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# Computer Methods and Advances in Geomechanics

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# Non-linear finite element analysis of safety factors

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**ABSTRACT:** In this paper a robust method is proposed to determine the safety factor of geotechnical constructions in finite element computations. The method is based on the reduction of the strength parameters of soil, the friction angle  $\phi$  and the cohesion  $c$ . Three examples show the practical application of the method.

## 1 INTRODUCTION

In structural engineering a factor of safety is always defined as the ratio of the collapse load over the the working load. The same definition is adopted in foundation engineering, at least for footings and piles. For soil bodies such as road/river embankments and earthen dams the situation is different. Here the dominating load is not a direct external force, but most of the load comes from soil weight. Very cohesive soil bodies can be loaded to collapse by increasing gravity, either numerically or in centrifuge tests, but not when the strength is dominated by friction. Therefore other definitions of safety are common in soil mechanics. The usual soil mechanics definition of safety is:

$$\text{safety factor} = \frac{S}{S_c} = \frac{c + \sigma' \tan \phi}{c_c + \sigma' \tan \phi_c}$$

where  $S$  is the shear strength, further defined by the well known Mohr-Coulomb criterion. Note that  $c$  is the cohesion,  $\phi$  is the friction angle and  $\sigma'$  is the normal pressure at the plane considered. Both  $c$  and  $\phi$  are effective strength parameters and  $\sigma'$  is an effective stress. We use the subscript  $c$  to indicate critical strength parameters; just high enough to ensure equilibrium. Indeed, for  $c=c_c$  and  $\phi=\phi_c$  the safety factor becomes equal to unity. The above definition of safety coincides with the classical definition as used in slip circle analysis on the condition that we define:

$$c_c / \tan \phi_c = \alpha c / \tan \phi \quad \text{with} \quad \alpha = 1$$

In this study the proportionality with  $\alpha=1$  will be retained in order to remain compatible to traditional slip circle analysis, but the method allows for other ratios of  $c_c$  and  $\tan \phi_c$ .

This study concentrates on the computation of the above safety factor by use of an elastic-plastic finite element method. In stead of the usual incrementation of loads, strength parameters will be decremented. This technique was first proposed by Zienkiewicz at al. (1975), but we make the procedure robust by adding an arc-length technique. A robust procedure was needed for implementing into the PLAXIS finite element package. Finally the potential of the method is demonstrated by considering a number of applications.

## 2 MOHR-COULOMB MODEL

Being interested in collapse loads rather than precise deformations, the elastic-plastic Mohr-Coulomb model is adopted. Hence

$$\underline{\dot{\sigma}} = \underline{D} (\underline{\dot{\epsilon}} - \underline{\dot{\epsilon}}^P), \quad \underline{\dot{\epsilon}}^P = \lambda \frac{\partial g}{\partial \underline{\sigma}}$$

where  $\underline{D}$  is the elasticity matrix and

$$\lambda = \frac{1}{d} \frac{\partial f^T}{\partial \underline{\sigma}} \underline{D} \underline{\dot{\epsilon}}, \quad d = \frac{\partial f^T}{\partial \underline{\sigma}} \underline{D} \frac{\partial g}{\partial \underline{\sigma}}$$

The symbols  $f$  and  $g$  are used to denote the yield function and the plastic potential function respectively. Considering planar deformations only, with the  $x$ - $y$ -coordinates in the plane and the  $z$ -axis normal to the plane, we define

$$f = \tau - (c \cot\phi + \sigma) \sin\phi$$

$$g = \tau - \sigma \sin\psi$$

where

$$\tau^2 = \frac{1}{4}(\sigma_{xx} - \sigma_{yy})^2 + \sigma_{xy}^2, \quad \sigma = -\frac{1}{2}(\sigma_{xx} + \sigma_{yy})$$

For the sake of convenience, it is assumed that the out-of-plane stress  $\sigma_{zz}$  is in

between the principal stresses  $\sigma - \tau$  and  $\sigma + \tau$ .

### 3 IMPLICIT INTEGRATION

Finite element analyses involve finite increments of stress and strain rather than rates as considered above. Several integration rules can be applied to pass from rates to finite increments. One of the most popular ones, as first proposed by Vermeer (1979) and most recently by Borja & Lee (1990), is the implicit integration scheme. Here the direction of flow is considered to be fully determined by the state of stress at the end of a load increment. Denoting finite increments by the symbol  $\Delta$  and  $t$  for time, we get

$$\Delta \underline{\varepsilon}^p = \int \lambda \frac{\partial g}{\partial \underline{\sigma}} dt = \frac{\partial g}{\partial \underline{\sigma}} \int \lambda dt = \frac{f^e}{d} \frac{\partial g}{\partial \underline{\sigma}}$$

where

$$f^e = f(\underline{\sigma}^* + \underline{D} \Delta \underline{\varepsilon})$$

where  $\underline{\sigma}^*$  is the stress at the beginning of the load increment. Details are given by Vermeer & Van Langen (1989). Having defined the increment of plastic strain as a function of the total strain increment, it is easy to compute the stress increment from:

$$\Delta \underline{\sigma} = \underline{D} (\Delta \underline{\varepsilon} - \Delta \underline{\varepsilon}^p)$$

### 4 FINITE ELEMENT FORMULATION

We consider a soil body subject to constant gravity and constant external loads. These are equilibrated by a known stress field  $\underline{\sigma}^0$ . Hence, the finite element equilibrium conditions give:

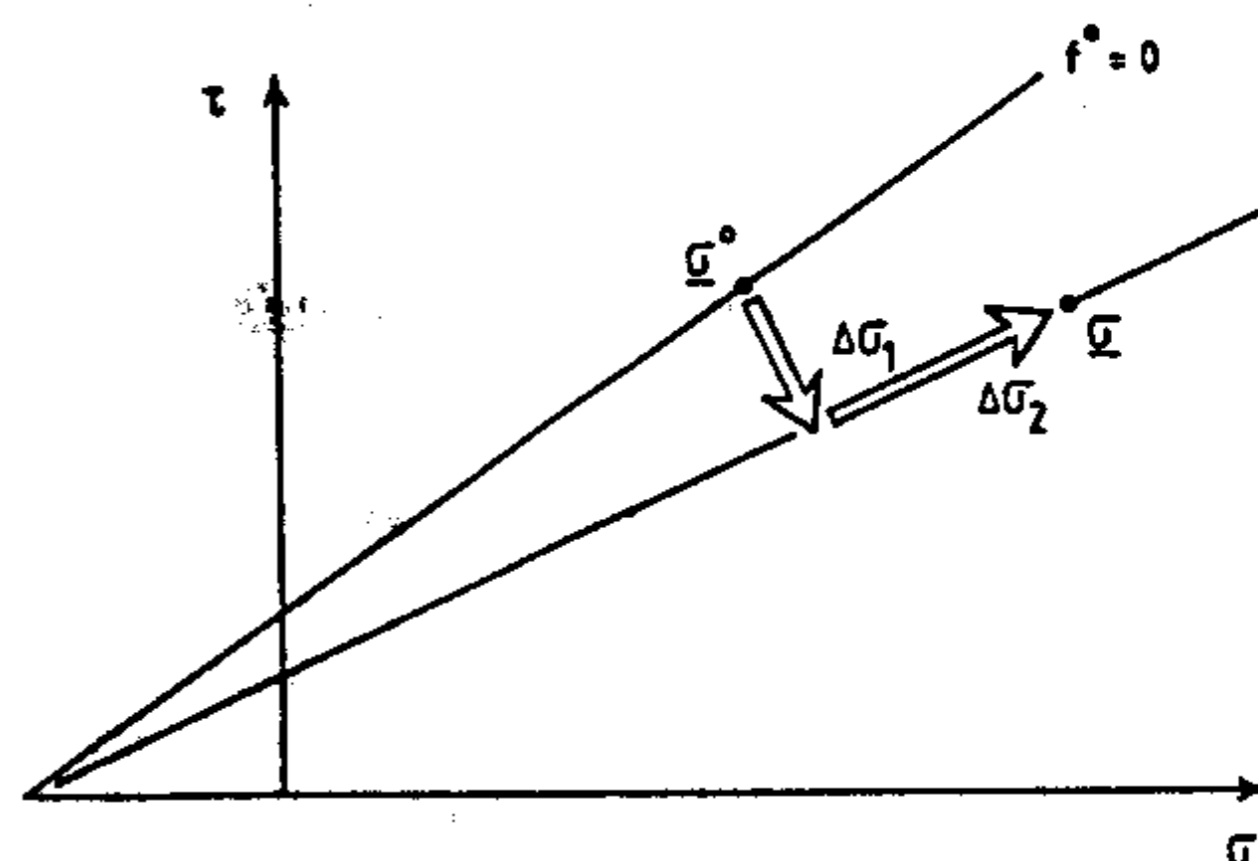


Fig. 1. Increments for strength-reduction

$$\int_V \underline{B} \underline{\sigma}^0 dV = \underline{Q}^0, \quad \int_V \underline{B} \underline{\sigma} dV = \underline{Q}^0$$

where both body forces and external loads are assembled in the vector  $\underline{Q}$  for nodal forces. We now apply a decrement of the strength parameters, so that the yield locus rotates as indicated in Fig.1. Obviously, the new stresses  $\underline{\sigma}$  (without superscript) will be at the new yield locus, at least for material points with stresses  $\underline{\sigma}^0$  above this line. Let us now define

$$\underline{\sigma} = \underline{\sigma}^0 + \Delta \underline{\sigma}^1 + \Delta \underline{\sigma}^2, \quad \Delta \underline{\sigma}^2 \equiv \underline{D} (\Delta \underline{\varepsilon} - \Delta \underline{\varepsilon}^p)$$

where  $\underline{\sigma}^0 + \Delta \underline{\sigma}^1$  is the closest point projection of  $\underline{\sigma}^0$  on the new yield surface. Substitution in the equilibrium equation, and using  $\underline{\varepsilon} = \underline{B} \underline{u}$ , gives:

$$\underline{K} \Delta \underline{u} = \underline{R}^0 + \Delta \underline{R} + \underline{P}$$

where  $\underline{K}$  is the elastic stiffness matrix and

$$\underline{R}^0 = \underline{Q}^0 - \int_V \underline{B} \underline{\sigma}^0 dV = 0$$

$$\Delta \underline{R} = - \int_V \underline{B} \Delta \underline{\sigma}^1 dV, \quad \underline{P} = \int_V \underline{B} \underline{D} \Delta \underline{\varepsilon}^p dV$$

In practice  $\underline{R}^0$  will not entirely vanish because of numerical procedures; separation from  $\Delta \underline{R}$  is important for later formulation of an arc-length control procedure.  $\underline{P}$  is a pseudo load vector, as used within an initial-stress iteration-procedure of the type:

$$\underline{K} \Delta \underline{u}^{k+1} = \underline{R}^0 + \Delta \underline{R} + \underline{P}^k,$$

$$\text{for } k=1, 2, \dots, k_{\max} \quad \text{and } \underline{p}^1 = 0$$

We adopt an initial stress procedure since a tangent stiffness procedures tend to break down in the fully plastic range due to ill-conditioning.

## 5 INDIRECT DISPLACEMENT CONTROL

The above strength reduction procedure is far from being robust. Indeed, by a step wise decrease of strength it is possible to approach collapse, but for some step the strength will be reduced too much, i.e. beyond critical, so that the iterative procedure will not converge to an equilibrium state of stress. In this manner the precise critical values of  $\tan\phi$  and  $c$  are never obtained. Therefore indirect displacement control is needed, or in other words arc-length control (Rheinboldt & Riks, 1986). Within this approach we add a load multiplier  $\beta$ , and obviously an extra scalar equation for solving  $\beta$ . The system of iteration equations is

$$\underline{K} \Delta \underline{u}^k = \underline{R}^o + \beta^k \Delta \underline{R} + \underline{P}^{k-1}$$

$$\Delta \underline{u}_o^T \Delta \underline{u}^k = \Delta \underline{u}_o^T \Delta \underline{u}_o$$

The subscript  $o$  is used to denote results from the previous load step. Instead of the node displacement vector of the previous step, other vectors can be used to obtain a linear scalar equation for  $\beta$ . Similarly non-linear equations are feasible.

## 6 STABILITY OF RIVER EMBANKMENT

The first practical application involves a river embankment in the tidal zone. The slope has a horizontal length of 10 m and a height of 5 m, so the ratio is 1:2. The most dangerous situation for a embankment is at the onset of low tide. The waterlevel outside the embankment is already low, but the phreatic level in the embankment is still high. The situation, which is modelled is shown in Fig. 2. We consider a homogeneous soil with material properties as indicated in Table 1.

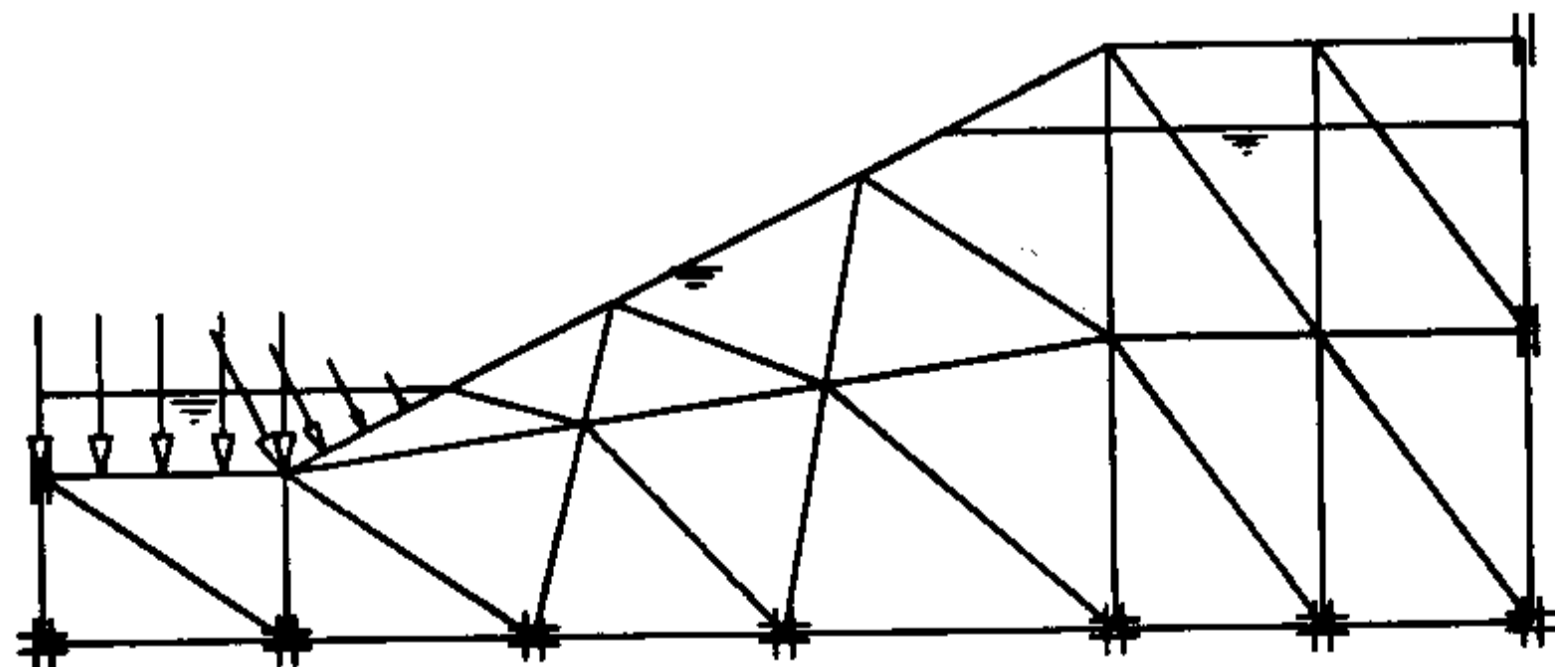


Fig. 2. River embankment at low tide

Table 1. Soil properties for embankment

Parameter	Symbol	Value
Dry weight	$\gamma_d$	16.0 kN/m <sup>3</sup>
Wet weight	$\gamma_w$	20.0 kN/m <sup>3</sup>
Friction angle	$\phi$	30.0 °
Cohesion	$c$	5.0 kPa
Shear modulus	$G$	1000 kPa
Poisson's ratio	$\nu$	0.3

In the first part of the analysis we apply the soil weight to introduce the stresses for the situation of Fig. 2. Some of the Gaussian integration points appear to be in plastic state already, but the embankment has not collapsed at all.

In the second part of the calculation we want to analyze the safety of the construction. Therefore we apply ten steps in which the strength parameters of the soil are stepwise reduced, according to the theory as described before. When further reduction of the strength parameters is not possible anymore, the construction has collapsed and the safety factor is obtained. Fig. 3 shows the computed strength-displacement curve for the crest point of the embankment. The value of the safety factor for this particular problem appears to be 1.20.

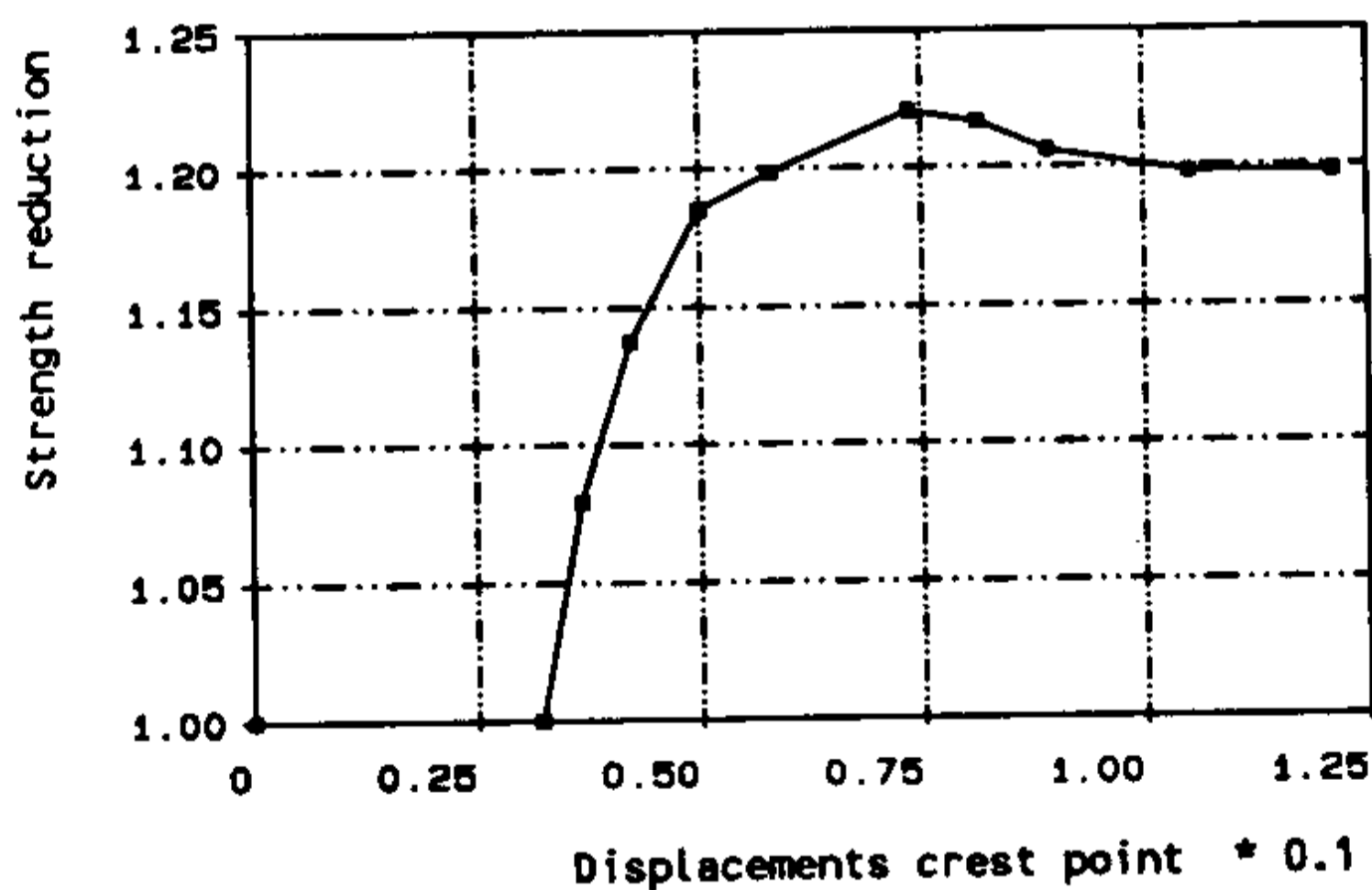


Fig. 3. Strength-displacement curve

Fig. 4 shows a plot of the incremental displacement contours at failure. In this plot one can recognize a nice slip circle. In the same plot the critical slip circle according to Bishop's method is plotted. Note that we performed a drained analysis both for the finite element approach and the Bishop approach.



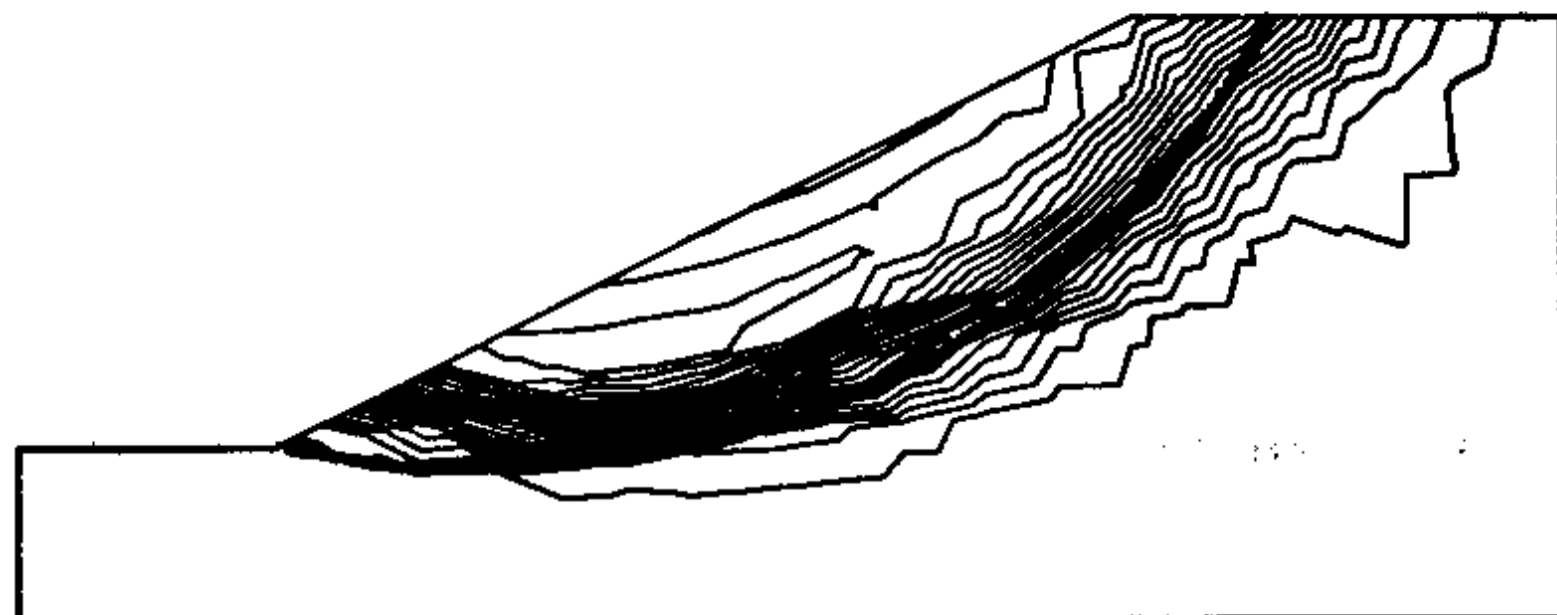


Fig. 4. Incremental displacement contours

Whilst the positions of the slip circles are somewhat different, the safety factor in both Bishop's method and the finite element calculation is exactly the same. This gives some validation of the proposed method. It might now be concluded that the calculation of the safety factor in finite element codes has no advantage in comparison with conventional methods. In this simple case this happens to be true, but the finite element method will also detect non-circular slip surfaces and associated factors of safety. The next examples will show that in general the finite element calculation of the safety factor gives information that cannot be obtained with conventional methods.

### 7 STABILITY OF A SHEET-PILE WALL

For the second application we consider the safety of an excavated building trench. The excavation depth is 21 m. The soil is supported by 32 m long sheet-pile walls, which are strutted at the top. During the excavation of the soil the water table remains at a constant level, both inside and outside the trench. Fig. 5 shows the finite element mesh of the trench. Special line elements, as available in the PLAXIS computer code, are used to model the

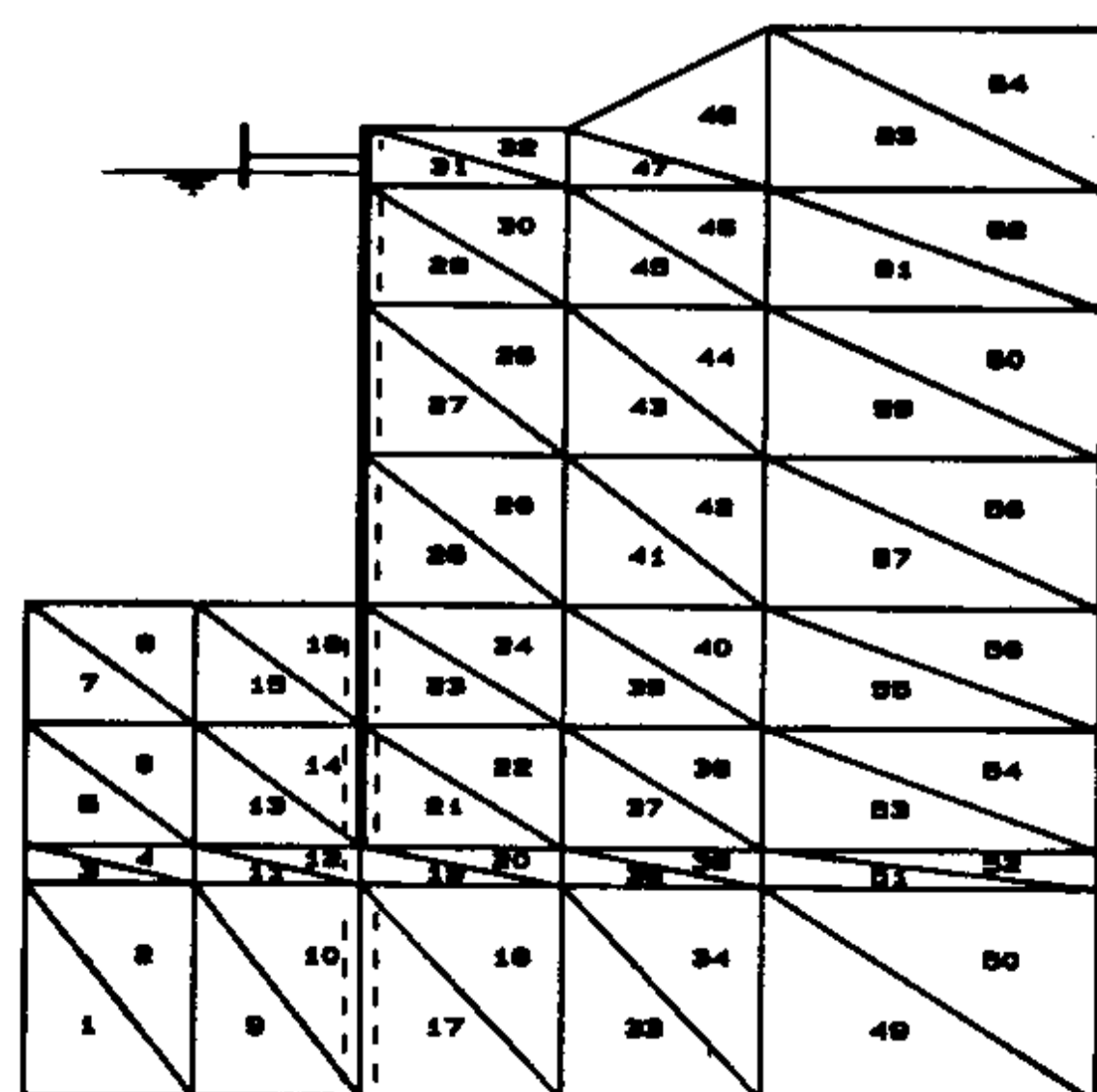


Fig. 5. Mesh for building trench

sheet-pile wall (Bakker & Brinkgreve, 1990). The dashed lines in Fig. 5 indicate joint elements to model reduced friction at the soil-structure interface (Van Langen & Vermeer, 1990).

The soil is mostly fine silty sand with some clay layers. In fact it consists of 7 layers, which correspond to the element layers in the mesh of Fig 5. The material properties for each layer are given in Table 2. Layer No. 1 is the top layer and layer No. 7 is the lowest layer. For the sheet-pile wall and the strut we used:

$$EI_{\text{sheet}} = 1.65 \times 10^6 \text{ kNm}^2/\text{m}$$

$$EA_{\text{sheet}} = 8.04 \times 10^6 \text{ kN/m}$$

$$EA_{\text{strut}} / L = 1.28 \times 10^5 \text{ kN/m}^2$$

Table 2. Properties for different layers

Layer	$\gamma_d$ kN/m <sup>3</sup>	$\gamma_w$ kN/m <sup>3</sup>	$\phi$ °	c kPa	G kPa
1	18.0	20.0	30.0	1.0	6000
2	16.0	16.0	25.0	8.0	3000
3		20.0	32.5	0.0	6000
4		20.0	32.5	0.0	8000
5		20.0	32.5	0.0	25000
6		18.0	27.5	4.0	9500
7		20.0	32.5	0.0	15000

The soil which is to be excavated, is modelled by external loads: horizontal tractions simulating the horizontal stresses at the sheet-pile wall and vertical tractions simulating the vertical stresses at the bottom of the trench.

The calculation consists of 3 parts. The first part is the introduction of the initial stresses including the tractions on the sheet-pile wall and the trench bottom.

The second part is the excavation of the trench by stepwise removing the external loads. Bending moments and strut forces occur in this part of the calculation. There is a stress reduction directly behind the sheet-pile wall mainly due to arching (Bakker & Brinkgreve, 1990). The extreme horizontal displacement of the sheet-pile wall is about 100 mm. The extreme bending moment amounts 2100 kNm/m and the strut force is 500 kN/m. Fig. 6 shows a plot of the incremental displacement contours at service state. A concentration of contour lines can be regarded as the initialisation of a shear band.

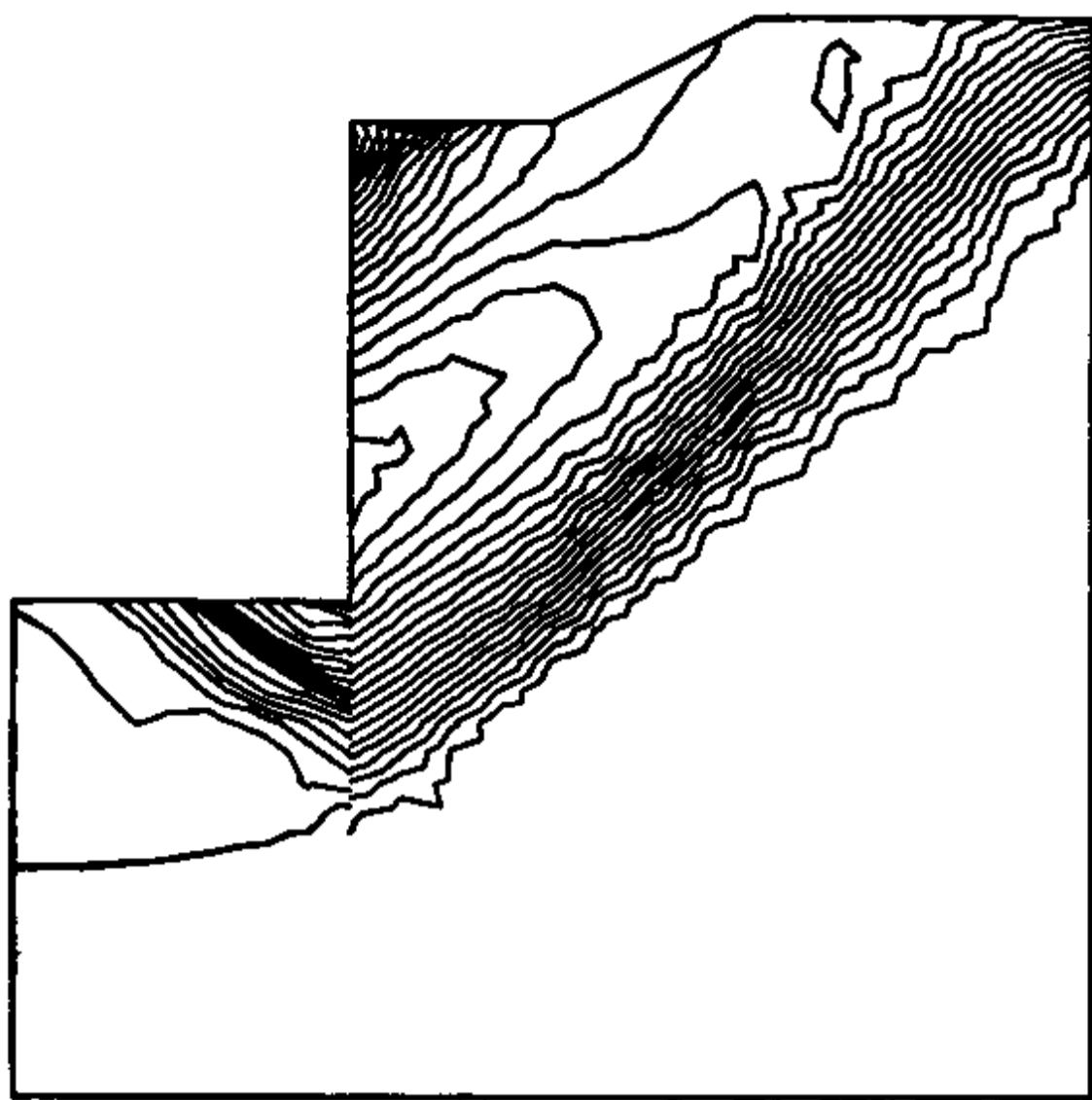


Fig. 6. Velocity contours at service state

The third part of the calculation is the determination of the safety factor by means of strength reduction. According to this calculation, the safety factor appears to be 1.65. The development of the bending moments during this calculation part is remarkable, as the clasping moment disappears and the sheet-pile wall tends to behave like a beam on two supports (Bakker & Brinkgreve, 1991).

Fig. 7 shows a plot of the incremental displacement contours at collapse. This plot visualizes a slip line going from the right-hand top, underneath the sheet-pile wall to the middle of the trench bottom. In comparison with the service state the slip line has moved drastically downwards.

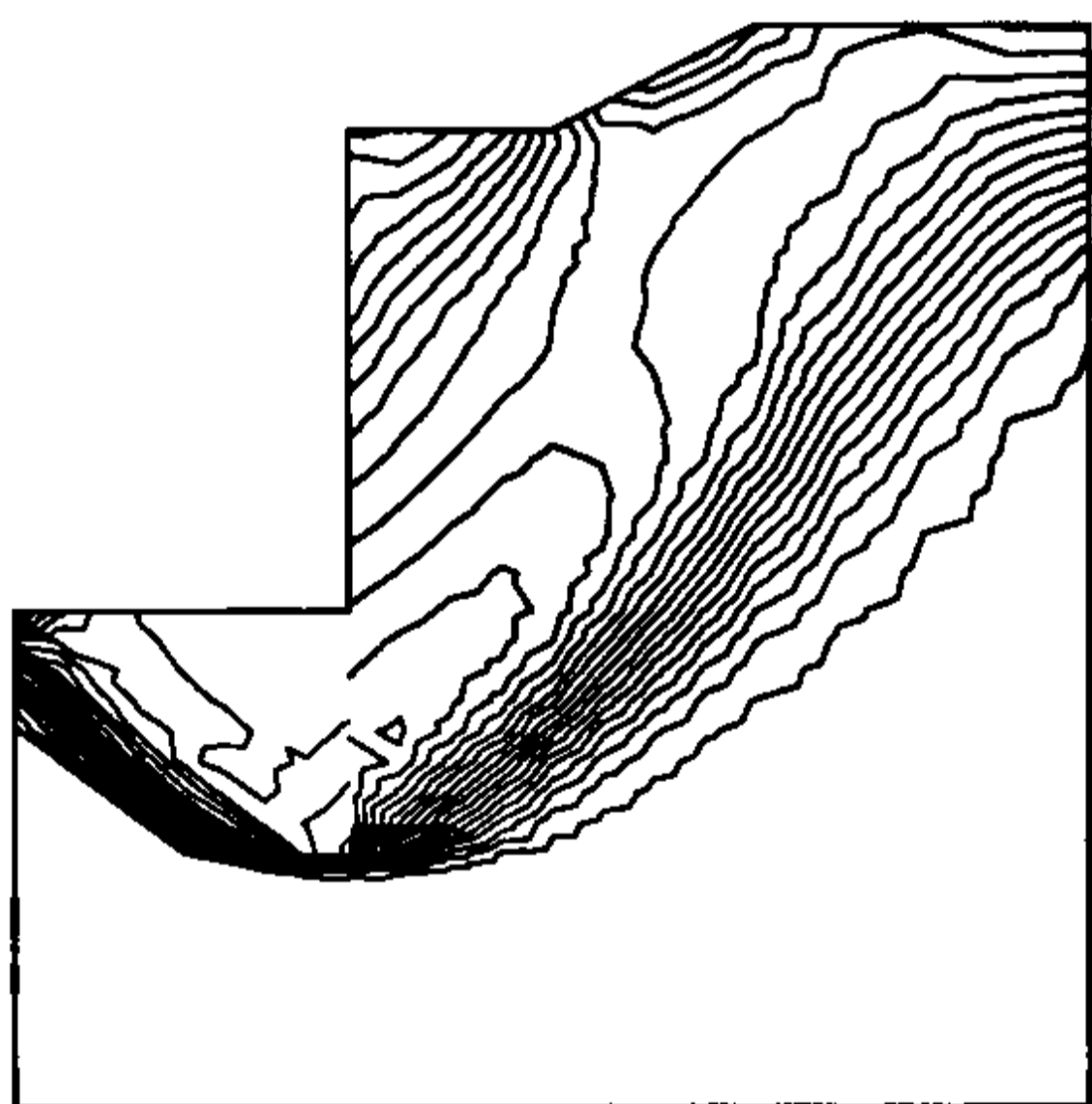


Fig. 7. Velocity contours at failure

### 8 EMBANKMENT STABILITY WITH GEOTEXTILE

The third application involves the safety of a road embankment. The embankment has a height of 6 m. In this example two situations are modelled: Firstly an

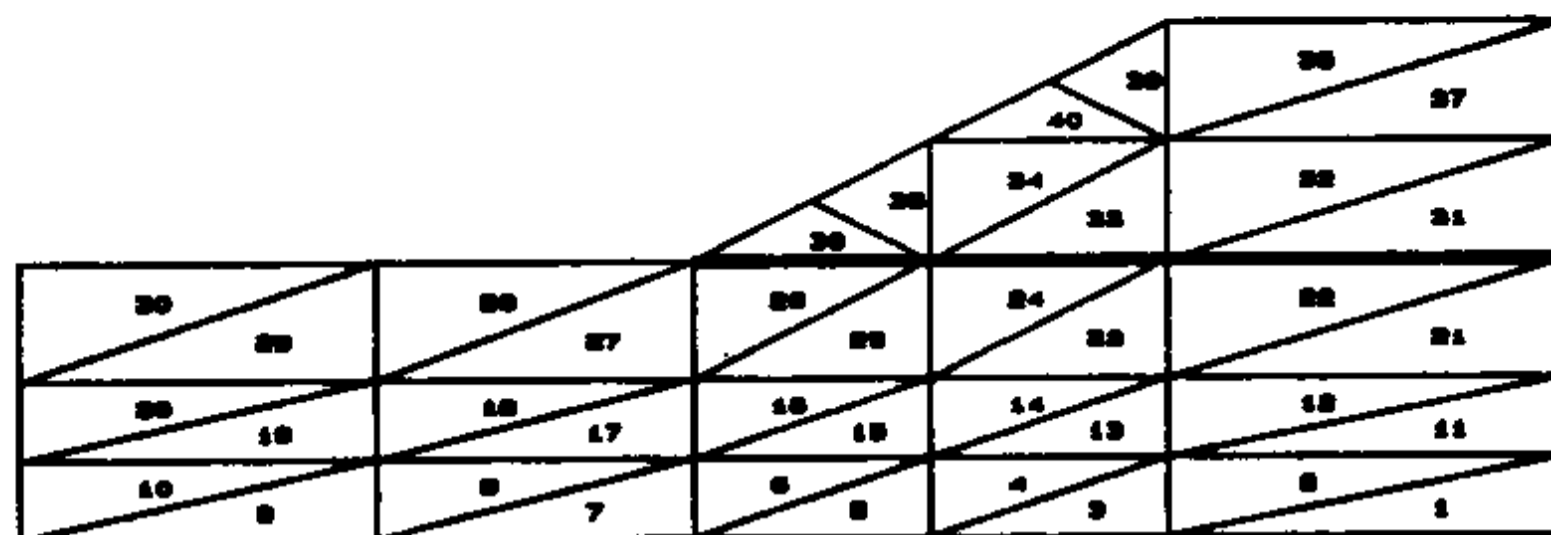


Fig. 8. Geometry of road embankment

ordinary case where the embankment is build just on top of the underground and secondly with the addition of a geotextile underneath the embankment. The situation is shown in Fig. 8.

The soil consists of four layers. The top layer is the embankment fill material. The underground consists of two clay layers with a layer of soft peat in between. The soil properties are given in Table 3.

Table 3. Properties for different layers

Layer	$\gamma_d$ kN/m <sup>3</sup>	$\gamma_w$ kN/m <sup>3</sup>	$\phi$ °	c kPa	G kPa	$K_0$
1	18.0	18.0	25.0	2.0	500	
2	18.0	18.0	24.0	5.0	500	0.60
3	11.0	11.0	15.0	2.0	250	0.75
4	18.0	18.0	24.0	5.0	500	0.60

Again the analysis consists of 3 parts. The first part is the introduction of the initial stresses in the underground. As the underground is horizontal the initial stresses can directly be derived from the soil weight and the  $K_0$ -values given in Table 3.

The second part of the calculation is the building of the embankment. At the end of this part the top of the embankment shows a settlement of 68 cm (without geotextile) and 66 cm (with geotextile). The geotextile seems to have little influence so far.

The third part is the calculation of the safety factor. Now the differences between both situations are much more apparent. In the case without geotextile the safety factor is less than 1.20 while in the case with geotextile the safety factor is over 1.30. In practice, only the latter situation is acceptable.

The difference in safety factor between the two situations is due to the fact that the failure mechanisms are completely different. Fig. 9 and 10 show plots of the incremental displacement contours of both situations. In the first case the failure mechanism is a slip circle going through the soft peat layer. In the second case the same mechanism is initially developed, but this is resisted by the geotextile. Collapse finally occurs in the embankment itself.

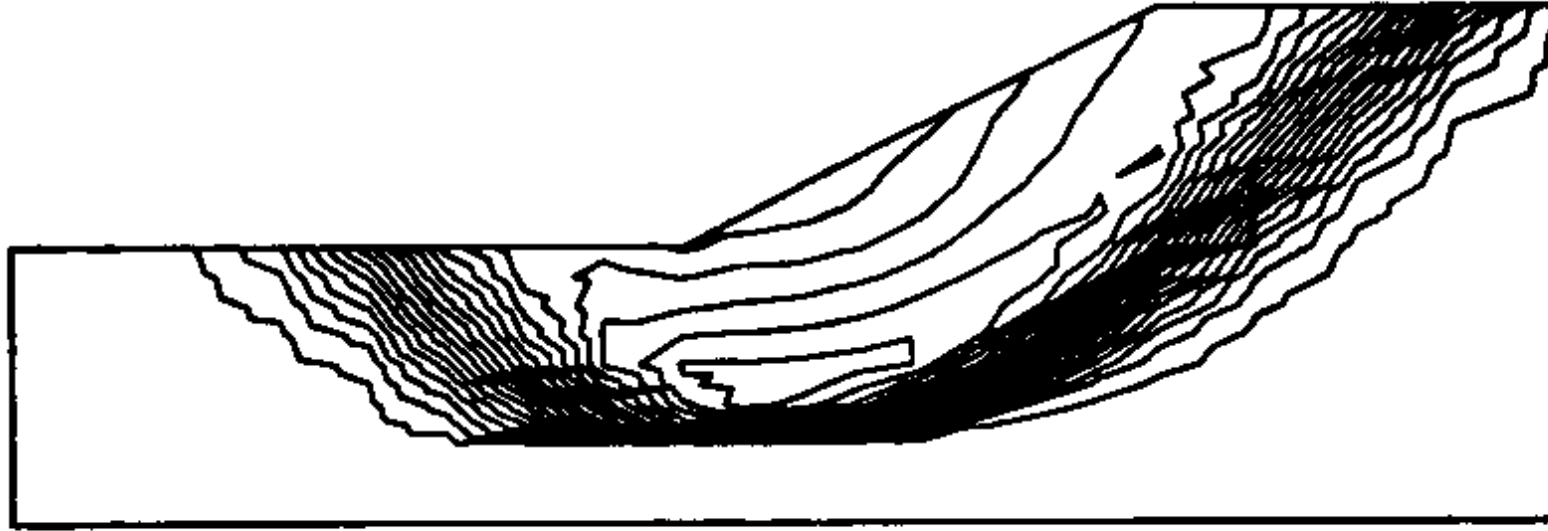


Fig.9. Velocity contours without geotextile

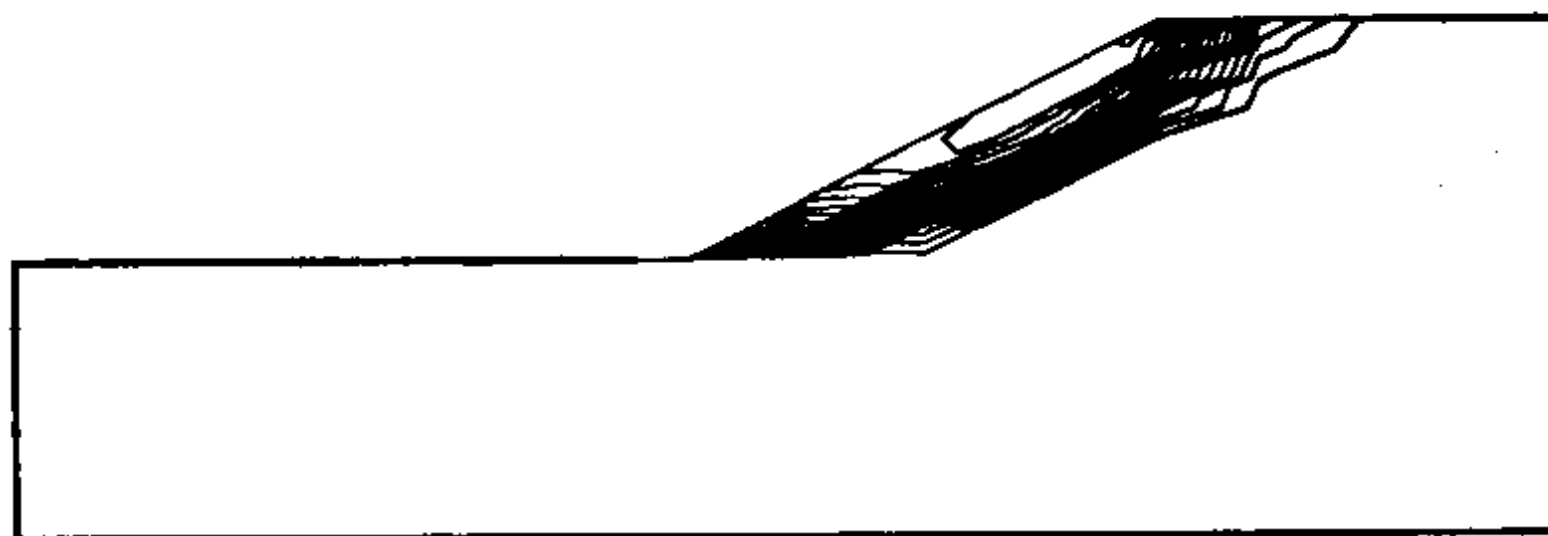


Fig. 10. Velocity contours with geotextile

Considering the results it would seem that geotextiles have an effect on stability rather than on deformation, but this will depend on the relative stiffness of the geotextile. For the present application we used  $EA = 2500 \text{ kN/m}$ .

## 9 CONCLUDING REMARKS

The proposed finite element method for the determination of the safety factor by means of strength reduction gives results which agree well with those from Bishop's slip-circle method, at least for simple situations when a circular surface occurs.

The advantage of the method in comparison with conventional methods is that it even deals with the most complex kind of geotechnical constructions. Besides one does not have to predefine the failure mechanism.

Arc-length control makes the procedure robust since failure need not be associated with a non-converging iterative procedure. Moreover the arc-length procedure gives the full strength-displacement curves beyond possible peaks.

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