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Dynamic analysis of soil-nailed slope

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An analytical method based on the kinematic limit approach is presented for stability analysis of nailed soil slopes. Earthquake effects were considered in an approximate manner in terms of seismic coefficient-dependent horizontal forces. Two kinds of failure surfaces, a planar failure surface and a circular failure surface, were considered in this study. The proposed method can be viewed as an extension of the method of slices, but it provides a more accurate treatment of forces because they are represented in an integral form. The factor of safety and nail forces obtained by the proposed method were found to be in good agreement with the published results for an 8 m high vertical soil-nailed wall.

Notation

Basic SI units are given in parentheses.

- c cohesion of soil (N/m^2)
- F_s factor of safety (dimensionless)
- *G* weight of soil wedge (N/m)
- *H* height of slope (m)
- *h* height of elementary slice (m)
- *K*_h horizontal seismic acceleration coefficient (dimensionless)
- $K_{\rm t}$ average tensile strength of reinforcement (N/m)
- N normal force on the base of slice (N/m)
- *n* number of reinforcement layers (dimensionless)
- *R* maximum shear resistance acting on the base of slice (N/m)
- *S* shear force on the base of slice (N/m)
- T_i tensile strength of reinforcement per unit horizontal spacing (N/m)
- $T_{\rm max}$ maximum axial force in nail (kN)
- *t* thickness of rupture layer (mm)
- [*v*] velocity jump vector (dimensionless)
- α angle of inclination of reinforcement (°)
- β slope angle (°)
- γ unit weight of soil (kN/m³)
- $\dot{\epsilon}$ strain rate (dimensionless)
- θ angle of failure plane (°)
- λ non-negative scalar multiplier (dimensionless)
- ξ angle of inclination of reinforcement with velocity of discontinuity (°)
- ϕ soil friction angle (°)
- $\dot{\omega}$ angular velocity of rotation (rad/s)

1. Introduction

Soil nailing is one of the recent techniques available for stabilising in situ soil slopes or cuts. The process of soil nailing includes installation of nails in excavated cuts or in slopes either by driving or grouting in predrilled holes. The stability of the slope face between nails is ensured by providing thin layers of shotcrete reinforced with wire mesh. The nails are generally steel bars, metal tubes or other metal rods that can resist not only tensile force but also shear stress and bending moment. The nailing method has been used in both granular and cohesive soils and in relatively heterogeneous deposits.

The general consensus among practising engineers is that soilnailed walls perform reasonably well under seismic conditions. However, documented performance of soil-nailed systems is almost non-existent in the literature. Most of the literature (Gassler, 1988; Koseki et al., 1998; Matsuo et al., 1998; Sakaguchi, 1996; Sivakumar Babu and Singh, 2008) on nailed soil structures emphasises the mechanism of reinforcement and the design of structures under static load only. The failure mechanisms for slopes under static loading have typically been extended to stability analysis of slopes under seismic loading using a pseudo-static approach (Mononobe and Matsuo, 1929), but the effects of seismic excitation on the failure pattern of slopes have not been addressed. Only a handful of full-scale and model tests in the laboratory have been conducted on a soilnailed system (Giri and Sengupta, 2009; Tufenkjian and Vucetic, 2000; Vucetic et al., 1993). Consequently only limited data on the earthquake resistance and corresponding failure mechanisms for steep nailed slopes are available to date. The seismic

resistance and the failure mechanism of a nailed soil slope during an earthquake event are not clearly understood and need to be investigated properly.

There are several methods currently available for the design of soil-nailed slopes, such as the German method (Stocker et al., 1979), the Davis method (Mitchell and Villet, 1987), the method developed by Gassler and Gudehus (1981), the French method (Schlosser, 1982), and the finite-element method (Dawson et al., 1999, Griffiths and Lane, 1999). The first four methods are based on the limit equilibrium approach whereas the fifth one is based on the limit analysis. The finite-element method is certainly the most comprehensive approach to analyse the performance of soil structures subjected to seismic loading. However, it requires accurate measurement of the properties of the component materials, which is often difficult to achieve. In addition, further difficulties arise from modelling failure in frictional materials. The majority of the existing methods of design and analysis of a soil-nailed system are based on the pseudo-static approach, where the effect of earthquake on a potential failure soil mass is represented by a horizontal force acting at the centre of gravity. The horizontal force is calculated as the product of a seismic intensity coefficient and the weight of the potential sliding mass. The stability of soil structures under this force is expressed by a factor of safety which is usually defined as the ratio of the resisting force to the destabilising force. Failure occurs when the safety factor drops below 1. However, pseudo-static analysis is generally considered conservative, since even when the factor of safety drops below 1, the soil structures could experience a finite displacement rather than a complete failure.

An alternative analytical method is presented here based on the kinematics theorem of limit analysis to study the stability of reinforced slopes under the seismic loading condition. The kinematic theorem (Juran *et al.*, 1988, 1990) states that slopes will collapse if the rate of work done by external loads and body forces exceeds the energy dissipation rate for any assumed kinematically admissible failure mechanism. Soil deformation is assumed to be plastic and failure is associated with Coulomb's yield condition.

In the present approach, the following assumptions were made.

- (a) The effect of pore pressure build-up and change of soil strength due to earthquake shaking were ignored. In most cases, weep holes are provided in the walls for safe drainage of water from behind the wall. Furthermore, the backfill soil is usually chosen as cohesionless and free-draining material.
- (b) The slope is made of homogeneous cohesionless soil. Strictly speaking, soil is not a homogeneous material, but it is often assumed to be one, only to simplify the calculations in hand.
- (c) The reinforcement layers are finite in number and have the same length. Seldom, nails of different length and dimension are utilised in a wall construction.
- (d) The resistance to shear, bending and compression is ignored.

The nails are typically not designed for bending and compressional loads.

- (e) The critical failure surface is assumed to pass through the toe of the slopes. For the shallow failure surfaces, spalling of the face wall is ignored as they may not necessarily represent the critical case.
- (f) The available pullout resistance is assumed to be either the bond strength between soil and reinforcement or the tensile strength of the reinforcement, whichever is smaller. This is a very common assumption in the traditional design of bond length.

Under the above assumptions, the reinforcements provide tensile forces acting in the horizontal direction. The rate of external work is due to soil weight and inertia force induced by the earthquake and the only contribution to energy dissipation is that provided by the reinforcement.

The possible failure modes considered are illustrated in Figures 1 to 3.

Figure 1 shows a rotational mechanism involving a circular failure surface passing through the toe of a slope of height, H, angle, β , and with reinforcements of uniform length, $L_{\rm T}$. The circular failure surface may extend within (in Figure 1(a)) and also beyond (in Figure 1(b)) the reinforced zone. Figure 2 shows a translational slope failure mechanism. Figure 3 shows the direct sliding mechanism, in which the reinforced soil mass slides over the bottom layers. Let us assume that a rigid plastic body occupies a domain V, and v is a velocity field on V with a discontinuous boundary S. When a particle tends to move to Sfrom the negative side to the positive side of the velocity field, a unit velocity vector normal to S can be defined as a velocity jump vector (Jacov, 1996). As shown in Figures 1 to 3, the velocity jump vector, v, makes an angle equal to the angle of friction ϕ with the failure surface(s). The slope is unstable when one of the above-considered failure mechanisms occurs.

2. Methods of analysis

An analytical method based on the kinematics limit equilibrium (Juran *et al.*, 1988, 1990) is presented for the stability analysis of nailed soil slopes. It is assumed in kinematic approach of limit analysis that the soil and the reinforcements are perfectly plastic and their deformation is governed by associated flow rule.

Mathematically,
$$\dot{\varepsilon}_{ij}^{p1} = \lambda \frac{\partial f(\sigma_{ij})}{\partial \sigma_{ij}}$$

1. $\lambda \ge 0$ if $f = 0$ and $\lambda = 0$ if $f < 0$

Where $\dot{\epsilon}_{ij}^{p1}$ is the plastic strain rate tensor in a kinematically admissible velocity field, σ_{ij} is the stress tensor associated with strain rate tensor, λ is a non-negative scalar multiplier and $f(\sigma_{ii}) = 0$ is the yield criteria. Mohr–Coulomb failure criteria





Figure 1. Circular failure surface: (a) extended within reinforced zone; (b) extended beyond reinforced zone



Figure 2. Translational slope failure mechanism



Figure 3. Direct sliding mechanism

are assumed and the discontinuity vector (velocity jump) is assumed to be inclined to the rupture surface at the angle of internal friction, ϕ .

The kinematics theorem of limit analysis (Juran *et al.*, 1988, 1990) states that when the rate of work done by the external forces and the body forces exceeds the rate of internal dissipation energy, the structure will collapse. This theorem is usually referred to as the upper bound theorem, because it allows the calculation of an upper bound to the limiting force causing failure.

Michalowski (1998) represented this theorem mathematically as

2.
$$\int_{V} D(\dot{\varepsilon}_{IJ}) \mathrm{d}V \ge \int_{S} T_{i} v_{i} \mathrm{d}S + \int_{V} \gamma_{i} v_{i} \mathrm{d}V$$

The left-hand side of the equation represents the rate of energy dissipation (D) during an incipient failure of a structure and the right-hand side includes the rate of work done by all the external forces. In the above equation, T_i is the stress vector on the boundary S, v_i is the velocity vector in the kinematically admissible mechanism. γ_i is the specific weight and V is the volume of the mechanism (refer to Figure 4). The mathematical form of the theorem states that the rate of energy dissipation is not less than the rate of work done by external forces in any kinematically admissible failure mechanism. The total force on the boundary, S, can be calculated only if velocity, v_i , on this boundary is constant. If the geometry of the structure is given and all loads and material parameters are known, the safety factor can be calculated. Earthquake effects are considered in terms of seismic coefficient-dependent horizontal forces. Two kinds of failure surfaces were considered in this study, namely a planar failure surface and a circular failure surface. The proposed method is an extension of the method of slices (Duncan and Wright, 2005), but it provides a more accurate treatment of forces because they are represented in an integral form.





2.1 Planar failure surface

Failure surfaces in homogeneous or layered non-homogeneous sandy slopes are essentially planar. A planar failure surface may occur in a slope where permeable soils such as sandy soil and gravel or other permeable soils with some cohesion exist and where shear strength is principally provided by friction. For cohesionless sandy soils the planar failure surface may occur if strong planar discontinuities develop, as in the soil beneath the ground surface in natural hillsides or in a man-made cutting.

Figure 5(a) shows a plane failure surface, AB for a slope with height, *H* and slope angle β . In the elementary slice, defg, as shown in Figure 5(b), the self-weight of the slice is given by $dG = \gamma h dx$ where γ is the unit weight of soil and *h* is the height of the slice. This method assumes that resultant inter-slice forces X_1 and X_r acting on de and fg, respectively, are equal and opposite and $\sum (X_1 - X_r)$ may be zero for the whole failure wedge. The inter-slice shearing forces acting on faces de and fg are opposite and parallel to each other and may be neglected when equilibrium of the total section is considered. The normal and shearing forces on the sliding surface, ef, are given by

3.
$$N = \gamma h dx \cos \theta - K_h \gamma h dx \sin \theta$$

4.
$$S = \gamma h dx \sin \theta + K_h \gamma h dx \cos \theta$$

where θ is the angle of the failure plane and K_h is the horizontal seismic coefficient.

The normal force, N, can produce a maximum shearing resistance, R, given by

5.
$$R = c dx \sec \theta + \gamma h dx (\cos \theta - K_{\rm h} \sin \theta) \tan \phi$$







Figure 5. (a) Planar failure mechanism; (b) forces acting on elementary slice defg

The equation of line AB in Figure 5(a) is, $y_1 = x \tan \beta$. The equation of line AC is, $y_2 = x \tan \theta$ and that of line CB is $y_3 = H$.

The total rate of work done is given by

6.

$$\dot{W} = \gamma v \cos \phi \left\{ \sin \theta \left[\int (y_1 - y_2) dx + \int (H - y_2) dx \right] \right\}$$
$$+ K_h \cos \theta \left[\int (y_1 - y_2) dx + \int (H - y_2) dx \right] \right\}$$

$$= \gamma v \cos \phi \left\{ \sin \theta \left[\int_{0}^{l} (x \tan \beta - x \tan \theta) dx + \int_{l}^{L} (H - x \tan \theta) dx \right] \right.$$
$$\left. + K_{h} \cos \theta \left[\int_{0}^{l} (x \tan \beta - x \tan \theta) dx + \int_{l}^{L} (H - x \tan \theta) dx \right] \right\}$$

The above expression can be simplified to

8. $\dot{W} = v \cos \phi (\sin \theta + K_{\rm h} \cos \theta) G$

where G is the weight of failed soil wedge and given by

9.
$$G = \frac{\gamma H^2}{2} (\cot \theta - \cot \beta)$$

The rate of internal energy dissipation due to cohesion and tensile force of reinforcement may be expressed as

$$\dot{D} = Rv\cos\phi + v\cos\phi\sum T_i\cos(\theta + \alpha)$$
10. $+ v\sin\phi\sum T_i\sin(\theta + \alpha)$

where $\sum T_i$ is the sum of tensile reinforcement force per horizontal spacing of reinforcement.

Substituting for R and on simplifying

$$D = \frac{cH}{\sin\theta} + Gv\cos\phi(\cos\theta - K_h\sin\theta)\tan\phi$$
$$+ v\cos\phi\sum T_i\cos(\theta + \alpha)$$
11.
$$+ v\sin\phi\sum T_i\sin(\theta + \alpha)$$

Now factor of safety can be expressed as a ratio of Equations 11 and 8 as

$$F_{\rm s} = \frac{\frac{cH}{\sin\theta} + G(\cos\theta - K_{\rm h}\sin\theta)\tan\phi}{G(\sin\theta + K_{\rm h}\cos\theta)} + \frac{\sum T_i\cos(\theta + \alpha) + \tan\phi\sum T_i\sin(\theta + \alpha)}{G(\sin\theta + K_{\rm h}\cos\theta)}$$
12.

2.2 Circular failure surface

Figure 6 shows a potential circular failure surface AB with centres (a, b) and radius *r*. Earthquake effect is approximated by a horizontal force equal to K_hG acting through the centre of gravity of the soil wedge. A number of vertical slices are assumed. The free body diagram of a typical vertical slice defg is shown in Figure 6. The self-weight of the slice is given by $dG = \gamma hdx$, where dx is the elemental width of slice. The interslice forces X_1 and X_r acting on de and fg, respectively, are equal



Figure 6. Circular failure mechanism

and opposite and parallel to the base of slice ef. $\sum (X_1 - X_r)$ is zero for the whole failure mass.

Resolving the forces acting on the base of slice ef

13.
$$dN = \gamma h dx \cos \theta - K_h \gamma h dx \sin \theta$$

14. $dS = \gamma h dx \sin \theta + K_h \gamma h dx \cos \theta$

15. where
$$\theta = \sin^{-1} (x - a)/r$$
 and $r = \sqrt{a^2 + b^2}$

The force dN can produce a maximum shearing resistance dR given by

$$dR = cdx \sec \theta + \gamma h dx(\cos \theta - \sin \theta) \tan \phi$$

16.
$$dR = cdx \sec \theta + \gamma h dx(\cos \theta - \sin \theta) \tan \phi$$

where c is the cohesion of the soil.

The equation of line, AB, in Figure 6, is

17a. $y_1 = x \tan \beta$

The equation of line BC is

17b. $y_2 = H$

The equation of the circular failure surface AC is

17c.
$$y_3 = b - \sqrt{r^2 - (x - a)^2}$$

Then the total rate of work done can be equal to the sum of work done by d*S* and is given by

$$\dot{W} = \gamma v \cos \phi \left\{ \sin \theta \left[\int_0^l (y_1 - y_3) dx + \int_l^L (H - y_3) dx \right] \right.$$
$$\left. + K_h \cos \theta \left[\int_0^l (y_1 - y_3) dx + \int_l^L (H - y_3) dx \right] \right\}$$
$$18. \qquad = \gamma v \cos \phi (I_s + K_h I_c)$$

where

$$I_{s} = \int_{0}^{l} (y_{1} - y_{3})(\sin\theta dx) + \int_{l}^{L} (y_{2} - y_{3})\sin\theta dx$$

19. $= (H^{2}/2)[(a\cot\beta + b) - (H/3)\operatorname{cosec}^{2}\beta]$

$$I_{c} = \int_{0}^{l} (y_{1} - y_{3}) \cos \theta dx + \int_{l}^{L} (y_{1} - y_{3}) \cos \theta dx$$

$$= \frac{\tan \beta}{6r} [2r^{2} + (l - a)^{2}] \sqrt{r^{2} - (l - a)^{2}}$$

$$+ \frac{b \tan \beta}{r} \left(\frac{a^{2}}{2} + \frac{b^{2}}{3}\right) + \frac{r}{2} (a \tan \beta - H) \sin^{-1} \left(\frac{l - a}{r}\right)$$

$$+ \frac{r}{2} (a \tan \beta - b) \sin^{-1} \frac{a}{r} - \frac{r}{2} (b - H) \sin^{-1} \left(\frac{L - a}{r}\right)$$

$$+ \frac{1}{6r} [4r^{2}L - ab^{2} + (L - a)(H - a)^{2}]$$

where

$$l = H \cot \beta$$
 and
21. $L = a + \sqrt{r^2 - (b - H)^2} l = H \cot \beta$

The rate of dissipation of internal energy is due to shear resistance and is given by

$$\dot{D} = \int dRv \cos \phi$$

$$= v \cos \phi [\int_{0}^{L} c \sec \theta dx$$

$$+ \int_{o}^{l} \gamma(y_{1} - y_{3})(\cos \theta - K_{h} \sin \theta) \tan \phi dx$$

$$+ \int_{l}^{L} \gamma(y_{2} - y_{3})(\cos \theta - K_{h} \sin \theta \tan \phi) dx]$$
22.
$$= v \cos \phi [rc\phi + \gamma \tan \phi (I_{c} - K_{h}I_{s})]$$

where

23.
$$\phi = \sin^{-1} \left[(L-a)/r \right] + \sin^{-1} \left(a/r \right)$$

The rotational failure mechanism is represented by a circular

failure surface. For a given slope angle and internal angle of friction, the toe failure is fully described by two parameters ϑ_0 and ϑ_f as shown in Figure 6. Rupture of the reinforcement is interpreted as a plastic flow process consistent with the flow rule. The energy dissipation rate in a single reinforcement intersecting a velocity discontinuity can be derived assuming that the discontinuity is of finite-thickness, *t*, with a high-velocity gradient as shown in Figure 7.

The reinforcement contributes to the stability of the structures only through its tensile strength (reinforcement resistance to shear, torsion and bending is neglected). The kinematics requires that the velocity jump vector [v] be inclined to the velocity discontinuity at an angle of internal friction ϕ . The reinforcement is inclined to the velocity discontinuity at an angle ξ as shown in Figure 7. No reinforcement is assumed to be pulled out and sector PQ with length $(t/\sin\xi)$ (where t is the thickness of rupture layer) is subjected to plastic flow. The rate of energy dissipation in a single reinforcement intersecting a velocity discontinuity per unit horizontal spacing of reinforcement may be calculated as

$$D = \int_{0}^{(t/\sin\xi)} T_{1}\dot{\varepsilon} dx$$
24.
$$= T_{1}[v]\cos(\xi - \phi)$$

where T_1 is the limit tensile force in the reinforcement per unit horizontal spacing and $\dot{\varepsilon}$ is the strain rate in the direction of reinforcement.

The strain rate in the reinforcement is given by

$$\dot{\varepsilon} = [v] \frac{\cos(\xi - \phi)}{t \sin \xi}$$



The average strength K_t of reinforcement is given by

$$26. \quad K_{\rm t} = nT_1 \frac{\cos \alpha}{H}$$

where n is the number of reinforcement layers.

The energy dissipation rate per unit area of the discontinuity surface becomes

$$\dot{D} = \int_{0}^{t/\sin\xi} K_{t} \sin\xi dx\dot{\varepsilon}$$
27.
$$= K_{t}[v] \cos(\xi - \phi) \sin\xi$$

For a circular failure surface, the energy dissipation rate per infinitesimal length $(rd\vartheta)$ increment (refer to Figure 7) is given by

28.
$$d\dot{D} = K_t \dot{\omega} r^2 \int \cos(\vartheta + \alpha - \phi) \sin(\vartheta + \alpha) d\vartheta$$

where $\dot{\omega}$ is the rate of rotation.

29.

30.

Integrating from ϑ_0 to ϑ_f and assuming K_t to be constant

$$\dot{D} = K_{t} \dot{\omega} r^{2} \int_{\vartheta_{0}}^{\vartheta_{f}} \cos(\vartheta + \alpha - \phi) \sin(\vartheta + \alpha) d\vartheta$$

$$= K_{t} \dot{\omega} r^{2} (\cos \phi \{ [\sin^{2} (\vartheta_{0} + \alpha) - \sin^{2} (\vartheta_{f} + \alpha)]/2 \}$$

$$+ \sin \phi \{ [(\vartheta_{f} - \vartheta_{0})/2]$$

$$- [(\sin 2(\vartheta_{f} + \alpha))/4]$$

$$+ [(\sin 2(\vartheta_{0} + \alpha))/4] \})$$

The total rate of internal energy dissipation is the sum of cohesive and tensile reinforcement forces and is given by

$$\begin{split} \dot{D} &= v \cos \phi [rc\phi + \gamma \tan \phi (I_{\rm c} - K_{\rm h} I_{\rm s})] \\ &+ K_{\rm t} \dot{\omega} r^2 (\cos \phi \{ \left[\sin^2 (\vartheta_0 + \alpha) - \sin^2 (\vartheta_{\rm f} + \alpha) \right] / 2 \} \\ &+ \sin \phi \{ \left[(\vartheta_{\rm f} - \vartheta_0) / 2 \right] \\ &- \left[(\sin 2 (\vartheta_{\rm f} + \alpha)) / 4 \right] \\ &+ \left[(\sin 2 (\vartheta_0 + \alpha)) / 4 \right] \}) \end{split}$$

The factor of safety, F_s , is calculated by taking the ratio of Equations 30 and 18, and is given by

$$F_{s} = \frac{rc\psi + \gamma \tan \phi (I_{c} - K_{h}I_{s})}{\gamma (I_{s} + K_{h}I_{c})} + K_{t}r \left(\frac{\{[\sin^{2}(\vartheta_{0} + \alpha) - \sin^{2}(\vartheta_{f} + \alpha)]/2\}}{\gamma (I_{s} + K_{h}I_{c})} + \tan \phi \{[(\vartheta_{f} - \vartheta_{0})/2] - [(\sin 2(\vartheta_{f} + \alpha))/4] + [(\sin 2(\vartheta_{0} + \alpha))/4]\} + [(\sin 2(\vartheta_{0} + \alpha))/4]\}}{\gamma (I_{s} + K_{h}I_{c})} \right)$$
31.

It may be observed from Equation 31 that the factor of safety for a given slope is a function of parameters such as the coordinate of failure circle (a, b), angle ϑ_f and ϑ_0 , and angle α . Thus minimum value of F_s can be found using the minimisation technique.

3. Verification of the proposed method

The published results reported by Sivakumar Babu and Singh (2008) for a soil-nailed wall supporting a vertical cut of 8 m high under seismic condition were utilised here to verify the present method of analysis. The geometry of the soil-nailed wall supporting the vertical cut is shown in Figure 8. The 8 m high wall was designed in conventional manner by using the allowable stress design procedure. The soil nails were 4.7 m long and placed in 100 mm diameter drill holes. They were grouted in place at 1 m apart at an angle of 15° with the horizontal axis. The wall was analysed numerically by using the finite-element method. The seismic records from Bhuj (Iyengar and Raghu Kanth, 2002) and Uttarkashi (Chandrasekaran and Das, 1992) earthquakes were utilised in the reported pseudo-static and dynamic analyses of the wall. The external failure mode of the nailed soil wall in terms of global stability and sliding stability was studied under static, pseudo-static as well as dynamic conditions. In the present study, the published pseudo-static results reported in the paper were compared with the results of the present analysis. The horizontal seismic coefficient, $k_{\rm h}$, for the pseudo-static analysis corresponding to the Bhuj and Uttarkashi earthquake was adopted from the paper and given as 0.106 and 0.241, respectively. All other material parameters adopted from the literature (Sivakumar Babu and Singh 2008) and utilised in the present analysis are given in

Parameter	Value
Cohesion of the backfill soil, c: kPa	1.0
Friction angle of the backfill soil, ϕ : °	30.0
Unit weight of the backfill soil, γ : kN/m ³	16.0
Slope angle, β : °	0.0
Angle of inclination of nails, α : °	15.0
Nail length: m	4.7
Maximum axial tensile load capacity of nails, T_i : kN	83.44
Table 1. Material parameters for the soil-nailed wall	



Figure 8. The geometry of the soil-nailed wall supporting an 8 m vertical cut

Table 1. Since the critical failure surface in the backfill was not given in the referred paper, it was found by the minimisation technique and shown in Figure 8. The factor of safety corresponding to the failure surface was first calculated from Equation 31. The total rate of internal energy dissipated was reduced from the corresponding factor of safety value. The rate of internal energy dissipated due to tensile reinforcement force was obtained from the value of total energy dissipated. The tensile force in the reinforcements was calculated from Equation 26. In the above calculations, the velocity jump vector [v] was kept constant, as factor of safety value is independent of velocity jump. Table 2 shows the comparison between the published results and the results from the present theory. The factors of safety against global stability of the nailed wall for the Bhuj and Uttarkashi earthquakes were found to be 0.91 and 0.78, respectively. These values are comparable with the corresponding published results (0.95 and 0.81). The maximum axial force in the nails as predicted by the present method was found to be 32.4 and 37.98 kN, respectively, for the Bhuj and Uttarkashi earthquakes. These values are also comparable with the corresponding published results.

Parameters	Published results from Sivakumar Babu and Singh (2008)		Results from the present method of analysis	
	For Bhuj equation	For Uttarkashi equation	For Bhuj equation	For Uttarkashi equation
Maximum axial force in nail, T _{max} , (kN)	34.5	40.73	32.4	37.98
Factor of safety against global stability	0.95	0.81	0.91	0.78
Table 2. Comparison of present method of analysi published results for an 8 m high vertical soil naile	is with d wall			

4. Conclusions

An analytical method based on the kinematic limit approach has been developed for stability analysis of soil-nailed slopes. Two kinds of failure surfaces, a planar and a circular failure surface, are considered in the formulation. The proposed method can be viewed as an extension of the method of slices, but it provides a more accurate treatment of forces because they are represented in an integral form. The published result of a soil nail wall supporting a vertical cut of 8 m high is utilised to verify the performance of the new methodology. The maximum axial force in the nails and the factor of safety predicted by the proposed method are found to be in good agreement with the corresponding published results for the 8 m high soil-nailed vertical cut under seismic conditions.

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