

1 A new analytical method for stability analysis of rock blocks with 2 ~~cavity~~—cavities in sub-horizontal strata by the considering 3 eccentricity effect

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10 **Abstract.** The basal cavity of a rock block formed due to differential weathering is an important predisposing factor for
11 rockfall; in hard-soft interbedded rocks. ~~The rock~~Rock block falling due to the eccentricity effect with the failure modes of
12 toppling or sliding is defined as biased rockfall in this study. Considering the non-uniform stress distribution due to the
13 eccentricity effect, a new analytical method for three-dimensional stability analysis of biased rockfall is proposed. ~~In~~
14 ~~addition, a~~A set of factors of safety (*Fos*) against partial damage (compressive and tensile damage of the soft underlying
15 layer) and overall failure (toppling and sliding of the hard rock block) are used to determine the rockfall susceptibility level.
16 The analytical method ~~was~~is applied and validated with ~~the~~ biased rockfalls ~~in~~on the northeast edge of the Sichuan ~~b~~Basin
17 in Southwest China, where ~~a~~ large amounts of rockfalls ~~develop~~have developed, composed of overlying thick sandstone and
18 underlying mudstone. The evolution process of biased rockfalls is divided into four stages, initial state, cavity formation,
19 partial unstable and failure. The proposed method is validated by calculating *Fos* of the typical unstable rock blocks in the
20 study area. ~~It is indicated that the~~ The continuous retreat of cavity causes ~~the~~ stress redistribution between the hard and soft
21 rock layers. Consequently, the development of the eccentricity effect leads to ~~the~~ damage ~~of~~to the underlying soft rock layer
22 and ~~the~~ further failure of the hard rock block. The critical cavity retreat ratio is determined ~~as to be~~ 0.33 to classify the low
23 and moderate rockfall susceptibility in the eastern Sichuan Basin. The proposed analytical method is effective for the early
24 identification of biased rockfall, which is significant for rockfall prevention and risk mitigation.

25 List of symbols

26 a	length of the block along the x direction
27 A	area of contact surfaces
28 b	width of the block along the y direction
29 c	cohesive force of the mudstone
30 d_i	width of the basal cavity in a certain direction
31 e_x	eccentric distance along the x direction

32	e_y	eccentric distance along the y direction
33	E_x	horizontal seismic force along <u>the</u> x direction
34	Fos	factor of safety
35	h	height of the block
36	h_w	height of the water in the fracture
37	H_x	water pressure along <u>the</u> x direction
38	I_x	moment of inertia with respect to <u>the</u> x -axis
39	I_y	moment of inertia with respect to <u>the</u> y -axis
40	k_e	earthquake contribution coefficient
41	k_1	<u>rainfall coefficient, taking 1 in the rainfall scenario and 0 in the non-rainfall scenario</u>
42	k_2	<u>earthquake coefficient, taking 1 in the seismic scenario and 0 in the non-seismic scenario</u>
43	k_3	<u>free surface coefficient, taking 1 for two free surfaces and 0 for three free surfaces</u>
44	M_{bx}	total bending moments with respect to the x -axis on the mudstone foundation
45	M_{by}	total bending moments with respect to the y -axis on the mudstone foundation
46	M_{bEx}	bending moment of E_x with respect to the x -axis on the mudstone foundation
47	M_{bHx}	bending moment of H_x with respect to the x -axis on the mudstone foundation
48	M_{bWx}	bending moment of W with respect to the x -axis on the mudstone foundation
49	M_{Ex}	overturning moment provided by E_x along <u>the</u> x direction
50	M_{Hx}	overturning moment provided by H_x along <u>the</u> x direction
51	M_{px}	stabilizing moment of p_n along <u>the</u> x direction
52	M_{Winx}	stabilizing moment provided by W along <u>the</u> x direction
53	M_{Woutx}	overturning moment provided by W along <u>the</u> x direction
54	N_z	total applied vertical load on the mudstone base
55	O	origin of the (x, y) coordinates
56	$p(x, y)$	pressure magnitude at point (x, y)
57	r_i	the basal cavity retreat ratio equal to the ratio of cavity width to block width in a certain direction
58	W	weight of the block
59	x	distance to O along the x -axis
60	y	distance to O along the y -axis
61	α	true dip of the contact surface
62	γ_s	unit weight of sandstone
63	γ_w	unit weight of water
64	θ_1	apparent dip of α on <u>the</u> -plane J1

65	θ_2	apparent dip of α on the -plane J2
66	σ_{cmax}	ultimate tensile-compressive strength of the mudstone
67	σ_{tmax}	ultimate tensile strength of the mudstone
68	τ_{max}	ultimate S shear S strength of the mudstone
69	φ	friction angle of the mudstone
70	ω_1	angle between the trend of the contact surface and the x direction
71	ω_2	angle between the trend of the contact surface and the y direction

72 1 Introduction

73 Rockfall is defined as the detachment of a rock block from a steep slope along a surface, on which little or no shear
74 displacement takes place ([Cruden and Varnes, 1996](#)). Rockfalls frequently occur in mountainous ranges, cut slopes, and
75 coastal cliffs, and they may cause significant ~~facilities-facility~~ damage and casualties in residential areas and transport
76 corridors ([Chau et al., 2003](#); [Volkwein et al., 2011](#); [Corominas et al., 2018](#)). Stability ~~and failure probability analysis~~ of rock
77 blocks are crucial for risk management and early warning of rockfall ([Kromer et al., 2017](#)).

78 Rockfall is widespread and poses high risk in the eastern Sichuan Basin, ~~s~~Southwest China ([Chen et al., 2008](#); [Chen and](#)
79 [Tang, 2010](#); [Zhang et al., 2016](#); [Zhou et al., 2017](#); [Zhou et al., 2018](#)). The rockfall in this area is attributed to the tectonic
80 setting of Jura-type folds and the stratum sequence, which is characterized by the interbedding of hard and soft layers. An
81 alternation of thick sandstone and thin mudstone layers ~~are-is~~ formed in the wide and gentle-angle synclines ([Zhang et al.,](#)
82 [2015](#); [Wu et al., 2018](#)). Weathering is known to be one of the main ~~predisposing~~ factors ~~for rockfall~~ ([Jaboyedoff et al., 2021](#);
83 [Zhan et al., 2022](#)). The cliff comprised of hard sandstone is the source of rockfall, and the underlying mudstone is more
84 susceptible to weathering. Along with the ~~extending-retreat~~ of basal cavities in ~~the~~ mudstone layer, ~~the gravity centre of the~~
85 ~~overlying sandstone block moves outward relative to the mudstone. In this case, the stress distribution in the contact surface~~
86 ~~of sandstone and mudstone is non-uniform. The mudstone on the outer side bears higher compressive stress than that on the~~
87 ~~inner side. the overlying sandstone blocks gradually become unstable because of the-This phenomenon can be defined as an~~
88 eccentricity effect, which ~~could fail with the mode of-~~leads to mudstone damage and failure of the overlying sandstone by
89 toppling or sliding. This type of rockfall is defined as biased rockfall in this study (Fig. 1). Similar rockfall patterns have
90 been widely reported in other regions, such as Joss Bay in England ([Hutchinson, 1972](#)), Okinawa Island in Japan ([Kogure et](#)
91 [al., 2006](#)), and the Colorado Plateau of the southwestern United States ([Ward et al., 2011](#)). ~~Extending-Retreat~~ of ~~the~~ basal
92 cavity is a main cause for the failure of ~~the~~ overlying block. Therefore, it is necessary to establish an analytical method,
93 considering the development of ~~the~~ basal cavity, to ~~analyzeanalyse~~ the stress distribution and stability of rock blocks, which
94 is fundamental to the susceptibility assessment and risk control of biased rockfall.

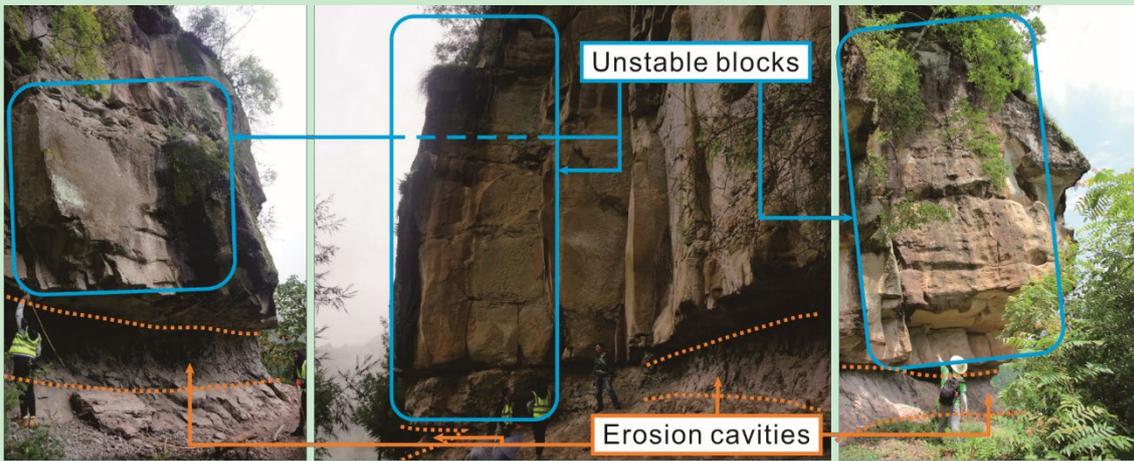
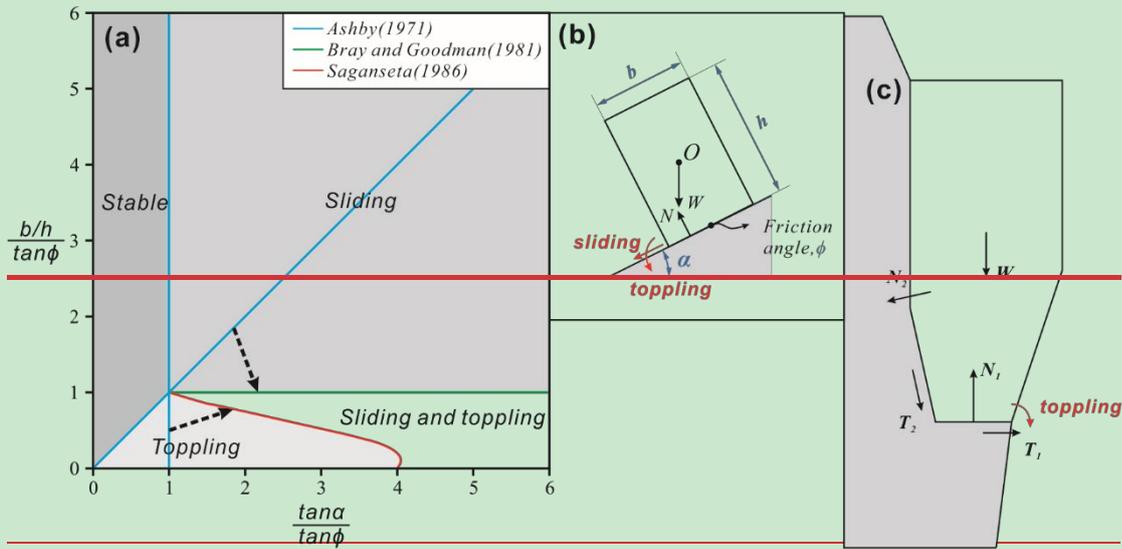


Figure 1 Potential ~~U~~unstable blocks and basal cavities caused by differential weathering.

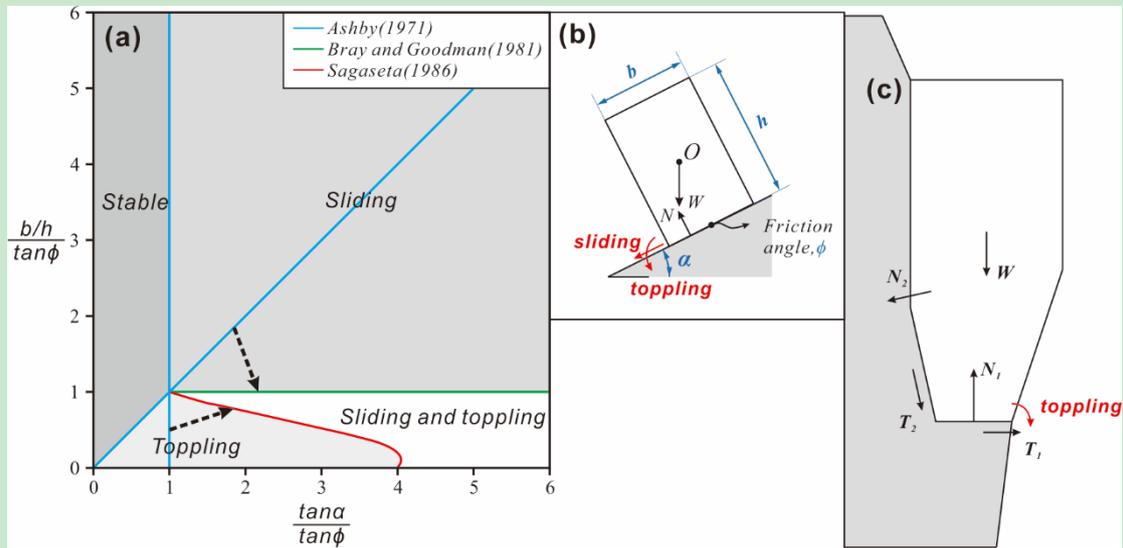
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97 Rockfall stability analysis methods include statistical analysis ([Frattini et al., 2008](#); [Santi et al., 2009](#)), empirical rating
 98 systems ([Pierson et al., 1990](#); [Ferrari et al., 2016](#)), and mechanical analysis ([Jaboyedoff et al., 2004](#); [Derron et al., 2005](#);
 99 [Matasci et al., 2018](#)). The statistical analysis and empirical rating systems are suitable for rockfall hazard assessment at a
 100 regional scale. The accuracy of statistical analysis depends on the completeness of rockfall inventories ([Chau et al., 2003](#);
 101 [Guzzetti et al., 2003](#); [D'amato et al., 2016](#)). However, its application to rockfall hazards is limited due to ~~the absence of~~
 102 ~~inventory data~~~~the lack of complete inventory data~~ ([Budetta and Nappi, 2013](#); [Malamud et al., 2004](#)). Empirical and semi-
 103 empirical rating systems are used where site-specific rockfall inventories are either unavailable or unreliable. Therefore,
 104 rockfall susceptibility can be assessed by heuristic ranking of selected predisposing factors ([Frattini et al., 2008](#); [Budetta,](#)
 105 [2004](#)). Mechanical analysis based on static equilibrium theory is the main method to ~~analyze~~~~analyse~~ the stability of site-
 106 specific rockfall using the factor of safety (*Fos*). [Ashby \(1971\)](#) ~~has~~ conducted stability analysis with ~~a~~ parallelepiped block
 107 resting on ~~an~~ inclined plane (Fig. 2a), ~~and~~ the solution was subsequently modified by [Bray and Goodman \(1981\)](#), and
 108 [Sagaseta \(1986\)](#). [Kogure et al. \(2006\)](#) utilized ~~the a~~ cantilever beam model to determine the critical state of limestone cliffs.
 109 [Frayssines and Hantz \(2009\)](#) proposed the limit equilibrium method (LEM) to predict block stability ~~considering against~~
 110 sliding and toppling in steep limestone cliffs (Fig. 2bc). [Chen and Tang \(2010\)](#) established a stability analysis method of
 111 three types of unstable rocks; in the Three Gorges Reservoir area with the LEM. [Alejano et al. \(2015\)](#) studied the influence
 112 of rounding of block corners on the block stability. [Zhang et al. \(2016\)](#) defined *Fos* based on fracture mechanics and studied
 113 the progressive failure process by ~~analyzing~~~~analysing~~ crack propagation. [Alejano et al. \(2010\)](#) and [Pérez-Rey et al. \(2021\)](#)
 114 deduced ~~the a~~ formula for *Fos* of blocks with more complex geometry.



115



116

117 **Figure 2** Traditional force analysis diagrams of the rock block. (a) and (b) are stability analysis diagrams of rock blocks in-under dynamic
 118 conditions, resting on an inclined plane with the-a dip angle of α . The rock block is generalized as a cuboid with the-dimensions of $b \times h$
 119 $\times h$ and weight W (as modified from Ashby (1971), Bray and Goodman (1981) and Sagaseta (1986)). (c) Force description of the toppling
 120 model proposed by Frayssines and Hantz (2009). In the above assumptions, N , T , and W are regarded as a-forces applied at a point.
 121 The supporting force at-on the contact surface is assumed to be applied at a point in the current LEM methods (i.e., N in Fig.
 122 2 b and c). However, the supporting force is actually a distributed force. The cavity generates an eccentricity effect on the
 123 overlying rock mass-block and results in a non-uniform distribution of the supporting force on the contact surface, which is
 124 not considered in the traditional LEM. Furthermore, most studies simplified the three-dimensional geometry of the slope
 125 rock block by one cross-section, which is used to represent the critical features of the slope structure. Nevertheless, for
 126 natural blocks with basal cavities (Pérez Rey et al., 2021), the cavities usually present different depths along different

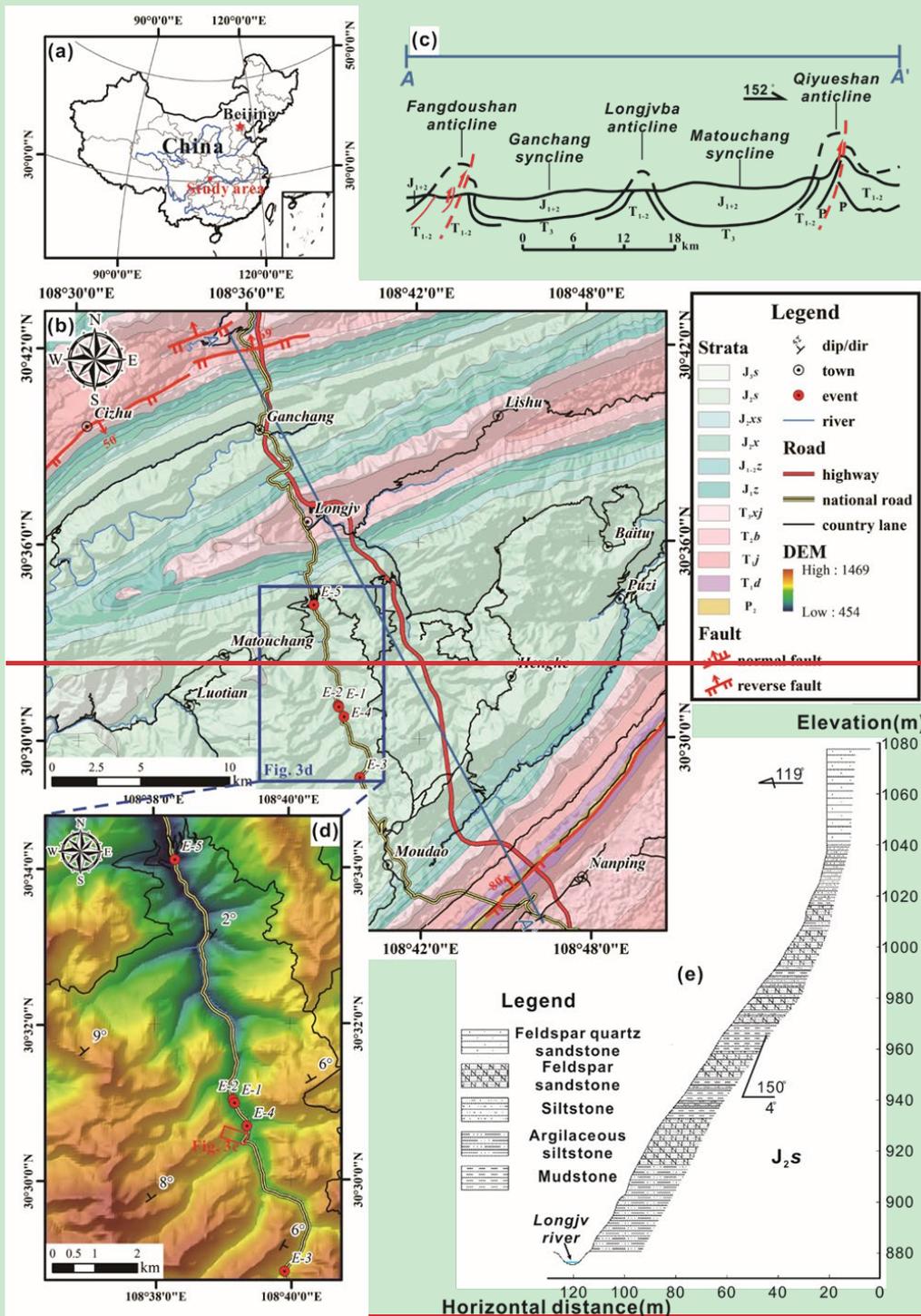
127 directions ~~usually present different depths~~(Pérez-Rey et al., 2021). Therefore, a three-dimensional model is necessary to
128 calculate the accurate stability. ~~Besides~~In addition, when a block has multiple free faces and a complex structure, ~~it's~~its
129 potential failure ~~will be~~is dominated by different modes, including rock mass damage and overall block failure. Therefore,
130 the probable failure modes should be determined prior to the calculation of *Fos*.

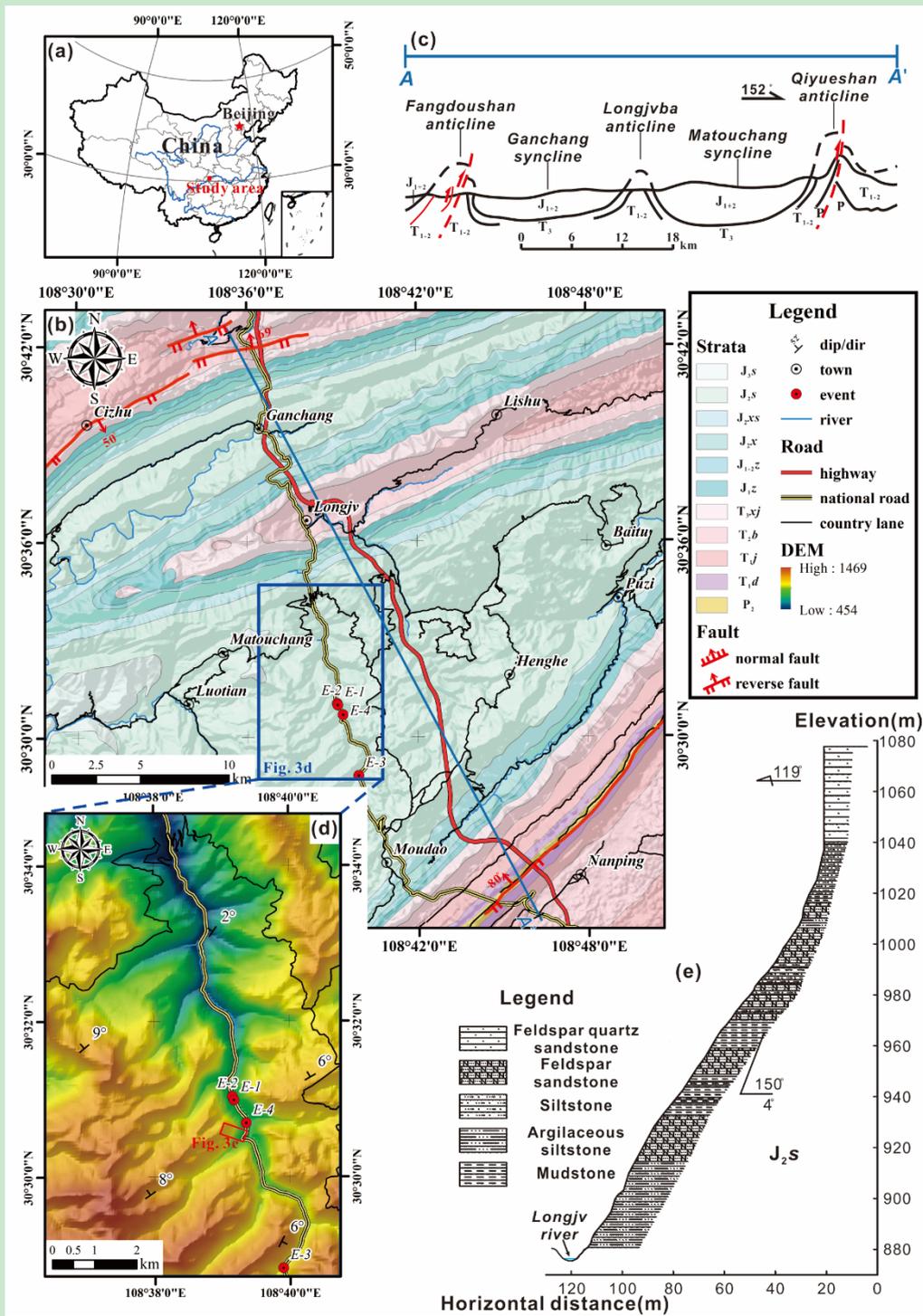
131 Based on rockfall investigation in the Eastern Sichuan Basin, China, the main objective of this study was to propose a new
132 three-dimensional method for the determination of failure modes and *Fos* of biased rockfall, considering the non-uniform
133 force distribution on the contact surfaces. Compared with the traditional LEM method, this study takes into account the
134 partial damage of the underlying soft rock and the overall instability of the overlying hard rock blocks, and can evaluate the
135 stability of biased rockfall more comprehensively. *Fos* of the typical unstable rock blocks in the study area ~~were~~are
136 calculated to validate the ~~new~~proposed method. In addition, the critical cavity retreat ratio (α) ~~in~~is this area ~~was~~is
137 ~~given~~analysed. This study is an extension of the basic LEM for rockfall, which ~~could~~can promote the accuracy of rockfall
138 stability analysis and facilitate rockfall prevention and risk mitigation.

139 2 Study area

140 2.1 Geological setting

141 The study area is located ~~in~~on the northeastern edge of the Sichuan ~~b~~Basin, China (Fig. 3a). Continuous erosion processes
142 generate moderate-low mountain and valley landforms (Yu et al., 2021). The tectonic structure of this area is characterized
143 by a series of ~~NEE-ENE~~ anticlines and synclines (Fig. 3b, c). In the anticlines ~~s~~ area, the rock layers dip relatively steeply,
144 where translational rockslides ~~is~~are the main mode of slope failure. The syncline area is dominated by ~~gentle~~gently dipping
145 strata and is prone to rockfall (Zhou et al., 2018). The study area is located in the core of the Matouchang syncline, where the
146 rock layers are sub-horizontal (Fig. 3d, e). In this valley, due to the longstanding fluvial incision, the relative relief is ~~about~~
147 approximately 500 m and the valley flanks are extremely steep (Fig. 3e). ~~Besides~~In addition, the toes of the hill slopes are
148 reshaped because of the construction of the G318 national road, which is the main traffic line and is always threatened by
149 rockfalls dropping from ~~the~~ steep rock slopes (shown in Fig. 3d and Table 1).





151

152 **Figure 3** (a) Location of the study area in China; (b) geological map of the study area; (c) tectonic sketch profile of A-A'; whose location

153 is shown in Fig. 3b; (d) rockfall-prone segment and key investigation areas. The red dots are the positions of historical rockfall events,

154 corresponding to the ~~serial~~ numbers in Table 1; (e) Geological cross-section of the hillslope in the Jitougou section of G318 national road,
155 which is marked by a red rectangle in Fig. 3d.

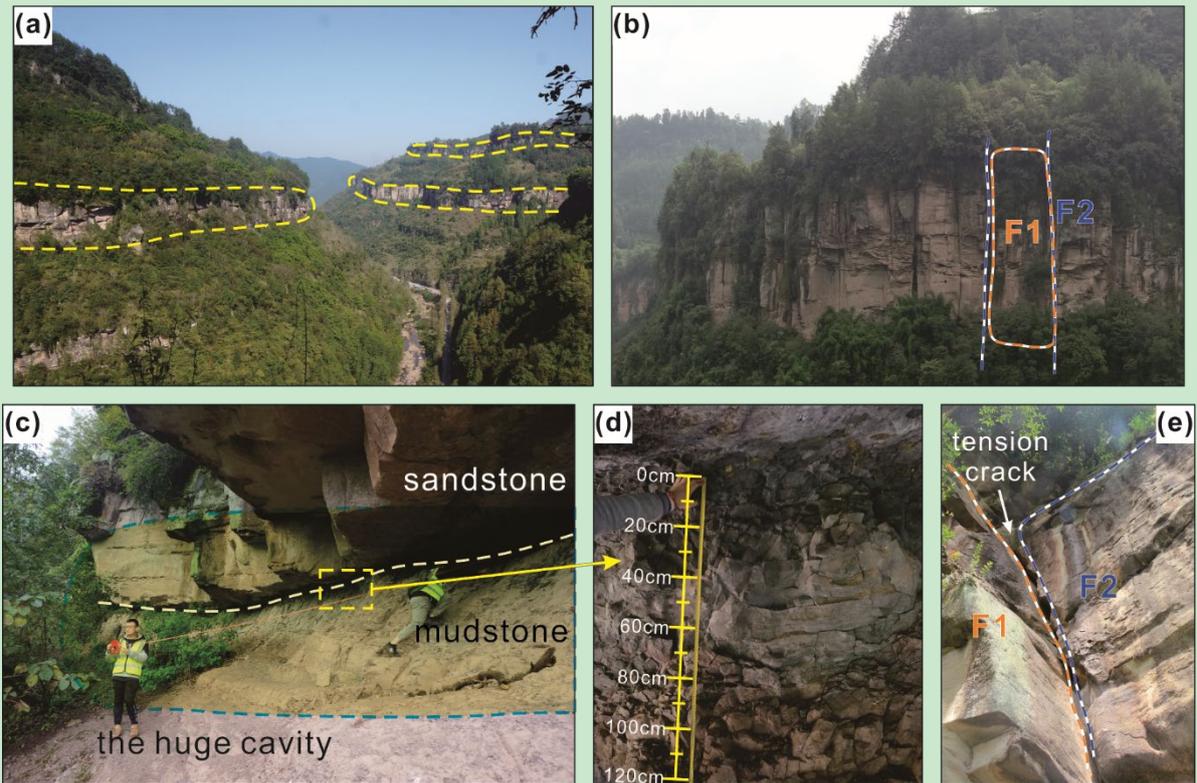
156 **Table 1** Historical rockfall events along G318 national road in the study area

Serial number <u>No</u>	Location	Time of occurrence (GMT+8)	Volume [m ³]	Consequence
E-1	K1698+900	2014-05 to 06*	Unknown	The power transmission facilities outside the road were smashed.
E-2	K1699+000	2015-02-14 23:00	About 240	A passing truck was stuck and two people dead.
E-3	K1690+700	2015-06-16	Unknown	The road was interrupted for a day.
E-4	K1698+400	2015-06-18 09:00	About 200	A vehicle was crashed into a gully and four people dead.
E-5	K1741+800	2020-04-21 05:30	About 232000	Eight houses were damaged and a gas station was affected.

157 *Note: The exact time is unknown.

158 2.2 Rockfall characteristics

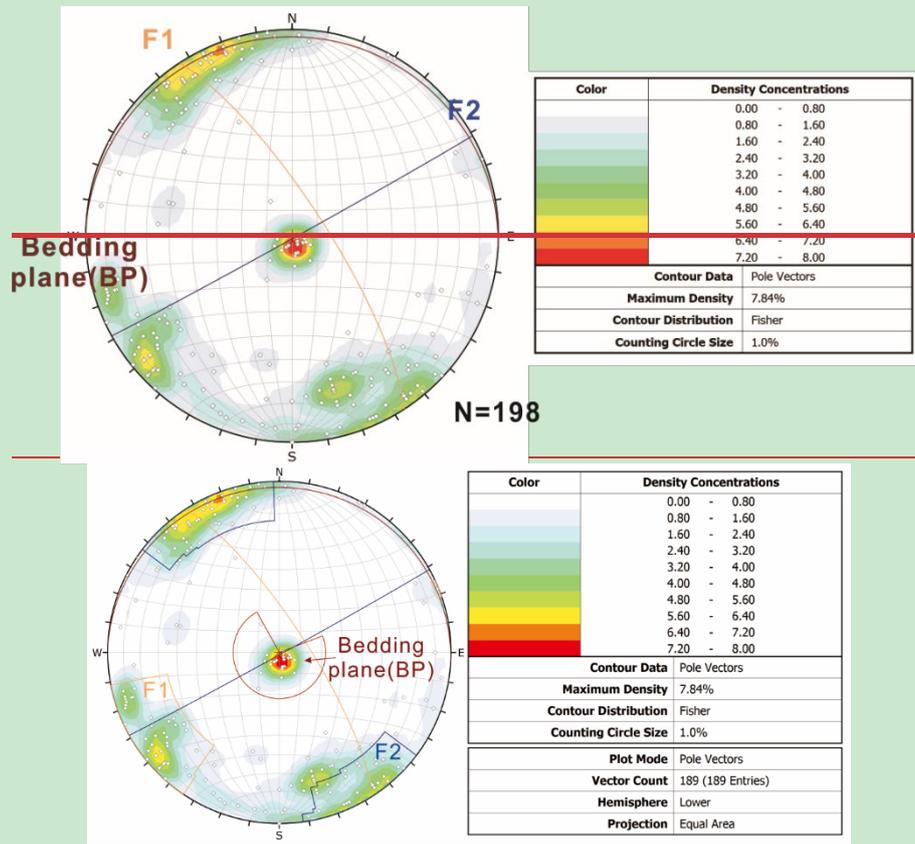
159 The slopes in the study area ~~have multi layer unstable rock blocks (Fig. 4a), which are~~ consist of a sub-horizontally
160 ~~interbedded~~interbedding of sandstone and mudstone layers. Therefore, there are multiple layers of potentially unstable rock
161 blocks in the hill slopes (Fig 4a). The thick sandstone has two sets of sub-vertical joints (Fig. 5), which cut the rock mass
162 into blocks as the potential rockfall source (Fig. 4b). Cavities ~~are~~have formed in the underlying mudstone layer (Fig. 4c, d).
163 Joints and bedding planes (BP) constitute the detachment surfaces between the blocks and steep slope (Fig. 4e). The
164 Eccentricity effect produced by the mudstone cavity plays an important role in the ~~rockfall evolution~~ process of rockfall.
165 When the basal mudstone cannot provide adequate supporting force, the blocks detach from the steep slope, and biased
166 rockfall occurs. ~~There are two possible failure modes of biased rockfall, namely s~~Sliding and toppling are two possible
167 failure modes of biased rockfall.



168

169 **Figure 4** Characteristics of biased rockfalls in the study area. (a) ~~Multiple-layers~~Multi-layer of rockfall sources, which is consist of thick
 170 sandstone. (b) Two sets of sub-vertical joints (F1 and F2) recognized by the UAV photos. (c) ~~A~~Large basal cavity developed in the
 171 underlying mudstone. (d) ~~The~~dense fractures on the mudstone surface generated by weathering and compression. (e) ~~A~~vertical tension
 172 crack in the rear of the block, through which precipitation can infiltrate.

173 ~~In the study area, rainfall is the main predisposing factors of rockfall. According to the historical rockfall events in this area,~~
 174 ~~precipitation is considered a triggering effect of rock instability.~~ The precipitation mainly infiltrates along the sub-vertical
 175 joints or cracks of the sandstone (Fig. 4e). However, the ~~draining~~drainage of fissure water is hysteretic due to the
 176 obstruction of basal mudstone. Therefore, transient steady flow exists in vertical cracks during heavy rainfall, and the
 177 hydrostatic pressure triggers the detachment of rock blocks. Thus, typical scenarios (such as rainfall intensity and earthquake)
 178 need to be considered in the stability analysis model.



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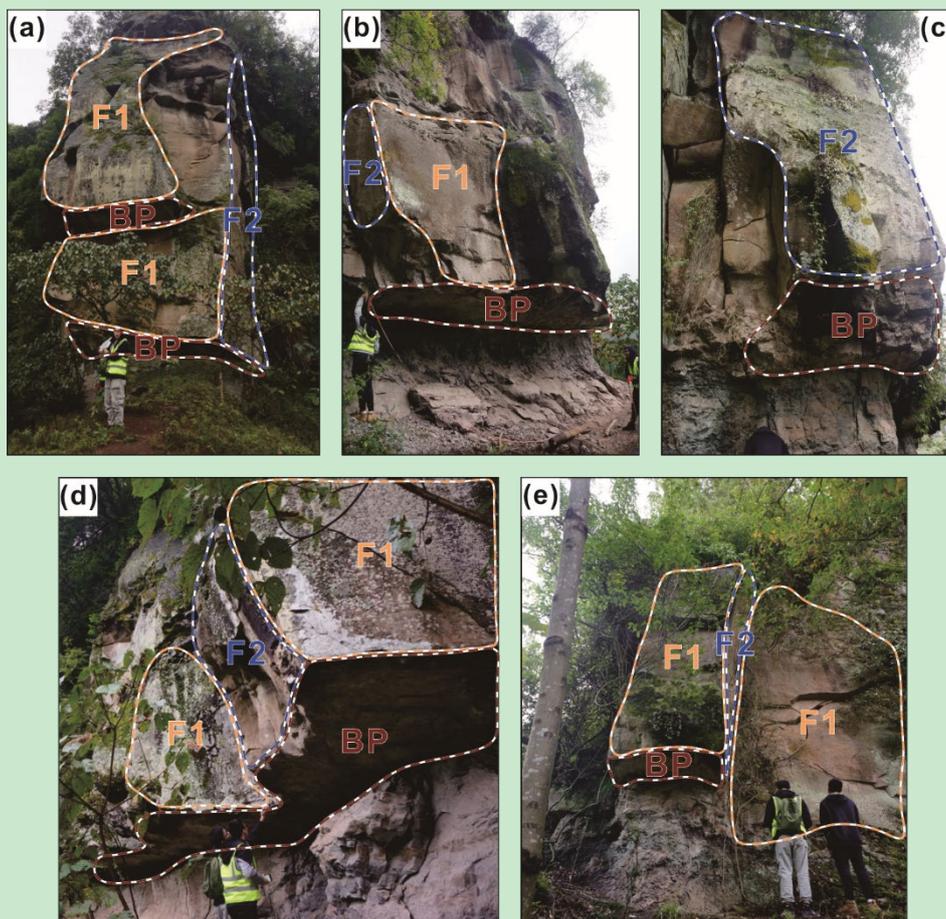
181 **Figure 5** Stereo net produced using compass-clinometer survey data, which shows the densities and orientations of five clusters. The data
 182 were collected in the rockfall-prone area shown in Fig. 3d.

183 **3 Calculation method**

184 **3.1 Geological models and assumptions**

185 A detailed geological investigation ~~for of~~ unstable rock blocks ~~were was~~ carried out in the study area (Fig. 6). The geological
 186 model of the rock block is mainly composed of the overlying sandstone and the underlying mudstone. The sandstone block is
 187 assumed to be a rigid body, which is divided by two sets of orthogonal vertical smooth joints without friction resistance.
 188 According to the relatively persistent sub-vertical fractures observed in the field, the vertical joints are assumed to be fully
 189 persistent in the geological model. Blocks with multiple free faces are prone to failure. The sandstone block is assumed to be
 190 a complete body without persistent discontinuity, and it will not disintegrate before it falls. Due to the cavity in ~~the~~ mudstone,
 191 the contact surface between sandstone and mudstone has exhibits an eccentricity effect. ~~The underlying mudstone plays the~~
 192 ~~role of a rectangular base, which provides where non-uniform distributed forces stresses are distributed~~ at different
 193 ~~locations positions.~~ Mudstone is loaded by compressive stress, tensile stress and shear stress, however, it doesn't present

194 ~~deformation.~~ Mudstone is mainly loaded by compressive stress and tensile stress. When the compressive stress of mudstone
 195 ~~exceeds its strength on the outer side, some initial damage appears.~~ The effective contact surface between mudstone and
 196 ~~sandstone is reduced, which aggravates the non-uniform distribution of stress.~~ In this way, the ability of mudstone to resist
 197 ~~the sliding and toppling of overlying sandstone is reduced.~~ In the field, compression deformation of mudstone can be
 198 ~~observed, which usually manifests as micro-fractures and cleavages (Fig. 4d).~~ The deformation is very slight and slow in the
 199 ~~short term.~~ In addition, the LEM is essentially a force/stress approach that does not take into account the deformation.
 200 ~~Therefore, in this study, it is assumed that the mudstone is not subjected to deformation.~~ The rock block ~~keeps-remains~~ in the
 201 state of static equilibrium prior to the final overall failure. Fig. 7 ~~expresses-displays~~ the four evolution stages of biased
 202 rockfall. In the initial stage, the base cavity has not yet formed, and the normal force acting on the contact surface is uniform
 203 in different positions. ~~The Eccentricity~~ effect leads to a non-uniform supporting force as the cavity grows, ~~and partially~~
 204 damage gradually develops ~~until-when~~ the non-uniform ~~force-stress~~ exceeds the compressive or tensile strength of the
 205 mudstone. Under the triggering effects of rainfall or earthquakes, the rock blocks ~~will beare~~ separated by sliding or toppling.



206

207 **Figure 6** The unstable blocks ~~were~~ labelled W02, W08, W18, W04, ~~and~~ W21, which are detached by the ~~dominated-dominating~~
 208 discontinuities in Fig. 5. ~~Obvious-b~~Basal cavities can be identified under the bedding planes of sandstone.

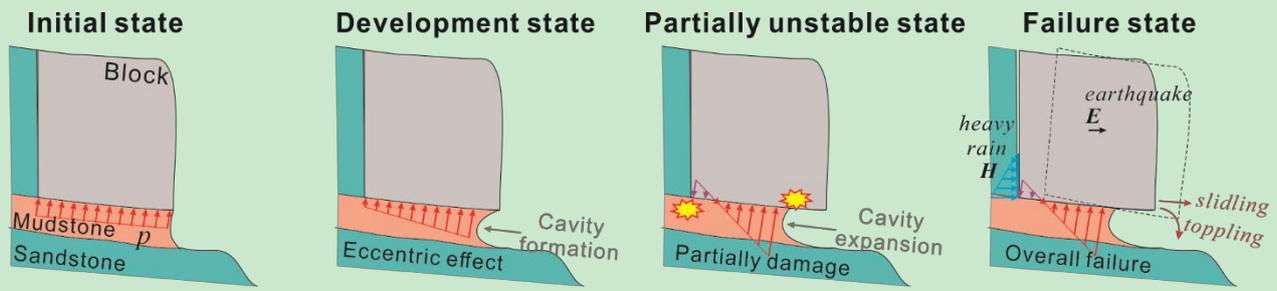


Figure 7 The evolution process of rock blocks from stable state to failure.

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210

211 Fig. 8 represents the mechanical model of the force equilibrium analysis of a rock block with two or three free faces. The
 212 rock block (the overlying sandstone) is generalized as a parallelepiped block. The underlying mudstone is impermeable, so
 213 rainfall can fill the joints and transmit horizontal hydrostatic pressure. The shear strength of the underlying mudstone is
 214 assumed to obey s the Mohr–CoulombMohr-Coulomb criterion. The predisposing factors s of rRainfall and earthquakes
 215 decrease *FOS* by generating hydrostatic pressure *H* in the vertical crack and horizontal seismic force *E* on the block.

216 A Cartesian coordinate system is established in three-dimensional space for the force analysis. The origin *O* is located at the
 217 centre of the contact surface between sandstone and mudstone. For the case with two free surfaces, the orientation of the free
 218 surfaces is set to be the positive direction of the *x*-axis and *y*-axis. For the case with three free surfaces, the negative
 219 direction of the *x*-axis is also a free surface. Joint J2 is perpendicular to the *x*-axis, and joint J1 is perpendicular to the *y*-axis.

	3D View	Side view along y direction	Side view along x direction
(a) Two free faces			
(b) Three free faces			
	3D View	Side view of the yz plane	Side view of the xz plane
(a) Two free faces			
(b) Three free faces			

Figure 8 Diagram of the force equilibrium analysis of the rock block model. (a) and (b) represent the case of unstable rock blocks with two or three free vertical surfaces, respectively.

3.2 Calculation processes

3.2.1 ~~Distributed force~~ Stress distribution at the block base

The following formulas are used to calculate the apparent dip of α (θ_1 and θ_2):

$$\theta_1 = \arctan(\tan \alpha \cdot \cos \omega_1) \quad (1)$$

$$\theta_2 = \arctan(\tan \alpha \cdot \cos \omega_2) \quad (2)$$

Where, ~~where~~ ω_1 and ω_2 are the angles between the trend of the contact surface and the x direction or y direction, respectively.

As shown in Fig. 8b, with respect to the x -axis, gravity, seismic forces, and hydrostatic pressure create a bending moments non-symmetrical stress distribution on the foundation. The bending moment of gravity with respect to the x -axis (M_{bWx}) is

$$M_{bWx} = W \cdot \frac{d_1 - d_3}{2} \cos \theta_1 \quad (3)$$

Assume-Assuming that the height of the water in the fracture is h_w , the hydrostatic pressure along the x direction (H_x) and its bending moment (M_{bHx}) are respectively expressed as;

$$H_x = \frac{\gamma_w h_w^2}{2} (b - d_2) \quad (4)$$

$$M_{bHx} = \int_{-\frac{b-d_2}{2}}^{\frac{b-d_2}{2}} \int_0^{h_w \cos \theta_1} \gamma_w \left(h_w - \frac{z}{\cos \theta_1} \right) \left(\frac{z}{\cos \theta_1} + \frac{a - d_1 - d_3}{2} \cdot \sin \theta_1 \right) dz dy \quad (5)$$

The horizontal seismic force along x direction (E_x) and its bending moment (M_{bEx}) are respectively expressed as;

$$E_x = k_e W \quad (6)$$

$$M_{bEx} = E_x \left(\frac{h}{2} - \frac{d_1 - d_3}{2} \sin \theta_1 \right) \quad (7)$$

The total applied vertical load (N_z) and the total bending moments along the x direction (M_{bx}) can be derived as;

$$N_z = W \cos \alpha - (H_x \cdot k_1 \cdot k_3 + E_x \cdot k_2) \sin \theta_1 - (H_y \cdot k_1 + E_y \cdot k_2) \sin \theta_1 \quad (8)$$

$$M_{bx} = M_{bWx} + M_{bHx} \cdot k_1 \cdot k_3 + M_{bEx} \cdot k_2 \quad (9)$$

Under natural condition, k_1 and k_2 are both equal to 0. Under rainfall conditions, $k_1 = 0$. Under earthquake conditions, $k_2 = 0$. For the case of two free faces, $k_3 = 1$; for the case of three free surfaces, $k_3 = 0$, where k_1 , k_2 and k_3 are the coefficients set to make Eq. (8) and Eq. (9) compatible with different calculation scenarios. Therefore, Eqs. (8) and (9) and the following formulas can be expressed in a unified form. In the natural scenario, k_1 and k_2 are both equal to 0. In the rainfall scenario, $k_1 = 1$. In the earthquake scenario, $k_2 = 1$. For the case of two free faces, $k_3 = 1$. For the case of three free surfaces, $k_3 = 0$.

250 Based on bending theory (Adrian, 2010), the eccentricity distance along the x direction (e_x) can be expressed as;

$$251 \quad e_x = \frac{M_{bx}}{N_z} = \frac{M_{bWx} + M_{bHx} \cdot k_1 \cdot k_3 + M_{bEx} \cdot k_2}{W \cos \alpha - (H_x \cdot k_1 \cdot k_3 + E_x \cdot k_2) \sin \theta_1 - (H_y \cdot k_1 + E_y \cdot k_2) \sin \theta_1} \quad (10)$$

252 The same method can be used to obtain e_y ;

$$253 \quad e_y = \frac{M_{by}}{N_z} = \frac{M_{bWy} + M_{bHy} \cdot k_1 + M_{bEy} \cdot k_2}{W \cos \alpha - (H_x \cdot k_1 \cdot k_3 + E_x \cdot k_2) \sin \theta_1 - (H_y \cdot k_1 + E_y \cdot k_2) \sin \theta_1} \quad (11)$$

254 According to the stress distribution of a rectangular shaped foundation (Adrian, 2010), the stress in the (x, y) coordinates,

255 $p(x, y)$, results to be;

$$256 \quad p(x, y) = \frac{N}{A} + \frac{Ne_x}{I_y} x + \frac{Ne_y}{I_x} y \quad (12)$$

257 With the formulas;

$$258 \quad I_x = \frac{(a - d_1)(b - d_2)^3}{12} \quad (13)$$

$$259 \quad I_y = \frac{(b - d_2)(a - d_1)^3}{12} \quad (14)$$

$$260 \quad A = (a - d_1 - d_3)(b - d_2) \quad (15)$$

261 By substituting Eq. (13-15) into Eq. (12), $p(x, y)$ can be derived as;

$$262 \quad p(x, y) = \frac{N}{A} \left[1 + \frac{12e_x}{(a - d_1 - d_3)^2} x + \frac{12e_y}{(b - d_2)^2} y \right] \quad x \in \left[-\frac{a - d_1 - d_3}{2}, \frac{a - d_1 - d_3}{2} \right], y \in \left[-\frac{b - d_2}{2}, \frac{b - d_2}{2} \right] \quad (16)$$

263 p_{max} and p_{min} can be derived from Eq. (16) as;

$$264 \quad p_{max} = p \left(\frac{a - d_1 - d_3}{2}, \frac{b - d_2}{2} \right) \quad (17)$$

$$265 \quad p_{min} = p \left(-\frac{a - d_1 - d_3}{2}, -\frac{b - d_2}{2} \right) \quad (18)$$

266 The mudstone foundation has both the compressive strength and tensile strength, so the value of $p(x, y)$ is modified to
267 obtain the two piecewise functions;

$$268 \quad p_p(x, y) = \begin{cases} \sigma_{cmax}, & p(x, y) \geq \sigma_{cmax} \\ p(x, y), & 0 < p(x, y) \leq \sigma_{cmax} \\ 0, & p(x, y) < 0 \end{cases} \quad (19)$$

$$269 \quad p_n(x, y) = \begin{cases} 0, & p(x, y) < -\sigma_{tmax} \\ p(x, y), & -\sigma_{tmax} \leq p(x, y) < 0 \\ 0, & p(x, y) \geq 0 \end{cases} \quad (20)$$

270 Here, $p_p(x, y)$ provides support normal force for the underlying-overlying sandstone, and $p_n(x, y)$ provides tension force.

271 3.2.2 Calculation of factors of safety

272 According to the principle of friction, the ultimate shear strength τ_{max} results to be;

$$\tau_{max} = \int_{\frac{a-d_1-d_3}{2}}^{\frac{a-d_1-d_3}{2}} \int_{\frac{b-d_2}{2}}^{\frac{b-d_2}{2}} [p_p(x, y) \tan \varphi + c] dy dx \quad (21)$$

Therefore, ~~the~~ FOS against sliding, FOS_{sl} , can be defined as:

$$FOS_{sl} = \frac{S_{stabilizing}}{S_{sliding}} = \frac{\tau_{max}}{W |\sin \alpha_s| + H_x \cdot \cos \omega_s \cdot \cos \alpha_s \cdot k_1 \cdot k_3 + H_y \cdot |\sin \omega_s| \cdot \cos \alpha_s \cdot k_1 + E \cdot \cos \alpha_s \cdot k_2} \quad (22)$$

When the block can slide freely, $\alpha_s = \alpha$, $\omega_s = 0$; when the block is constrained to slide along a joint plane (e.g., J1), $\alpha_s = \theta_1$ or θ_2 , $\omega_s = \omega_1$ or ω_2 . For the case of an anaclinal slope, the sliding direction is opposite to the free surface. Therefore, the rock block does not slide, and FOS_{sl} is not exist considered in the model.

With regard to stability against toppling, along the x direction, the part of the block above the mudstone base provides the stabilizing moment $M_{W_{inx}}$, and the part of the block above the cavity provides the overturning moment $M_{W_{outx}}$. When ~~the~~ tension exists, there ~~will be~~ an additional stabilizing moment. M_{px} , $M_{W_{inx}}$, $M_{W_{outx}}$ and M_{px} can be derived as:

$$M_{W_{inx}} = W \frac{a-d_1}{a} \cos \theta_1 \cdot \left(\frac{a-d_1}{2} \right) \quad (23)$$

$$M_{W_{outx}} = W \frac{d_1}{a} \cos \theta_1 \cdot \frac{d_1}{2} \quad (24)$$

$$M_{px} = - \int_{\frac{b-d_2}{2}}^{\frac{b-d_2}{2}} \int_{\frac{a-d_1-d_3}{2}}^{\frac{a-d_1-d_3}{2}} p_n(x, y) \cdot \left(\frac{a}{2} - d_1 - x \right) dx dy \quad (25)$$

~~And~~ M_{Hx} and M_{Ex} can be derived as:

$$M_{Hx} = \int_{\frac{b-d_2}{2}}^{\frac{b-d_2}{2}} \int_0^{h_w \cos \theta_1} \gamma_w \left(h_w - \frac{z}{\cos \theta_1} \right) \left(\frac{z}{\cos \theta_1} + (a-d_1) \sin \theta_1 \right) dz dy \quad (26)$$

$$M_{Ex} = E_x \left(\frac{h}{2} + \left(\frac{a}{2} - d_1 \right) \sin \theta_1 \right) \quad (27)$$

Therefore, the FOS against toppling along the x direction, FOS_{tox} , results ~~to be~~ in

$$FOS_{tox} = \frac{M_{stabilizing}}{M_{overturning}} = \frac{M_{W_{inx}} + M_{px}}{M_{W_{outx}} + M_{Hx} \cdot k_1 \cdot k_3 + M_{Ex} \cdot k_2} \quad (28)$$

Similarly, ~~the~~ FOS_{toy} can be obtained as:

$$FOS_{toy} = \frac{M_{stabilizing}}{M_{overturning}} = \frac{M_{W_{iny}} + M_{py}}{M_{W_{outy}} + M_{Hy} \cdot k_1 + M_{Ey} \cdot k_2} \quad (29)$$

The smaller value is selected as the FOS of the toppling failure mode, FOS_{to} :

$$FOS_{to} = \min(FOS_{tox}, FOS_{toy}) \quad (30)$$

When the stress on mudstone exceeds its strengths, it ~~has causes~~ partial damage and decreases the stability of the rock block.

Therefore, ~~the~~ FOS with the consideration of compressive strength (FOS_{co}) and tensional strength (FOS_{te}) can be derived as:

$$FOS_{co} = \frac{\sigma_{tmax}}{p_{max}} \quad (31)$$

$$Fos_{te} = \frac{\sigma_{etmax}}{-p_{min}} \quad (32)$$

297

298 Fos_{co} and Fos_{te} represent the current damage degree of mudstone due to compressive stress and tensile stress, respectively.
299 When the stress exceeds the ultimate strength, the strength of the mudstone is reduced to the residual value, and the initial
300 deformation appears. The ability of mudstone to provide resistance to the sliding and toppling of sandstone blocks is thus
301 reduced, and Fos_{sl} and Fos_{to} subsequently decline. The smaller the value of Fos_{co} and Fos_{te} , the greater the damage to the
302 underlying mudstone. The effective contact area between sandstone and mudstone becomes smaller as the development of
303 compressive and tension damage, which significantly affects the stability of the overlying sandstone block.

304 Finally, four Fos of unstable rock block ~~were~~ are obtained. Fos_{sl} and Fos_{to} are routine indicators ~~directly about representing~~
305 the stability of sandstone blocks ~~that directly characterize its stability~~. Fos_{co} and Fos_{te} are two indicators proposed in this
306 study, for the stability analysis of biased rockfall, which describe the damage state of the underlying mudstone base. It is
307 necessary to ~~aggregate simultaneously consider~~ four Fos to judge-evaluate the stability of unstable ~~biased rockfall~~ rock mass.

308 The entire calculation process is shown in Fig. 9.

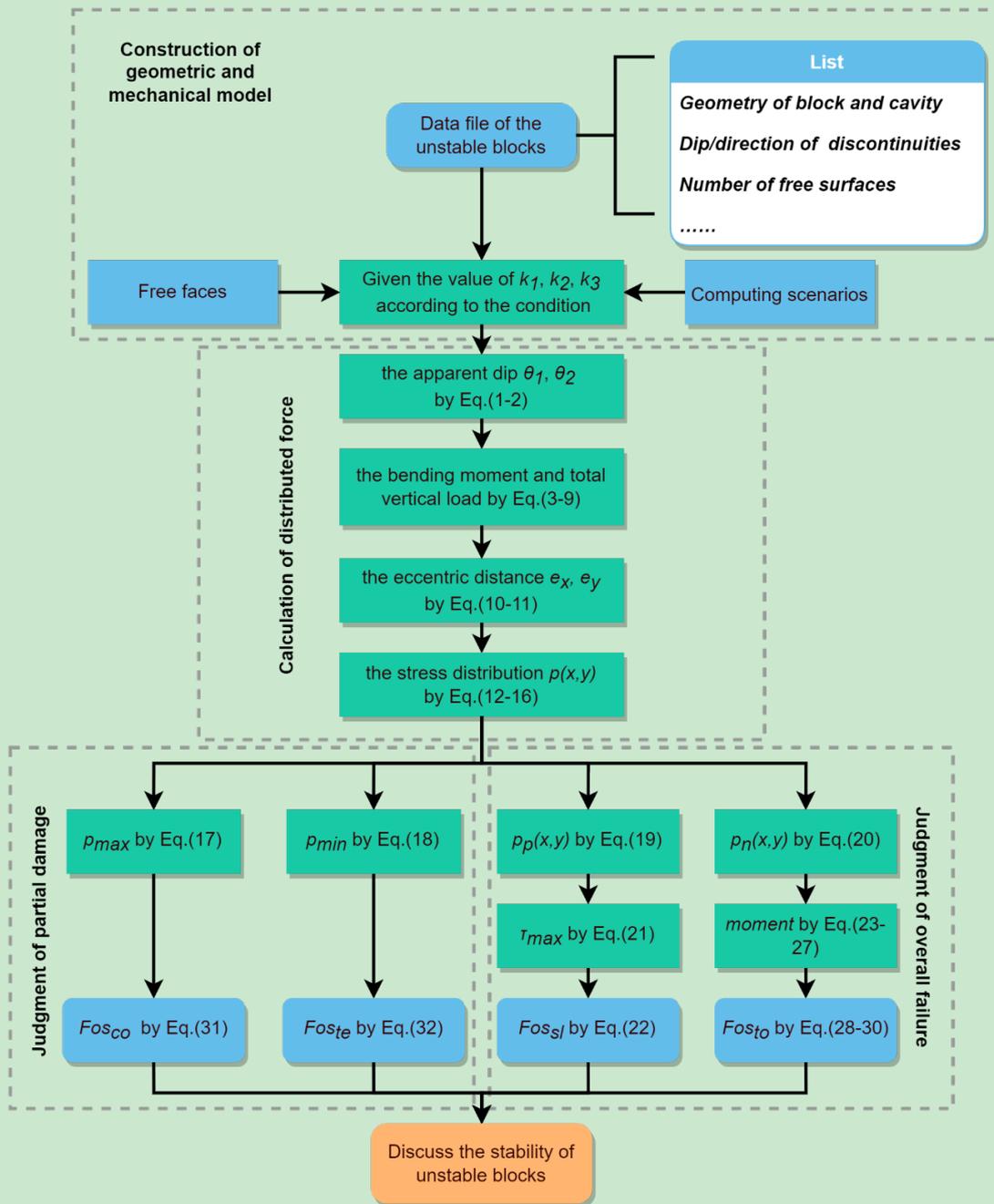


Figure 9 The calculation process of Fos of the unstable rock blocks.

309

310

311 4 Parameters and results

312

313 A detailed field investigation was carried out in the source area of rockfall (Fig. 3d ~~and 4a~~). The size of the blocks ~~is~~ was
314 determined by on-site measurement with tape and a laser rangefinder. The basal cavities in mudstone were measured with a
315 steel ruler, and the morphological characteristics of mudstone foundation ~~are~~ were mainly described with the average erosion
316 depth of the cavity. The attitude of discontinuities was measured by compass. The mechanical parameters for the *Fos*
317 calculation of rock blocks ~~are~~ were ~~abundantly recorded in~~ determined by referring to published literature and the
318 investigation reports ~~and published literatures~~ in this area. The unit weight of the sandstone block (γ_s) is 25 kN/m³ (Tang et
319 al., 2010), the friction angle of the contact surface (φ) is set to ~~be~~ 25° and the cohesion (c) is set to ~~be~~ 70 kPa (Zhang et al.,
320 2016). Because of the strength degradation of mudstone foundations s due to intense ~~weathered~~ weathering, the maximum
321 compressive stress of mudstone (σ_{cmax}) is replaced by the bearing capacity of mudstone foundations s (2300 kPa), which is
322 obtained through plate load test s in adjacent areas s (Zheng et al., 2021). ~~Besides~~ In addition, the maximum tensile stress of
323 mudstone (σ_{tmax}) is value as one-one ninth of σ_{cmax} . The height of the water level (h_w) is set to be one-third of h , and an
324 earthquake contribution coefficient k_e of 0.05 is considered in stability calculations. The data obtained from the field survey
325 were organized according to the coordinate system of the geological model in Section 3.1, and *Fos* was calculated according
326 to the calculation steps in Section 3.2. The calculated geometric parameters and *Fos* results are shown in Table 2.

Table 2 Geometric parameters of rock blocks in the study area and Fos results obtained from the analytical method in section 3.

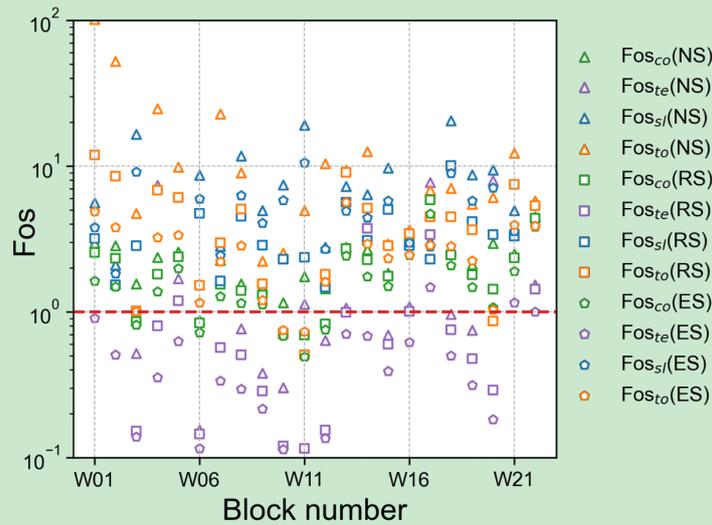
Block number	Free faces	h [m]	a [m]	b [m]	d_1 [m]	d_2 [m]	d_3 [m]	α [°]	Dip direction [°]			NS (Natural scenario)					RS (Rainfall scenario)				ES (Earthquake scenario)			
									BD	J1	J2	Fos_{te}	Fos_{co}	Fos_{sl}	Fos_{to}	Fos_{min}	Fos_{te}	Fos_{co}	Fos_{sl}	Fos_{to}	Fos_{te}	Fos_{co}	Fos_{sl}	Fos_{to}
W01	3	23	7.2	6.1	0.65	0.25	0.17	6	78	7	97	-	2.99	5.61	101.54	2.99	-	2.56	3.18	11.91	0.90	1.63	3.81	4.88
W02	3	23	6.42	5.25	0.78	0.4	0.31	16	148	51	141	-	2.84	2.10	52.28	2.10	-	2.33	1.54	8.49	0.51	1.48	1.82	3.79
W03	2	20	3.5	2.6	0.84	0.55	-	7	341	53	143	0.52	1.56	16.53	4.72	0.52	0.15	0.86	2.83	1.02	0.14	0.81	9.12	1.01
W04	2	19	4.6	4.6	0.62	0.77	-	7	273	65	155	7.35	2.37	-	24.74	2.37	0.80	1.81	-	6.83	0.35	1.38	-	3.23
W05	2	15	16.7	5.6	2.13	1.36	-	5	283	50	140	1.70	2.57	-	9.86	1.70	1.19	2.39	-	6.10	0.63	1.99	-	3.36
W06	3	20	16.7	9.7	7.5	4.2	3.9	5	302	226	316	0.15	0.87	8.67	1.53	0.15	0.15	0.84	4.73	1.52	0.12	0.72	5.96	1.16
W07	2	22	9.2	3.7	0.64	0.8	-	12	324	315	405	-	2.27	2.82	22.86	2.27	0.57	1.55	1.62	2.97	0.34	1.28	2.44	2.21
W08	2	23	12	7.9	2	1.9	-	3	317	332	422	0.76	1.55	11.75	8.99	0.76	0.51	1.40	4.51	5.09	0.29	1.14	6.29	2.84
W09	2	18	8.4	6	0.9	2.5	-	8	60	335	425	0.38	1.48	4.98	2.23	0.38	0.29	1.30	2.87	1.56	0.22	1.12	4.08	1.20
W10	2	23	5.7	3.3	1.3	0.85	-	5	329	313	403	0.30	1.16	7.41	2.53	0.30	0.12	0.71	2.30	0.71	0.11	0.68	5.84	0.75
W11	3	22	1.1	2	0.1	0.64	0.1	4	327	120	210	1.13	1.74	19.08	4.97	1.13	0.12	0.69	2.37	0.51	0.07	0.49	10.57	0.73
W12	2	25	3.9	4	0.74	0.96	-	12	355	297	387	0.64	1.44	2.78	10.36	0.64	0.15	0.82	1.48	1.81	0.14	0.75	2.70	1.61
W13	2	12	11.9	10.9	3	2.28	-	7	36	73	163	1.06	2.77	7.28	9.39	1.06	0.99	2.71	5.63	9.02	0.70	2.41	4.93	5.65
W14	3	19	13	5	0	1.1	0	8	296	73	163	-	2.67	6.40	12.57	2.67	3.75	2.28	3.09	5.15	0.68	1.75	4.41	2.94
W15	2	18	22	6	8.3	0	-	8	351	200	290	0.70	1.84	9.74	2.93	0.70	0.60	1.75	5.03	2.83	0.39	1.50	5.79	2.34
W16	3	11	5.2	7.6	0	2.9	0	13	42	144	234	1.09	3.04	3.46	3.65	1.09	1.01	2.96	2.84	3.45	0.62	2.45	2.98	2.45
W17	3	7	8	2	0	0.56	0	20	30	156	246	7.71	6.72	3.07	6.83	3.07	3.40	5.87	2.29	4.49	1.48	4.70	2.81	2.86
W18	2	12	8.5	4.5	1.61	1.27	-	2	252	253	343	0.97	2.66	20.49	7.05	0.97	0.75	2.46	10.06	4.50	0.50	2.08	8.90	2.82
W19	2	15	4.2	5.2	1.6	0.68	-	5	28	56	146	0.75	2.12	8.71	5.49	0.75	0.48	1.80	4.17	3.66	0.31	1.48	5.79	2.24
W20	3	15	1.8	1.7	0.23	0.5	0.3	4	20	63	153	7.96	2.95	9.44	6.08	2.95	0.29	1.43	3.39	0.87	0.18	1.07	7.12	1.03
W21	3	20	18.9	9	0	2	0	7	348	71	161	-	2.51	4.96	12.25	2.51	-	2.36	3.31	7.48	1.15	1.90	3.58	3.95
W22	2	7	5.4	5.7	1	1.65	-	6	294	53	143	1.53	4.48	-	5.78	1.53	1.44	4.38	-	5.37	1.00	3.81	-	3.88

Note: (-) means the value is not existing. NS-Natural scenarios, RS-Rainfall scenarios, ES-Earthquake scenarios. When there is no tensile stress in the mudstone foundation, Fos_{te} has no value. For the case of an anaclinal slope, blocks do not slide and Fos_{sl} has no value. Both parameters are replaced by "-".

333 **5 Discussion**

334 **5.1 Characteristics of rock block stability Distribution Characteristics of Fos**

335 There are up to 12 results of Fos per potential unstable block with the consideration of three scenarios and four failure
 336 modes (i.e., partial damage and overall failure). Most of Fos_{te} values are less than 1 in any-all scenarios (purple points in
 337 Fig.10), except for two blocks (i.e., W17 and W20), whose Fos_{te} values are also close to 1 under rainfall or earthquake
 338 scenarios. Although most of Fos_{co} values (green points in Fig. 10) are greater than 1, they are closer to the critical state of
 339 $Fos = 1$ than Fos_{sl} and Fos_{to} (represented by blue and yellow points in Fig. 10, respectively). The compression damage of
 340 the exposed mudstone can be investigated in the field survey (Fig. 4d). However, it is difficult to observe the phenomenon of
 341 tensile damage inside the mudstone base. In the case of weak tensile strength, the mudstone base suffers from tensile failure,
 342 and compression failure usually occurs before tension failure. According to the results, their Fos_{te} and Fos_{co} are less than 1
 343 or close to 1, which means that the underlying mudstone has been partially damaged due to slight compressive or tensile
 344 failure, and the blocks are potentially unstable with the current depth of the basal cavity. However, most of the blocks do not
 345 exhibit the overall failure, and they still exist on the slope. Meanwhile Moreover, their Fos_{sl} and Fos_{to} values are greater
 346 than 1 in different scenarios, which is consistent with the this actuality. Due to the uncertainty of mechanical parameters, The
 347 results indicate it is possible that most of the blocks are close to a critical state, in which they are partially damaged but the
 348 whole block is still stable.

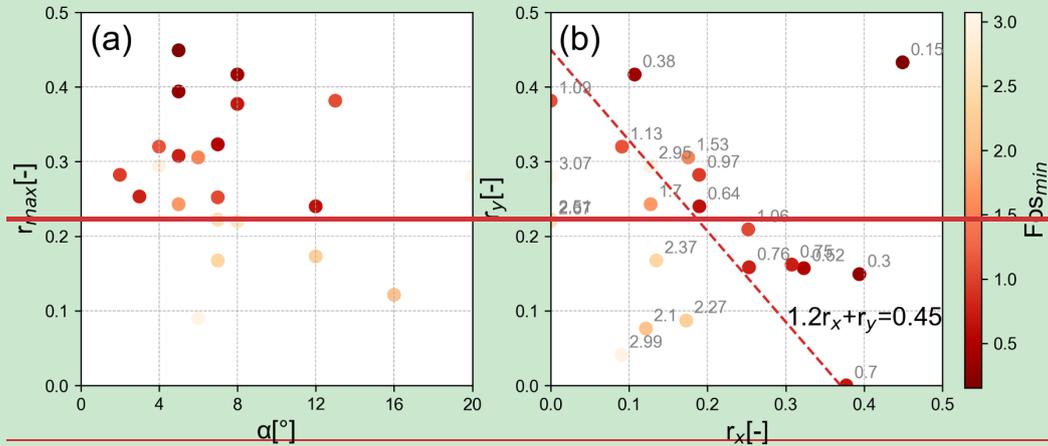


349 **Figure 10** Distribution of Fos in different scenarios. Shapes represent different scenarios and colours represent different failure
 350 modes.
 351

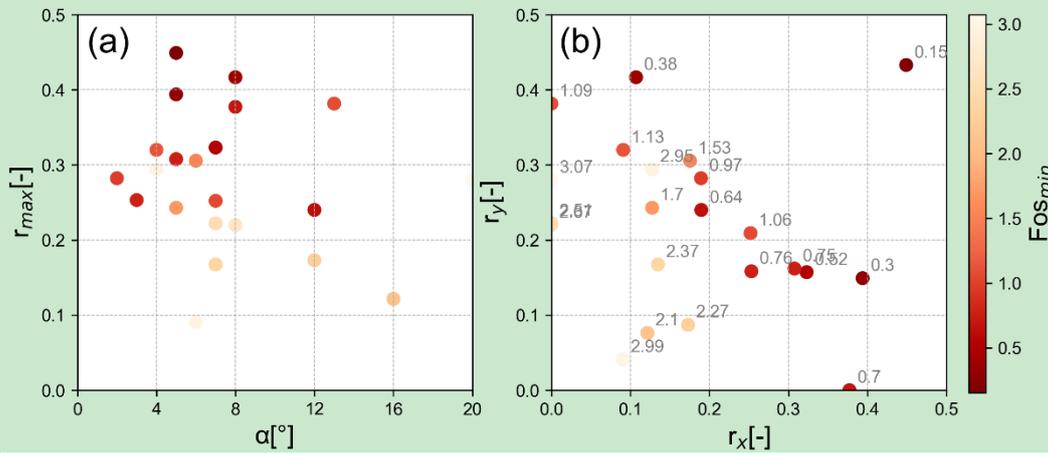
352 **5.2 Relationship between Fos and geometric parameters**

353 Fig. 11 presents the relationship between Fos_{min} and two main geometric parameters, the dip of the contact surface and the
 354 retreat ratio. In general, the dip angle of the bedding plane contact surface (α) is the key factor influencing the sliding failure
 355 mode. The horizontal axis in Fig. 11a is α between the rock blocks and underlying mudstone. Most of the points in Fig. 11a
 356 are in the interval $[0, 8^\circ]$, which is consistent with the features of sub-horizontal stratum-strata in the study area. The shade of
 357 the points does not change significantly in the x -axis direction, as Fig. 11a shows, which indicates that the dip of contact
 358 surface has little correlation with rockfall stability in this area. Therefore, compared with the maximum retreat ratio (r_{max}),
 359 the dip of the contact surface has less influence on rockfall stability in the study area. There was a significant positive
 360 correlation between the retreat ratio (r_{max}) and Fos_{min} : as Fig. 11b shown. In Fig. 11b, as the retreat ratios increase in the
 361 positive direction of the x -axis and y -axis, the rock blocks show a notable tendency to be unstable, all the points can be
 362 clearly divided into two parts by a red dashed line, Fos_{min} of the points in the upper part are all lower than the critical state
 363 ($Fos = 1$). Therefore, if r_x and r_y can be obtained through the detailed field investigation, the block stability can be
 364 preliminarily determined by the formula in the Fig. 11b.

365



366



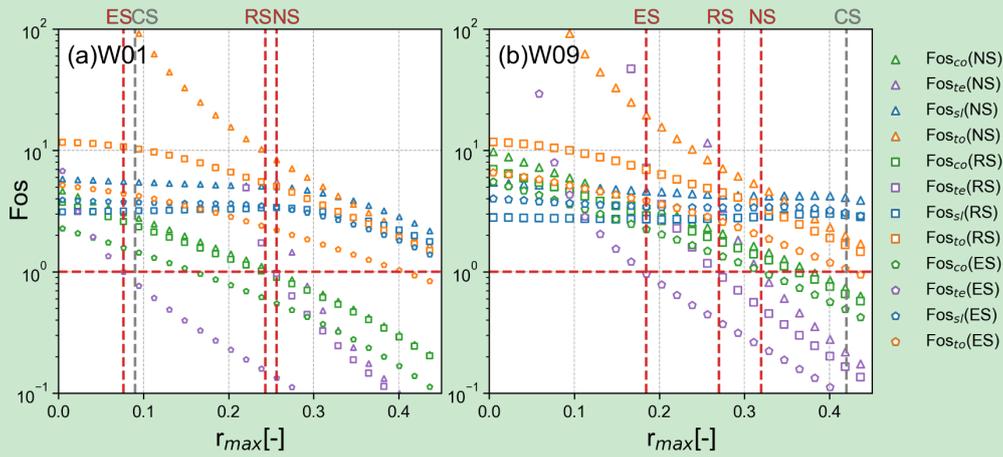
367 **Figure 11** ~~The relationship~~ Correlation between Fos and ~~main geometric parameters~~ the dip of contact surface and retreat ratio. Here, α is
368 the dip angle of the contact surface between rock block and underlying mudstone, r_x and r_y are the retreat ratio along x direction and y
369 direction, respectively, equal to d_1/a and d_2/b , and r_{max} is the larger ~~one~~ of r_x and r_y .

370 5.3 Definition of rockfall susceptibility

371 ~~In order to~~ To explore the variation ~~of~~ in Fos with the progressive erosion process of the cavity on the blocks, the cavity
372 retreat velocities in different directions are assumed to be equal (5 mm/year, [Zhang et al. \(2016\)](#)). ~~The rockfall evolution also~~
373 ~~can be well displayed from Fig. 12.~~ Fig. 12 shows the variations in Fos of two specific blocks during the evolution process
374 of the mudstone cavity. The instability of the blocks starts from the failure (or damage) of the foundation. In the initial stage,
375 the cavity is small, and the overlying block is stable; all Fos values are greater than 1.0. The cavity expands over time as the
376 mudstone weathers; then, the contact area decreases, and non-uniform distributed stress arises. When the stress exceeds the
377 ultimate strength of mudstone in a partial area, Fos_{co} and Fos_{te} decrease significantly, as shown in Fig. 12. The instability
378 of the blocks starts from the failure (or damage) of the foundation. Fos_{te} and Fos_{co} reach the critical state much earlier than
379 Fos_{sl} and Fos_{to} . For these two specific blocks, when r_{max} increases to 0.4, Fos_{sl} and Fos_{to} are still higher than 1.0. This
380 means that the rock blocks can remain globally stable in this condition.

381 These results further elucidate the stability analysis model proposed in this study. Fos_{co} and Fos_{te} introduced in this model
382 present the damage state of basal mudstone caused by compressive and tensile stresses, which do not provide global
383 instability of the overlying block as sliding and toppling. However, Fos_{co} and Fos_{te} are important preliminary signs of
384 subsequent global failure of the rock block. The damage in the basal mudstone can significantly accelerate weathering and
385 prompt expansion of the cavity, which will lead to global failure. The lower Fos_{co} and Fos_{te} are, the lesser the safety margin
386 of the blocks. Therefore, the four Fos used in this study can provide a more comprehensive quantification of rockfall
387 stability.

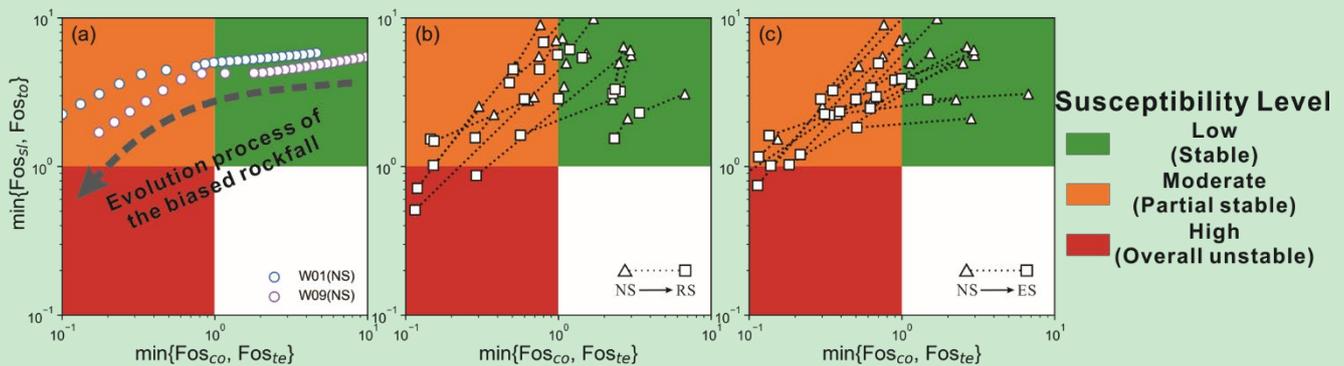
388 This result is consistent with Fig. 10, in which ~~90.9%~~ 63.7% of the purple and green points (Fos_{te} and Fos_{co}) are ~~near the~~
389 ~~line of $Fos = 1$ located between $Fos = 0.7$ and $Fos = 2.0$.~~ This result ~~also well agrees with the field insight, that is most~~
390 ~~rock blocks are potential unstable and many fractures appear in the mudstone~~ can be validated by the field phenomena. In the
391 study area, rock damage (e.g., micro-fractures and cleavages) can be observed in the underlying mudstone. However, most
392 overlying rock blocks are stable at the present time. This means that Even if Fos_{sl} or Fos_{to} is higher than 1, ~~in fact~~ its
393 foundation has begun to be damaged. In the case of heavy rain or earthquakes, Fos_{sl} and Fos_{to} may be reduced to less than
394 1, and the rockfall ~~will~~ occurs.



395

396 **Figure 12** Variation of Fos with r_{max} . (a) and (b) are the results for W01 and W09, respectively, which represent the situation of the
 397 blocks with two and three free faces. The grey dotted line (CS) approximately represents the current state of the unstable blocks. The
 398 red dotted lines correspond to the critical values of r in different scenarios.

399 Based on the above analysis meaning of four Fos , rockfall susceptibility can be divided into three levels. When both Fos_{co}
 400 and Fos_{te} are greater than 1, the overall rock block is stable, and the mudstone base is not damaged, which is defined as
 401 “low susceptibility” and represented by the green area in the Fig. 13. With the development of the cavity erosion, when
 402 Fos_{co} or Fos_{te} is less than 1 and Fos_{sl} and Fos_{to} are higher than 1, the base begins to undergo be damaged, and the
 403 overlying sandstone blocks still maintain remain relatively stable. This state is defined as “moderate susceptibility” and
 404 represented by the orange area. When Fos_{sl} or Fos_{to} is less than 1 in some scenarios, the rock blocks are in a “high
 405 susceptibility” state, which means that rockfalls are highly likely to occur. Fig. 13a indicates that along with the increase of
 406 in the cavity retreat ratio, the susceptibility of W01 and W09 changes from low susceptibility to moderate susceptibility in
 407 the natural scenario. As Fig. 13b and c show, when rainfall or earthquake occurs, Fos_{sl} or Fos_{to} of some blocks are less
 408 than 1, which means that some blocks have evolved to the state of high susceptibility and the overall sandstone blocks are
 409 unstable.



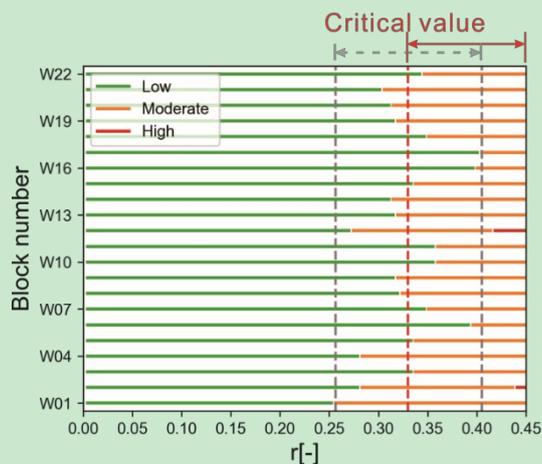
410

411 **Figure 13** Rockfall susceptibility based on the combination of four Fos distribution. The susceptibility is defined as three levels,
 412 represented by red, orange and green respectively. (a) shows the progressive failure process of the rock block changing from low

413 susceptibility to moderate susceptibility, as the ~~increase of~~ cavity retreat ratio increases (illustrated by W01 and W09 in the natural
 414 scenario. (b) and (c) show the change ~~of in~~ susceptibility of biased rock blocks, when the scenario changes from natural conditions s
 415 rainfall and earthquake conditions.

416 5.4 Critical retreat ratio in the study area

417 The cavity plays an important role in the progressive failure process of biased rockfall. ~~In order to analyze~~ analyze the effect
 418 of the retreat ratio on the stability of rock blocks, all blocks in the study area were selected to calculate d their Fos and
 419 susceptibility level with the ~~increased-increasing~~ r , whose retreat ~~velocity-velocities~~ in different directions are assumed to be
 420 equal. Fig. 14 shows that along with the increase ~~of in the~~ retreat ratio, the susceptibility level of rock blocks changes from
 421 low to moderate susceptibility. Corresponding to the critical state of $\min \{Fos_{co}, Fos_{te}\} = 1$ of all blocks, the minimum
 422 retreat ratio is 0.26565, and the maximum retreat ratio is 0.41050, ~~which are as~~ marked by the vertical ~~graygrey~~ dotted line in
 423 ~~the~~ Fig. 14. According to the ~~statistics-statistical~~ analysis of critical retreat ratios, ~~both the~~ mean and median are 0.3318 and
 424 0.3308, respectively. Therefore, the critical retreat ~~ratio~~ rate of the rock blocks in the study area ~~is can be~~ determined ~~to be as~~
 425 0.33, which is marked by the vertical red dotted line in the Fig. 14. ~~This result could provide evidence for field investigation~~
 426 ~~and regional risk control of biased rockfall. The critical retreat ratio calculated by this method can be used for the~~
 427 ~~preliminary identification of potential unstable rock blocks in a specific area, which can help concentrate limited risk~~
 428 ~~treatment resources on these priorities. The rock blocks with a retreat ratio exceeding 0.33 can be regarded as probable~~
 429 ~~rockfall source, which should be prevented with the highest priority. For unstable rock blocks in sub-horizontal formation,~~
 430 ~~filling the cavity with a support structure is an effective mitigation measure. It should be emphasized that the mechanical~~
 431 ~~parameters and analysis scenarios significantly affect the critical value. Therefore, the elaborative risk control of a given~~
 432 ~~rockfall should be arranged based on its specific parameters and analysis scenarios.~~



433
 434 **Figure 14** ~~The e~~Effect of the retreat ratio (r) on the Fos of the rock block, which is illustrated by all blocks in the study area. ~~The critical~~
 435 ~~value of r is 0.33.~~

436 5.5 Limitations

437 This study involves the development of an analytical model for the three-dimensional stability of biased rockfall, combining
438 the basic LEM method and the consideration of the eccentricity effect. Due to the complexity of rock structure and force
439 analysis, it is necessary to highlight the limitations of this model.

440 First, this study uses a three-dimensional coordinate system and bending theory. It is difficult to consider diverse shapes of
441 rock blocks, and the rock block was simplified as a prismatic column. The assumption of fully persistent discontinuities may
442 underestimate the stability of rock blocks, and ignores the stress transmission in joints or rock bridges. Then, following the
443 basic framework of the general LEM method, this study assumed that the rock is not subjected to deformations. The
444 complete stress-strain behaviour, such as the deformation in the mudstone layer, is not considered in this study. Furthermore,
445 the block stability is strongly influenced by the uncertainty of mechanical parameters. However, because of the difficulties in
446 sampling strongly weathered mudstone, it is difficult to obtain adequate parameter values for uncertainty statistics. These
447 limitations will be important considerations in future studies.

448 **6 Conclusion**

449 ~~Rockfall usually causes amounts of monetary damage and death in mountainous area. For the biased rockfall in sub-~~
450 ~~horizontal formations, the traditional LEM method usually overestimates the stability of rock blocks with natural cavities.~~
451 ~~Due to differential weathering in sub-horizontally interbedded of hard rock and soft rock, multi-layer biased rockfalls~~
452 ~~develop on steep slopes. In mountainous ranges, cut slopes, and coastal cliffs, rockfall may cause significant facility damage~~
453 ~~and casualties in residential areas and transport corridors.~~ The aim of this study was to present a new three-dimensional
454 analytical method for the stability of rock blocks with basal ~~cavity~~cavities. ~~In the geological model, In this method,~~ a non-
455 uniform distributed ~~force~~stress due to the eccentricity effect is applied at the contact surface, ~~in place of instead of~~ a point
456 force. The method considers four failure modes according to the rockfall evolution process, including partial damage of the
457 soft foundation (Fos_{co} and Fos_{te}) and overall failure of the rock block (Fos_{sl} and Fos_{to}).

458 Taking the northeast edge of the Sichuan Basin in Southwest China as the study area, The proposed method was-is used to
459 calculate the Fos of the typical unstable rock blocks in the study area biased unstable rock blocks. The results are consistent
460 with the natural states of the rock blocks show that in the natural scenario, the underlying mudstone of some rock blocks has
461 been partially damaged, and compression failure of the mudstone has been observed in the field. Some rock blocks are
462 expected to fail as a whole in rainfall or earthquake scenarios. Besides, The statistical analysis indicates that the retreat ratio
463 is the crucial factor influencing the Fos of biased rockfall. On the basis of different combinations of four Fos , rockfall
464 susceptibility was classified into three levels. As the retreat rate increases, the rock blocks undergo an evolution process from
465 stability to partial instability and then overall instability. Based on the current mechanical parameters of the eastern Sichuan
466 Basin, The critical retreat ratio from low to moderate rockfall susceptibility is 0.33.

467 The proposed method improves the three-dimensional geomechanical model of a rock block with a basal cavity; by
468 considering non-uniform distributed force-stress at the contact surface, which could promote the accuracy of rockfall
469 stability analysis. ~~However, Due to the assumptions adopted and because of the complexity of mechanical the failure~~
470 ~~mechanism of biased rockfall and the assumptions adopted in the method, it is essential to highlight the limitations of this~~
471 ~~method. In the proposed method, the contact surface between rock block and underlying mudstone is assumed to be a~~
472 ~~rectangle. However, in reality its natural shape is irregular, which results in complex distribution of supporting force. In~~
473 ~~addition, in the geomechanical model, the failure surface of biased rockfall is set to be the contact surface of rock block and~~
474 ~~mudstone. However, the natural failure surface may be formed along the cleavages in the mudstone, which will lead to the~~
475 ~~changes in mechanical parameters of stability analysis there are some limitations in this method, mainly including the~~
476 ~~simplification of boundary conditions and rock deformation. Further research is clearly needed for the improvement of~~
477 ~~geomechanical model of biased rockfall. These limitations will be important considerations in future studies.~~

478 **Data availability**

479 All raw data can be provided by the corresponding authors upon request.

480 **Author contributions**

481 XS, BC and JD planned the campaign; XS and BC performed the field measurements; XS, BC, WW and BL designed and
482 developed the methodology. XS, BC and JD analysed the data; XS and BC wrote the manuscript draft; JD and WW
483 reviewed and edited the manuscript.

484 **Competing interests**

485 The authors declare that they have no conflicts of interest.

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