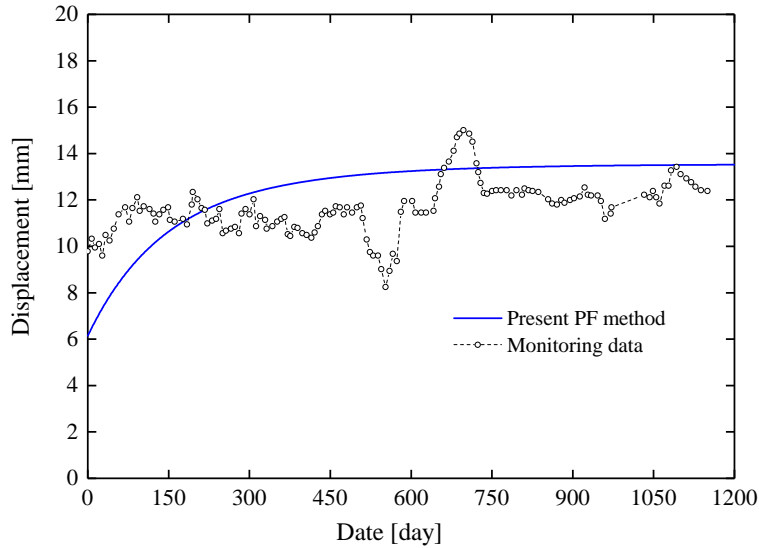
Figure 11: Instantaneous displacement and damage distributions induced by gravity effect inside the slope

513 The time-dependent behavior of the slope is then studied by taking the instantaneous  
514 results presented in Figure 11 as the initial conditions. As the gravity effect is already  
515 taken into account and there is no other applied loads, the long-term stability of the slope  
516 is entirely controlled by the time-dependent mechanical behavior of rock layers. Based on

517 the *in-situ* measured data reported in [17], the time-dependent deformation of the slope  
 518 during a forty month period (from 1<sup>st</sup> January 2010 to 15<sup>th</sup> April 2013) is investigated in  
 519 this work. Indeed, for the purpose of monitoring the deformation evolution of the left bank  
 520 slope with the time, some monitoring points have been set, for instance the points *PD14*  
 521 and *PD28* in the section *L9 – L9* as shown in Figure 10(a). The comparison between the  
 522 measured displacement and numerical prediction is presented in Figure 12. It seems that  
 523 the numerical predicted evolution of displacement at the two monitoring points is in good  
 524 agreement with the measured data.



(a) monitoring point PD14



(b) monitoring point PD28

Figure 12: Comparisons of displacement evolution between measured and calculated values at two monitoring points (From 1<sup>st</sup> January 2010 to 15<sup>th</sup> April 2013)

525 The distribution of displacement and damage increments with respect to the instanta-  
526 neous values at the end of calculation period (1200 days) are shown in Figure 13. Differently  
527 with the instantaneous results shown in Figure 11, the maximum time-dependent displace-  
528 ment increase is obtained inside the weak layer IV and around the fault layer  $f_2$ . At the  
529 same time, the damage distribution inside the weak layer IV exhibits a time-dependent  
530 growth. In order to show some quantitative examples, the variations of damage variable  
531 along two selected cut lines  $A - A'$  and  $B - B'$  defined in Figure 10(a) are also presented  
532 for three representative instances. One can see that the maximum damage value increases  
533 from 0.161 to 0.307 on the line  $A - A'$  and 0.156 to 0.295 on  $B - B'$ . This clearly shows that  
534 the time-dependent damage evolution related to micro-crack growth is the main mechanism  
535 driving the time-dependent deformation of the slope.

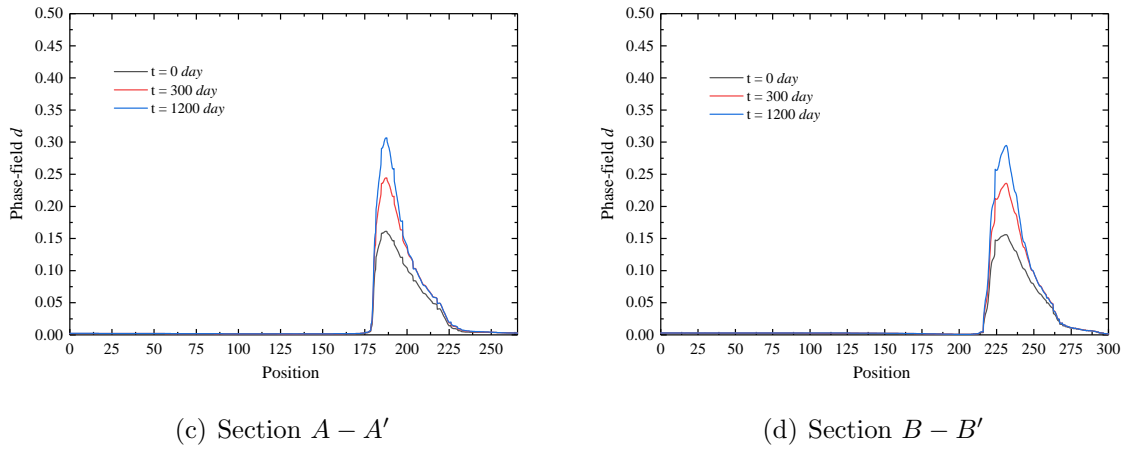
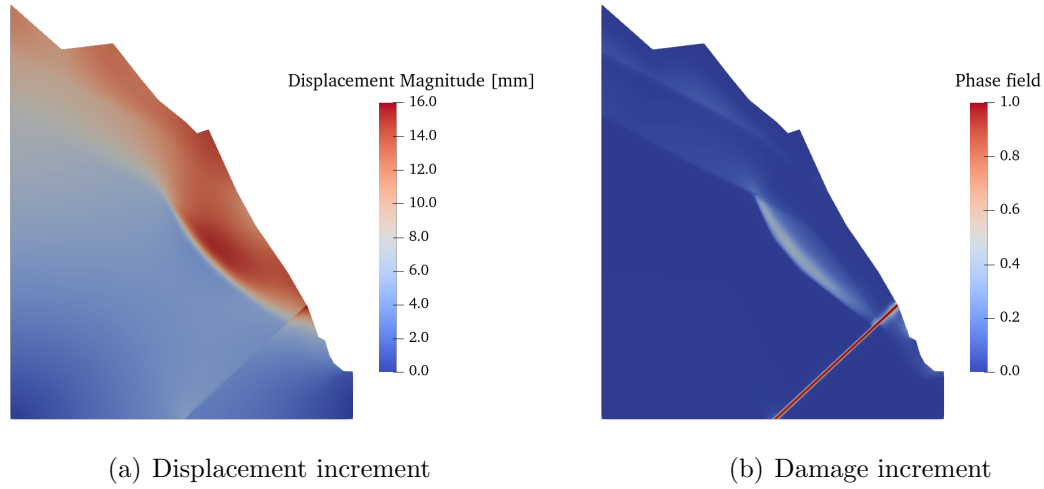


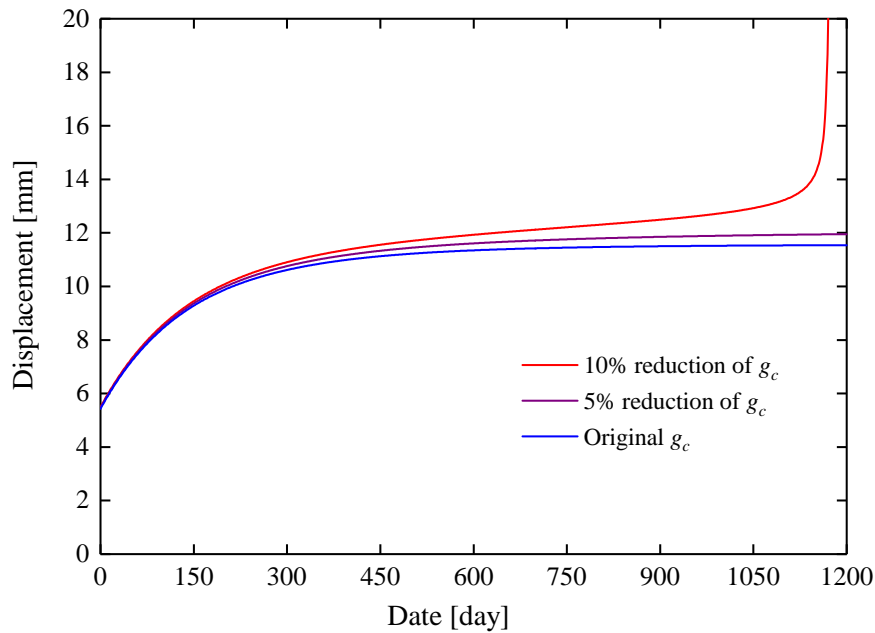
Figure 13: Distributions of displacement and damage increments with respect to the instantaneous values at the end of calculation period (1200 days), and variation of damage along the two cut lines defined in Figure 10(a) at three different instances

#### 5.4. Sensitivity analysis of water weakening effect

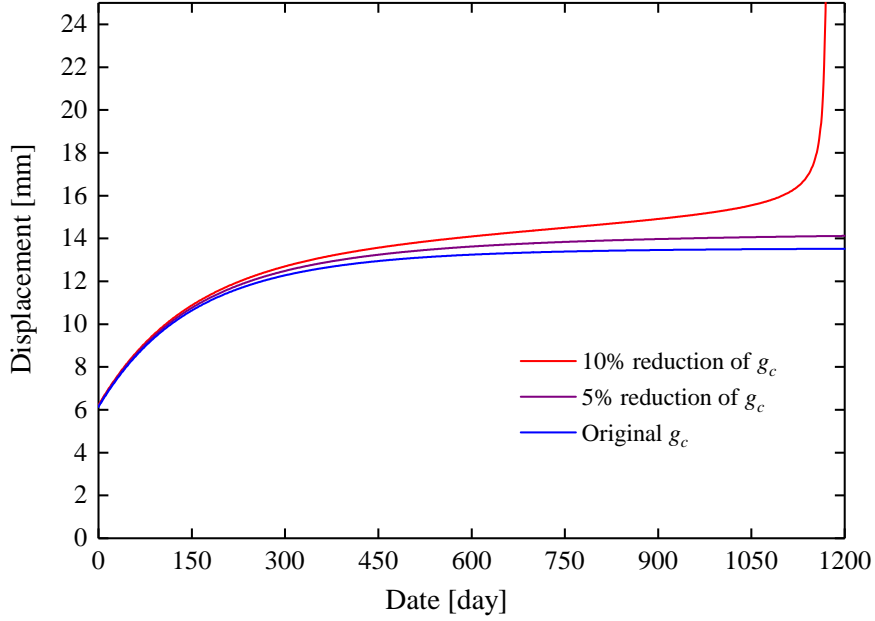
Most slopes are subjected to water saturation change due to raining and reservoir level variation. It is also known that the mechanical properties of most rocks are sensitive to water content. In particular, a number of previous studies have shown that the critical fracture energy or stress intensity factor of rocks could be significantly reduced by an increase of water saturation degree [5, 75, 76]. In the case of the Jinping left bank slope, according to the experimental study reported in [77], the failure strength of all rock layers considered here can be reduced by 10% due to water saturation. Therefore, it is crucial to investigate the consequence of such water weakening effect of the long-term deformation and stability

545 of slopes.

546 However, as a first approximation, a simplified approach is adopted in this work. We  
547 shall not perform fully coupled hydromechanical calculations. The water weakening effect  
548 is taken into account simply by considering that the critical fracture energy  $g_c$  for each  
549 rock layer is reduced by different ratios. The predicted displacement evolution with time at  
550 the two monitoring points are presented in Figures 14 and 15 for two ratios of reduction.  
551 It is clear that the displacement is enhanced by the reduction of  $g_c$ . The amplification  
552 of displacement is quite small for a reduction of 5%. However, with a reduction of  $g_c$  by  
553 10%, the displacements at two monitoring points are significantly amplified and exhibit an  
554 accelerating phase leading to sliding of slope. This seems to indicate that if the critical  
555 fracture energy parameter  $g_c$  is reduced by water saturation to some critical value, it is  
556 possible to generate the instability of slope.



(a) Point PD14



(b) Point PD28

Figure 14: Consequences of critical energy parameter  $g_c$  reduction on displacement evolution at two monitoring points

557 In order to better investigate the instability mechanism induced by the weakening of rock  
558 fracture resistance, in Figure 15, we show the damage variable distribution at six different  
559 time stages. One can see that at the first stage, the damage is mainly concentrated in  
560 the fault layer  $f_2$  and the other zones are weakly affected by cracking process, as shown in  
561 Figure 15(a). Then, the damage zone propagates into the layer  $IV$  due to the increase of  
562 creep strains (see Figure 15(b)). There is further the coalescence of damage between these  
563 two layers causing macroscopic cracking ( $d$  close to 1) around the toe zone of the slope, as  
564 illustrated in Figure 15(c). From the stage  $d$ , the damaged zones propagate towards to the  
565 top region of the slope (see Figure 15(d) and 15(e)). Finally, in Figure 15(f), a large cracked  
566 band is formed and it leads to the instability of the slope.

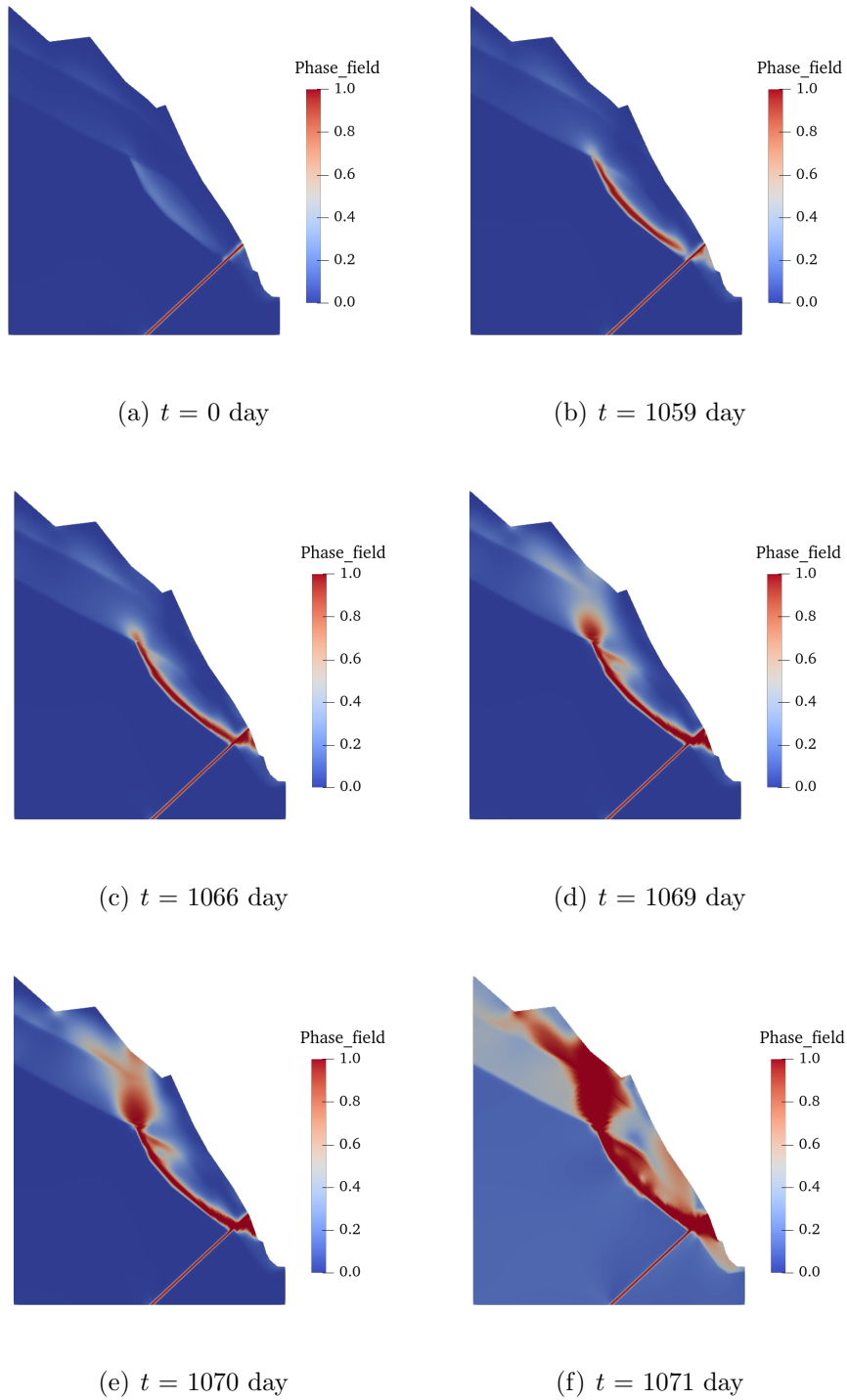


Figure 15: Evolution of damage distribution at six different time stages

## 567 **6. Conclusions**

568 In this paper, we have proposed a new phase-field model for modeling the time-dependent  
569 cracking process of rocks. The viscoplastic deformation is coupled with the evolution of  
570 cracks. More precisely, the viscoplastic flow rate is enhanced by the induced damage through  
571 the weakening of creep threshold while the viscoplastic strain energy contributes to the  
572 damage evolution. The proposed model has been applied to describe the short and long term  
573 mechanical behavior of marble from the left bank of Jinping hydropower station. Numerical  
574 results were compared with experimental data from both laboratory tests and *in – situ*  
575 measurements. It was found that the main features of marble mechanical responses were  
576 correctly described by the proposed model. The phase-field based model provides an efficient  
577 tool for modeling the transition from diffuse damage to localized cracks.

578 From the results obtained from the present study, it can be concluded that the induced  
579 damage is one of the keys mechanisms of brittle rocks such as marble. The macroscopic  
580 failure of those rocks is mainly due to the coalescence of micro-cracks leading to macro-  
581 scopic fractures. The time-dependent evolution of damage or growth of micro-cracks is a  
582 key parameter controlling the long-term stability of structures in such rocks. Moreover,  
583 there is a strong interaction between viscoplastic deformation and damage evolution. It is  
584 crucial to take into account this type of interaction in long-term stability analysis of slopes.  
585 Furthermore, the weakening effect of water saturation on the fracture toughness of rocks is  
586 another key factor to be taken into account. Only a very simple approach has been adopted  
587 in this study. Further investigations should be performed for the in-depth characterization  
588 of the water weakening effect on long-term mechanical properties of those rocks in order to  
589 develop coupled hydromechanical modeling.

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