

**STABILITY OF SLOPES IN A TWO-LAYER  
SYSTEM OF ANISOTROPIC SOILS**

By

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## NOMENCLATURE

$\beta$	slope angle
$H_1$	thickness of top layer
$H_2$	thickness of bottom layer
$\gamma_1$	unit weight of soil in top layer
$\gamma_2$	unit weight of soil in bottom layer
$C_1$ and $C_2$	cohesion along vertical and horizontal planes, respectively, in top layer
$C'_1$ and $C'_2$	cohesion along vertical and horizontal planes, respectively, in bottom layer
$\alpha$ and $\lambda_1$	geometrical parameters characterizing the slip surface in top layer
$\alpha'$ and $\lambda_2$	geometrical parameters characterizing the slip surface in bottom layer
$\theta$	geometrical parameter
$f$	the angle between the failure plane and the plane normal to the direction of the major principal stress
$i$	angle of major principal stress from vertical, measured clockwise
$C_u$	undrained strength
$p$	relative strength index
$n$	thickness ratio
$K$ and $K'$	coefficients of anisotropy
$m$	$\gamma_2/\gamma_1$
$n$	$H_2/H_1$
$C'_1$	$(p + 1) C_1$

$$C_2'$$

$$(p + 1) C_2$$

$$\frac{C_2}{C_1}$$

K

$$\frac{C_2'}{C_1'}$$

K'

## CHAPTER I

### INTRODUCTION

The stability of earth slopes is a matter of considerable importance in the construction of highways, railways, and earth dams, as well as in connection with landslides. During the last four decades, numerous efforts have been made to deal with the problem of stability of slopes with the aim of computing the factor of safety with respect to the sliding of slopes of cuts and embankments. The properties of the material (soil) in which sliding occurs deviate quite considerably from those of elastic solids. Hence, the laws of strength of materials and theory of elasticity do not hold true precisely for soils.

The absence of any mathematical theory which could be used to deal with real soil materials, coupled with the necessity to solve problems in soils engineering, require that several assumptions be made idealizing this material, the most important among them being homogeneity and isotropy. Since most of the methods for stability analysis are based on these two basic assumptions, they give only a rough estimate of the factor of safety.

In nature, there are two deviations from the ideal homogeneous material. The first is the case in which the subsoil consists of layers of distinctly different soils. The second is the case of a soil deposit which lacks any distinct stratification but whose properties vary from one point to another over a wide range. This type of non-homogeneity

makes it difficult to arrive at representative soil properties to be used in calculations.

Again, in nature, most soils are anisotropic because of the mode of deposition, the stress metamorphosis after deposition, or both. There is considerable evidence in literature showing that the stability of earth slopes is influenced by non-homogeneity and anisotropy in strength (Gibson and Morgenstern, 1962; Lo, 1965; Livneh, 1967; and others).

In this thesis, an analytical method is presented for analyzing slope stability problems in a two-layered system of non-homogeneous and anisotropic soils. Chapter II deals with a brief review of published literature regarding the methods available for slope stability analysis. Since the published material in this area of soil mechanics is quite extensive, only that material which is directly related to the present work is reviewed in detail. The reader is referred to the references cited in the bibliography for additional information.

Chapter III deals with the analytical method for solving slope stability problems in a two-layered system of non-homogeneous and anisotropic soils. The working formulae for stability number and factor of safety are derived in detail. In Chapter IV, the results obtained in this analysis are discussed, and the conclusions drawn from this study are listed. Some aspects related to this area which merit further research are suggested. Charts, slope angle versus stability number, thickness ratio versus stability number, and relative strength index versus stability number are presented (Figures 7 through 30).

Two hypothetical problems dealing with slope stability in a layered system are worked out in Appendix A. These demonstrate the

usefulness of the charts in analyzing stability problems in layered soils. The computer program used in obtaining numerical results in this study is given in Appendix B.

## CHAPTER II

### REVIEW OF LITERATURE

The construction of earthen embankments is as old as civilization itself. Many ancient man-made embankments exist in China, India, and the Middle East. Rao (1961) describes 18 existing earth dams constructed in southern India between 1000 A.D. and 1800 A.D.

The dams described above, as well as those built in other countries during that or earlier periods, were designed by men who had only an empirical knowledge of mechanics and material properties. The successful functioning of these structures clearly indicates that embankment design could be carried out based on some experience and intuition. The main drawback of this approach was that it did not permit a quantitative assessment of the safety of structure. Nor could it handle unusual conditions encountered in embankment design. Real progress in this direction was to wait until tools of mathematics and mechanics and scientific knowledge of soil behavior were available.

Probably the earliest contribution dealing with soil behavior was that of Coulomb (1773), who has given the expression for the critical height of vertical clay slope as

$$H_c = \frac{4c}{\gamma} \frac{\cos \varphi}{1 - \sin \varphi}$$

where  $c$  = cohesion,

$\gamma$  = unit weight, and

$\varphi$  = internal friction of clay.

In 1820, Francais extended Coulomb's analysis to the case of a clay bank sloping at an angle of  $\theta$  to the horizontal and, on the assumption that the slip surface was a plane passing through the toe of the slope, deduced the following expression for the height of limiting stability:

$$H = \frac{4c}{\gamma} \left\{ \frac{\cos \varphi \sin \theta}{1 - \cos(\theta - \varphi)} \right\} .$$

This equation is generally, but incorrectly, attributed to Culmann, who published it forty-six years later in his book "Die Graphische Statik" (1866).

The earliest pioneering work in stability of clay slopes was done by Collin (1846). He published a memoir which contained careful field observations on approximately fifteen slips in cuttings, embankments, and earth dams, a description of shear box tests on clay samples, and an approximate analysis of stability which was the primitive forerunner of the present-day  $\varphi = 0$  analysis. Yet after its publication, the book remained almost unknown for seventy years until the study of the stability of clay slopes was again placed on a sound basis by the work of such men as Resal (1910), Bell (1915), Petterson and Hultin (1916), Fellénus (1916-22), Frontard (1922), and others.

The curious neglect of such an important work and, indeed, the almost stationary position of soil mechanics during this interim period is due to two main causes. Firstly, the mechanical properties of soils and especially clays are considered to be more complex than those of construction materials such as masonry, concrete, and steel. Consequently, research was more profitably directed toward structures and hydraulics. Secondly, the period 1840-1910 was dominated in the field of soil mechanics by the conventionalized theories of bearing capacity,

earth pressure and slope stability, due principally to Poncelet (1840) and Rankine (1857 and 1862), which although of some value for sands are in most cases misleading for clays. Cohesion was ignored, the curved surface of rupture was forgotten, and the false "angle of repose" reigned supreme.

Modern developments leading to the present state of art began during the second decade of the century. The revival of interest appears to have stemmed from several serious landslides in Sweden and from massive landslides that took place during the construction of the Panama Canal. There are probably two other causes for the rapid progress that followed this renewal of interest. First, at about this time, improved and more widespread understanding of the soil properties was developing. Second, there appeared on the scene the great guiding genius of Terzaghi, who was to weld the principles of mechanics and the properties of soil into a coherent whole that would lay the foundation for a new science.

The slip of Stigberg quay into the harbor at Gothenberg, Sweden, in 1916 touched off an investigation that had profound consequences in slope stability analysis. Petterson and Hultin (1916) investigated the slide. In addition to analyzing this slide, they also studied earlier case histories, assuming circular rupture surfaces. According to the concepts of the time, clay was treated purely as a frictional material.

Professor Moller studied the problem concerning friction angles and safety factors more carefully than had been done earlier and published a book Erddruch-Tabellen. He brought out a second edition of the book in 1922 with the subtitle "Augmented with New Earth Pressure Investigations." This is the first publication on circular sliding



surfaces to be internationally known. Professor Fellenius continued working on circular sliding surfaces and published several papers during 1918-26. As a partly new method of analysis instead of the friction method formerly solely used, he took up the question of bringing the cohesive strength of soil into the calculations. He referred to investigations and experiments made by the Geotechnical Commission of the Swedish State Roads. The report, published in 1922, contained many examples of circular slides in Sweden. By direct mathematical treatment combined with graphical methods, he arrived at a series of tables and diagrams for predicting the critical slip surfaces. The circular arc analysis for slope stability problems has, in time, come to be known as the "Swedish Method."

Professor Terzaghi (1936) compared circular arc analysis with logarithmic spirals and cycloids and decided that the circular arc method is the most convenient, as well as sufficiently accurate, for many engineering problems. Since the development of the Swedish Method, many new procedures have been advanced to solve the slope stability problem. These are tabulated, showing the assumptions involved and the originators, in Table I.

Most of the methods available for performing slope stability analysis may be categorized as limit equilibrium methods. The basic assumption of the limit equilibrium approach is that Coulomb's failure criterion is satisfied along the failure surface. A weakness of the limit equilibrium method is that it neglects the soil's stress-strain relationship. In an attempt to take into account the stress-strain relationship in analyzing the stability of slopes, it has been suggested that the theory of plasticity be applied to the problem (Drucker and

TABLE I

## CLASSIFICATION OF SLOPE STABILITY ANALYSIS BY LIMIT EQUILIBRIUM METHOD

(After Fank and Hirst, 1970)

Type of Failure Plane	Name of Method	Type of Solution	Basic Assumptions	References
Straight line	Culmann Method	Analytical	Failure occurs on a plane through the toe of the slope.	Culmann (1866)
	Method of Infinite Slope	Analytical	The slope is constant with unlimited extent.	Resal (1910)
			A vertical column is typical of the entire mass. No cohesion may be depended on within the depth to which tension occurs.	Frontard (1922)
Wedge	Semigraphical-analytical	Sliding block mechanism is assumed with lateral earth forces.	Culmann (1866), Terzaghi and Peck (1948)	
Circular Arc	Slices Method	Semigraphical-numerical	The lateral forces are equal on two sides of each slice.	Fellenius (1939)
	Bishop's Method	Analytical-Numerical	Oblique side forces on each slice are considered.	Bishop (1955)
	Simplified Bishop's	Analytical-Numerical	Vertical component of lateral earth forces are considered to be equal and opposite.	Bishop (1955), Little and Price (1958)

TABLE I (CONTINUED)

Type of Failure Plane	Name of Method	Type of Solution	Basic Assumptions	References
	$\varphi$ - Circle Method	Analytical-Graphical	Resultant acting on rupture arc is tangential to a concentric circle with radius = $R \sin \varphi_d$ .	Gilboy (1933), Taylor (1937), Casagrande (1934)
	Modified $\varphi$ - Circle	Analytical-Graphical	The resultant misses tangency to the $\varphi_d$ circle by a small amount. Radius of $\varphi_d$ circle = $K R \sin \varphi_d$ .	Taylor (1937)
Logarithmic Spiral	Log-spiral Method	Analytical	No assumptions required to make the problem statically determinate.	Rendulic (1935), Taylor (1937), Spencer (1969)
Irregular	Irregular	Analytical-Numerical	General slip surface. Forces between slices are considered.	Morgenstern and Price (1965)

Prager, 1952). Analytical procedures have been developed to solve homogeneous and isotropic earth slopes using plastic theory (Chen, Giger and Fang, 1969; Fang and Hirst, 1970; Chen and Giger, 1971).

In most of these methods, however, the basic principles have remained the same. The earlier methods were based on the assumptions that the soil is homogeneous and isotropic. Neither of these is true, as may be seen from the brief review of the published literature cited elsewhere in this chapter. In view of this, some recent procedures have utilized more realistic assumptions for the solution of slope stability problems (Morgenstern and Gibson, 1962; Lo, 1965; and others).

It is often necessary to determine the factor of safety of cuts in normally consolidated clays. These clays characteristically exhibit a linear increase of strength with depth, as shown in Figure 1. Close to the surface, the clay is usually overconsolidated due to desiccation; and the strength of this crust is higher than that of the material immediately below it. Morgenstern and Gibson (1962) have analyzed the slope stability problem in homogeneous soils with strength increasing linearly with depth, as shown in Figure 2.

The expression for the factor of safety established by Morgenstern and Gibson implies that the ground water table is at or above the ground surface. Their approach to this problem was extended by Hunter and Schuster (1969) to include an analytical solution for the common case of ground water table below the ground surface. This method permits a better estimate of the factor of safety where the shear strength  $C$  is greater than zero at the ground surface but still increases linearly with depth. Hunter and Schuster present graphically the results of

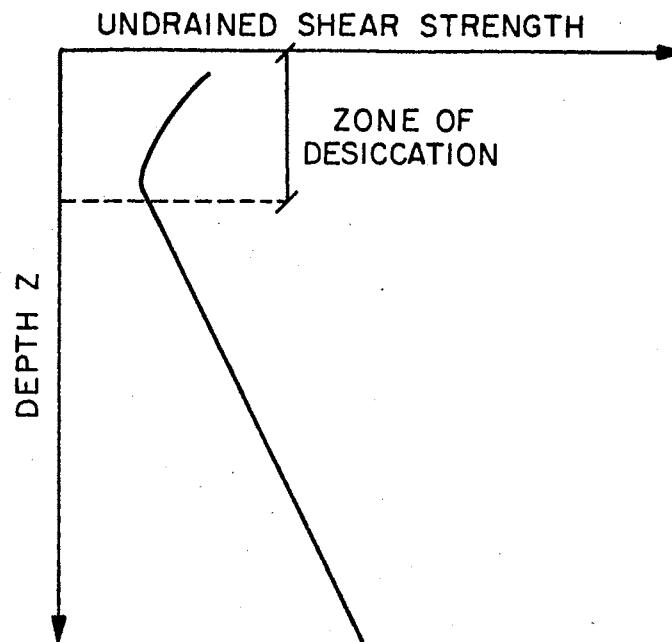


Figure 1. Typical Variation of Undrained Strength with Depth for a Normally Consolidated Clay

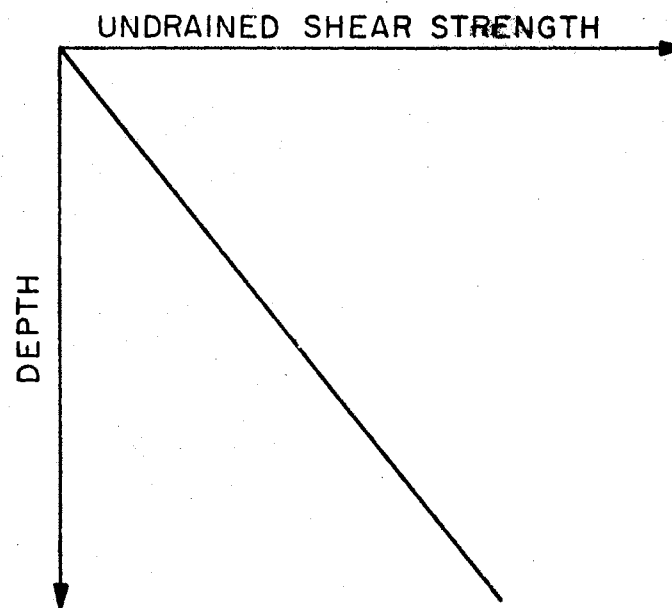


Figure 2. Idealized Strength Depth Relationship

computations of stability numbers for a range of slope inclinations and depths of water table.

In nature, the clay-size particles are nonsymmetrical in shape. They are usually longer in two directions than in the third. Due to this nonsymmetrical shape, the soil deformations and the soil-forming processes produce within the soil fabric an anisotropic structure. This anisotropic structure is reflected by directional variations in the physical properties of soil such as strength, compressibility, and permeability. There is ample evidence in literature confirming the occurrence of anisotropic soil fabric and anisotropic physical properties in natural and remolded cohesive soils. (Mitchell, 1956; Hvorslev, 1960; Duncan and Seed, 1966; and others).

Anisotropy of clays with respect to strength is probably related to the orientation of clay particles. The structure of clay was studied by many investigators (Lambe, 1953; Mitchell, 1956; Pacey, 1956; Martin, 1962; and others). These studies have shown that the clay particles tend to become oriented parallel to the major principal plane during anisotropic consolidation. Mitchell (1956) studied the structure of seven undisturbed marine clays and one lacustrine clay with the aid of a petrographic microscope. From these studies, he concluded that these clays had some degree of parallel orientation. With one exception, these clays were consolidated to nearly 3 kg/sq cm. Quigley and Thompson (1966), studying the relationship between soil fabric and the anisotropic consolidation characteristics of Leda Clay from Ottawa found that anisotropic consolidation of undisturbed samples causes reorientation of the clay platelets into a plane perpendicular to the direction of major principal consolidation pressure. Martin (1962)

studied the structure of Kaolinite Clay using X-ray diffraction technique. He compared the peak amplitudes of diffracted X-rays from 002 and 020 planes of the clay and concluded that the clay was approximately "ideally oriented" after one-dimensional consolidation to 197 kg/sq cm, and approximately "ideally random" after isotropic consolidation to 1 kg/sq cm.

It may be concluded from these studies that there is a tendency for the clay-size particles to become oriented parallel to the plane on which the major principal stress acts during consolidation. This parallel particle orientation might be expected to cause anisotropy in physical properties such as strength, compressibility, and permeability. Unconsolidated undrained triaxial tests on samples trimmed in different directions have shown that anisotropically consolidated clays are anisotropic with respect to undrained strength.

For a soil to be perfectly isotropic, the coefficient of earth pressure at rest,  $K_0$ , should be 1, as isotropic consolidation requires a hydrostatic state of stress to exist. As early as 1920, Terzaghi reported the value of  $K_0$  for a coarse sand to be 0.42. In 1925, he reported a value of 0.7 for a yellow residual clay and a blue marine clay; and experiments by Kjellman with triaxial equipment yielded values ranging from 0.5 to 1.5.  $K_0$  is supposed to be a function of stress history of soil. The assumptions that  $K_0 = 1 - \sin \phi'$  (Jaky) or  $K_0 = 0.95 - \sin \phi'$  (Brooker and Ireland, 1965) indicate that few soils have  $K_0 = 1$ .

In reviewing the published literature regarding anisotropic strength characteristics of clays, the definition for shear strength as shown in Figure 3 is followed.

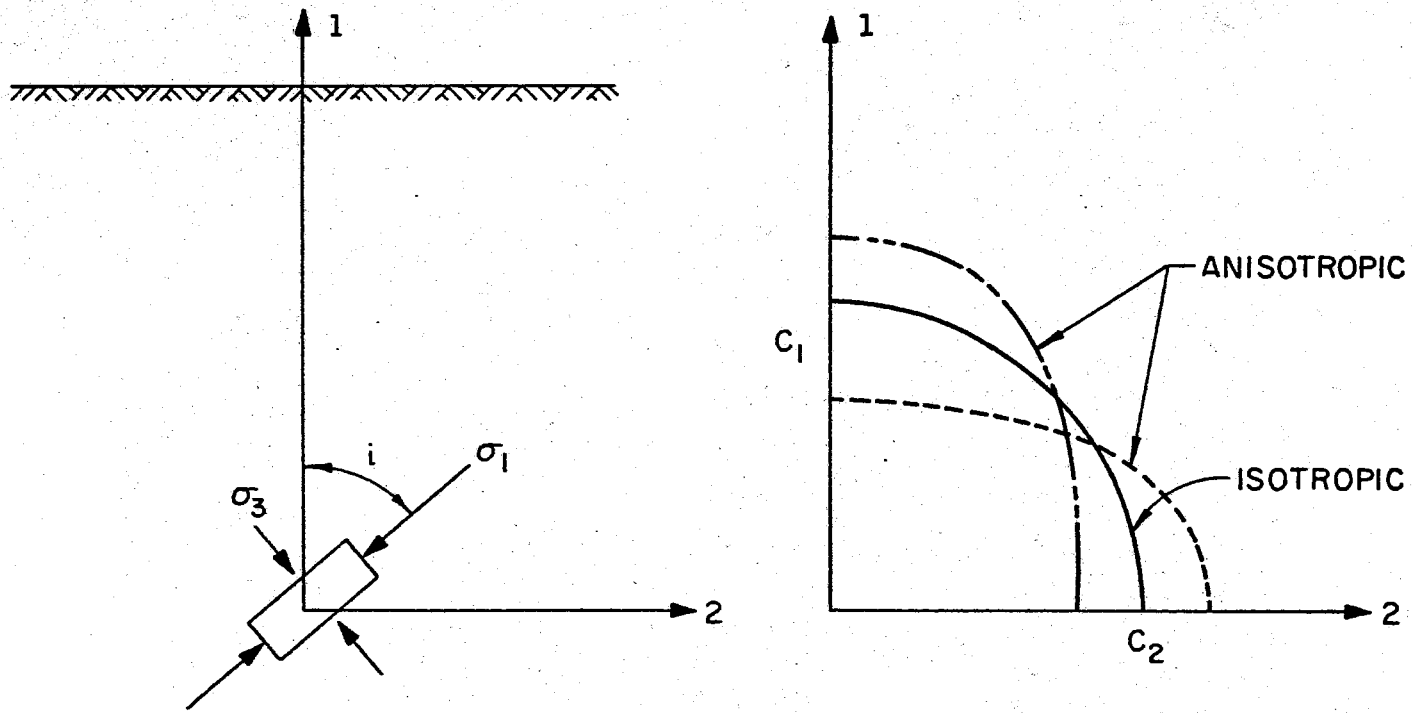


Figure 3. Definition of Shear Strength Variation with Direction (After Lo, 1965)



The physical vertical and horizontal directions, which usually coincide with lines perpendicular and parallel to the bedding planes of soil deposit are the principal directions. If the sample is tested with the major principal stress direction coinciding with the principal directions, the strengths thus determined ( $C_1$  and  $C_2$ ) are known as principal strengths. When the major principal stress makes an angle,  $i$ , with the vertical, the strength then determined is designated as  $C_i$ . For a soil with isotropic strength characteristics, the principal strengths  $C_1$  and  $C_2$  and  $C_i$  are equal. In other words, the curve traced by  $C_i$  in a vertical plane is a circle. However, for a soil having anisotropic strength characteristics, the principal strengths  $C_1$  and  $C_2$  are not equal and the curve traced by  $C_i$  is not a circle. The ratio of principal strengths  $C_2/C_1$  is termed as the degree of anisotropy. (Ranganatham and Mathai, 1967, denote it as the Coefficient of Orthotropy.) Depending on the stress history, clay particle orientation, etc., the ratio  $C_2/C_1$  is less than or greater than one. For convenience, the former is designated as M-anisotropy and the latter as C-anisotropy (Lo, 1965). Soil deposits with  $C_2/C_1$  equal to unity are rather rare in nature.

Lo (1965) performed unconfined compression tests on undisturbed samples of clay from Welland, Ontario, Canada. It was reported that the horizontal strengths were less than the vertical strengths, the ratio  $C_2/C_1$  varying between 0.64 and 0.8 (M-anisotropy).

Aas (1965), using vanes of different shapes, performed vane tests on Canadian clays. He reported the ratio of undrained shear strengths acting along horizontal and vertical failure surfaces to be 1.5 to 2.0 (C-anisotropy). Ward, Samuels, and Butler (1959) have reported from their tests on London clay that horizontal strengths were greater than

vertical strengths. The ratio  $C_2/C_1$  was established to be  $1.3 \pm 0.1$  (C-anisotropy). The higher strengths exhibited in the horizontal samples may be related to the fact that London clay is heavily overconsolidated and the horizontal stresses in the ground are considerably higher than the vertical stresses (Skempton, 1961). Skempton has shown that the ratio of horizontal-to-vertical stress in the overconsolidated London clay varies from 2.5 at the top to 1.5 at a depth of 100 feet below the surface. Some examples in which  $C_u$ , the undrained strength, depends on principal stress directions during shear are furnished in Table II.

From the brief review of published literature, it may be concluded that in nature the rule is anisotropy, isotropy being an exception. There is ample evidence in literature to show that stability of earth masses is affected by strength anisotropy (Lo, 1965; Ranganatham and Mathai, 1967; Livneh, 1967; and others). These investigators have used different strength variations, which are reviewed below, to account for strength anisotropy.

Lo (1965) developed a general method of stability analysis for anisotropic soils. Rigorous solutions were obtained for two cases: (a) the vertical strength was constant with depth, and (b) the vertical strength varied linearly with depth. He assumed the following strength variation, as suggested by Carillo and Casagrande (1942).

$$C_i = C_2 + (C_1 - C_2) \cos^2 i,$$

where

$C_i$  = shear strength when the major principal stress at failure is inclined at an angle,  $i$ , to the vertical

$C_1$  and  $C_2$  = principal strengths in directions of principal stresses.

Charts of slope angle versus stability number were presented.

TABLE II  
SOME EXAMPLES IN WHICH  $C_u$  DEPENDS ON PRINCIPAL  
STRESS DIRECTIONS DURING SHEAR (TESTS IN  
SITU OR ON UNDISTURBED SAMPLES)

1.  $C_u$  from field vane lower than  $C_u$  from piston samples or block samples. (Vold, 1956; Coates and McRostie, 1963).
2.  $C_u$  from field vane for vertical plane lower than for horizontal plane. (Aas, 1965).
3.  $C_u$  from block samples with axis horizontal lower than with axis vertical in lightly overconsolidated clay. (Lo, 1965).
4.  $C_u$  from block samples with axis horizontal higher than with axis vertical in heavily overconsolidated clay. (Ward, Marsland, and Samuels, 1965).

Livneh (1967) has also studied the effect of strength anisotropy on slope stability. In his analysis, the following variation for strength was assumed (Figure 4).

$$C_{\alpha} = C_1 \left[ 1 + (n - 1) \sin^{2K} (\alpha - \psi) \right]$$

where

$C_1$  = minimum cohesion,

$C_2$  = maximum cohesion,

$n$  = the anisotropy index defined as the ratio  $C_2/C_1$ ,

$\alpha$  = the angle between the horizontal axis and an arbitrary axis of reference,

$\psi$  = the angle between the horizontal axis and the minimum cohesion axis, and

$K$  = a positive integer permitting characterization of different cohesion patterns in terms of angle  $\alpha$ .

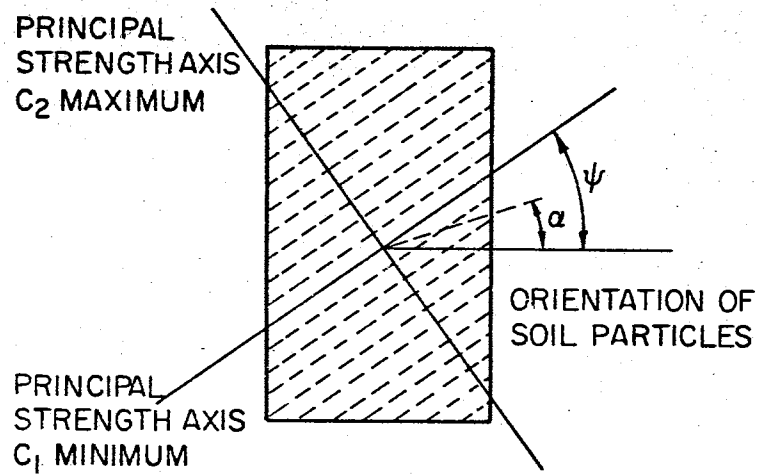
Livneh presented charts giving the slope angle versus stability factor ( $\gamma H/C$ ) for various values of  $\psi$ ,  $K$ , and  $n$ . He showed that neglecting the anisotropy factor would lead to results that are either conservative or in error on the unsafe side. Disregarding the anisotropy factor and assuming that the soil is isotropic may lead to a decrease in the computed factor of safety. For example, a 27 percent decrease is obtained for  $\beta = 10^\circ$ ,  $n = 2$ ,  $K = 1$ ,  $n_d = 1.5$ , and  $\psi = 60^\circ$ .

Ranganatham and Mathai (1967) have analyzed the effect of strength anisotropy on the stability of earth masses. They have assumed the following variation to account for anisotropy:

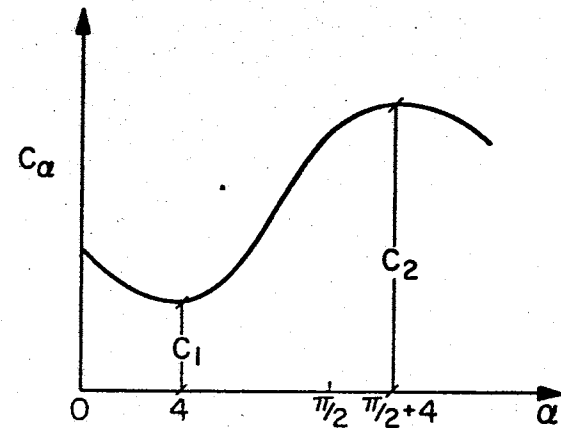
$$C = C_h (\cos^2 \theta + n \sin^2 \theta)$$

where

$C$  = cohesion along a plane inclined at an angle to  $\theta$  to horizontal,



(a)



(b)

Figure 4. The Definition of Anisotropic Cohesional Medium  
 (a) The Principal Strength Axes  
 (b) The Variation of Cohesion with  $\alpha$   
 (After Livneh, 1967)

$C_h$  = cohesive strength along the horizontal plane,

$C_v$  = cohesive strength along the vertical plane, and

$n = C_v/C_h$ , called the anisotropic strength ratio (also called the coefficient of orthotropy).

The proposed method followed the analysis of Janbu (1954) (based on dimensionless parameters for base failure in purely cohesive soil) except for strength anisotropy. Charts were presented giving the slope angle versus the stability number ( $C/\gamma H$ ) for various values of the coefficient of orthotropy and depth to hard stratum. It was noticed that for vertical cohesive strength ranging from half to double the horizontal, the stability number changed by about +30 percent to -40 percent of the isotropic case. From the numerical results, it was concluded that the influence of anisotropy on stability is much greater than is the depth to hard stratum.

In the above methods, two assumptions are made: (1) Soil mass is homogeneous, and (2) it is purely cohesive ( $\varphi = 0$  condition).

When estimating the stability of foundations and slopes, it is often assumed that the soil is homogeneous and isotropic; but it is known that the shear strength increases with depth beyond the zone of desiccation and is also dependent on the direction of the failure surface. While it is difficult to describe the exact functional relationship between the shear strength and depth and direction of failure surface, Rangana-  
tham, Sani, and Sreenivasulu (1969) investigated through carefully-planned experiments the probable variation of strength with depth and direction of failure plane and used these findings in evaluating slope stability. Experimental work was done to obtain the variation in shear strength (1) with direction of failure surface, keeping consolidation

pressure constant, and (2) with consolidation pressure, keeping the direction of failure surface constant.

This experimental study lends support to the hypothesis that the undrained strength of an element of soil along a plane other than the horizontal or vertical is equal to the vectorial sum of those acting on the projected areas of the element in the vertical and horizontal planes. Expressed mathematically, the strength  $C_\theta$  along a plane inclined at an angle  $\theta$  to the horizontal is given by

$$C_\theta = C_h \cos^2 \theta + C_v \sin^2 \theta$$

where  $C_h$  and  $C_v$  are strengths in the horizontal and vertical planes, respectively.

The shear strength at any depth  $Z$  in relation to the direction of failure surface was defined as follows:

$$\begin{aligned} C_{hZ} &= C_{ho} \left(1 + l_h \frac{Z}{H}\right) \\ C_{vZ} &= C_{vo} \left(1 + l_v \frac{Z}{H}\right) \\ &= n_{co} C_{ho} \left(1 + l_v \frac{Z}{H}\right) \end{aligned}$$

where

- $C_{hZ}$  = strength along the horizontal plane at any depth  $Z$ ,
- $C_{vZ}$  = strength along the vertical plane at any depth  $Z$ ,
- $C_{ho}$  = strength along the horizontal plane at the surface,
- $C_{vo}$  = strength along the vertical plane at the surface, and
- $l_h$  and  $l_v$  = coefficients defining the variation of  $C_h$  and  $C_v$  over a significant depth  $H$ .

On substituting for  $C_{hZ}$  and  $C_{vZ}$  in the equation for  $C_\theta$ , the following expression for the undrained shear strength at any depth  $Z$  on any failure surface inclined at an angle  $\theta$  to the horizontal is obtained:

$$C_{\theta Z} = C_{ho} \left(1 + l_h \frac{Z}{H}\right) \cos^2 \theta + n_{co} \left(1 + l_v \frac{Z}{H}\right) \sin^2 \theta .$$

Control charts, providing a critical combination of stability number and tangent of friction angle were presented for a given slope ( $\beta = \tan^{-1} 1/25$ ) and various values of the coefficient of orthotropy, depth to hard layer, etc. Numerical results presented demonstrate the influence of strength anisotropy and strength increase with depth on the control charts.



CHAPTER III  
ANALYTICAL METHOD FOR STABILITY OF EARTH  
SLOPES IN NON-HOMOGENEOUS AND  
ANISOTROPIC SOILS

Introduction

It is evident from the review of published literature (Chapter II) that soil is neither homogeneous nor isotropic. Consequently, when analyzing the stability of earth slopes, this fact should be recognized and accounted for.

In this chapter, an analytical method is suggested for evaluating the stability of slopes in a two-layered system of anisotropic soils. The basic assumptions made in the analysis are listed below. Following this, the working formulae used in arriving at the factor of safety for the slope are derived in detail.

Basic Assumptions

1. The controlling potential surface of failure is either cylindrical or a combination of planar and cylindrical surfaces, as shown in Figure 5.
2. The soil in each layer is homogeneous with respect to shear strength.
3. The coefficient of anisotropy is the same at all points in the slope.



4. The anisotropic strength in each layer is characterized by the following equation:

$$C_i = C_h + (C_v - C_h) \cos^2 i$$

where

$C_i$  = shear strength along a plane inclined at an angle,  $i$ , to the vertical; and,

$C_h$  and  $C_v$  = shear strengths along the horizontal and vertical planes respectively.

5. The stability of the slope is analyzed by considering the stability of the individual layers.

#### Derivation of Working Formulae

The working formulae for the factor of safety are derived separately for the two layers.

#### Layer 2

There are two possible types of slip surfaces, as shown in Figure 6. These could be designated as Case (a) and Case (b). For each one of these two cases, the expressions for disturbing moment, resisting moment, and the factor of safety are derived in detail.

Case (a). For limiting equilibrium of the mass above the potential surface of rupture ADC (Figure 6a), the total disturbing moment about  $O_2$  must be equal to the total resisting moment about the same point.

While evaluating the disturbing moment for this case, the mass of soil above the interface ED is taken to be acting as surcharge. Hence, the disturbing moment due to this is considered in addition to

that contributed by the soil above the slip surface AD. The expressions for these two moments are derived below.

Let the disturbing moment due to surcharge load (EBCD) be  $M_{D_1}$ .

Weight of Soil Mass  $W_1$

$$= \gamma_1 H_1 H_2 (\cot \lambda_2 - \cot \beta) - \frac{H_1^2}{2} \cot \beta.$$

Let  $l_1$  be the lever arm for this mass about  $O_2$ .

$$\begin{aligned} l_1 &= \gamma_1 H_1 \left[ H_2 (\cot \lambda_2 - \cot \beta) - H_1 \cot \beta \right] \left[ R_2 \cos \alpha_2 \right. \\ &\quad \left. + \frac{H_1}{2} \cot \beta - \frac{H_2}{2} (\cot \lambda_2 - \cot \beta) \right] + \frac{\gamma_1 H_1^2}{2} \cot \beta \\ &\quad \times \left[ R_2 \cos \alpha_2 - H_2 (\cot \lambda_2 - \cot \beta) + \frac{2}{3} H_1 \cot \beta \right] \\ &\quad \div \gamma_1 \left[ H_1 H_2 (\cot \lambda_2 - \cot \beta) - H_1^2 \cot \beta + \frac{H_1^2}{2} \cot \beta \right]. \end{aligned}$$

Moment of EBCD about  $O_2$

$$= W_1 l_1.$$

$$\begin{aligned} M_{D_1} &= \gamma_1 \left[ H_1 \left\{ H_2 (\cot \lambda_2 - \cot \beta) - H_1 \cot \beta \right\} \left\{ R_2 \cos \alpha_2 \right. \right. \\ &\quad \left. \left. + \frac{H_1}{2} \cot \beta - \frac{H_2}{2} (\cot \lambda_2 - \cot \beta) \right\} + \frac{H_1^2}{2} \cot \beta \right. \\ &\quad \left. \times \left\{ R_2 \cos \alpha_2 - H_2 (\cot \lambda_2 - \cot \beta) + \frac{2}{3} H_1 \cot \beta \right\} \right]. \end{aligned}$$

Let the disturbing moment due to mass of soil enclosed in ADE (Figure 6a) be  $M_{D_2}$ .

Weight of soil mass (triangle AED)

$$= \frac{\gamma_2 H_2^2}{2} (\cot \lambda_2 - \cot \beta).$$

Weight of soil mass (segment AD)

$$= \frac{\gamma_2 R_2^2}{2} (\alpha' - \frac{1}{2} \sin 2\alpha').$$

Total weight of soil mass ( $W_2$ )

$$= \frac{\gamma_2 H_2^2}{2} (\cot \lambda_2 - \cot \beta) + \frac{\gamma_2 R_2^2}{2} (\alpha' - \frac{1}{2} \sin 2\alpha').$$

Let  $l_2$  be the lever arm for this mass about  $O_2$ .

$$\begin{aligned} l_2 = & \left[ \left\{ \frac{H_2^2}{2} (\cot \lambda_2 - \cot \beta) \right\} \left\{ R_2 \cos \alpha_2 - \frac{1}{3} (2H_2 \cot \lambda_2 \right. \right. \\ & \left. \left. - H_2 \cot \beta) \right\} + \left\{ R_2^2 \alpha' - \frac{R_2^2}{2} \sin (\alpha_3 - \alpha_2) \right\} \right. \\ & \left. \times \frac{2}{3} \frac{R_2 \sin^3 \alpha'}{(\alpha' - \sin \alpha' \cos \alpha')} \cos (\alpha_2 + \alpha') \right] \bigg/ \frac{H_2^2}{2} (\cot \lambda_2 \\ & - \cot \beta) + R_2^2 \alpha' - \frac{R_2^2}{2} \sin 2\alpha'. \end{aligned}$$

Moment of ADE about  $O_2$

$$= W_2 l_2.$$

$$\begin{aligned} M_{D_2} = & \gamma_2 \left[ \frac{H_2^2}{2} (\cot \lambda_2 - \cot \beta) \left\{ R_2 \cos \alpha_2 - \frac{1}{3} (2H_2 \cot \lambda_2 \right. \right. \\ & \left. \left. - H_2 \cot \beta) \right\} + \frac{2R_2^3}{3} \sin^3 \alpha' \cos (\alpha_2 + \alpha') \right]. \end{aligned}$$

The total disturbing moment ( $M_{D_{II}}$ ) is the sum of  $M_{D_1}$  and  $M_{D_2}$ .

$$\begin{aligned}
M_{D_{II}} = & \gamma_1 \left[ H_1 \left\{ H_2 (\cot \lambda_2 - \cot \beta) - H_1 \cot \beta \right\} \left\{ R_2 \cos \alpha_2 \right. \right. \\
& + \left. \frac{H_1}{2} \cot \beta - \frac{H_2}{2} (\cot \lambda_2 - \cot \beta) \right\} + \frac{H_1^2}{2} \cot \beta \\
& \times \left. \left\{ R_2 \cos \alpha_2 - H_2 (\cot \lambda_2 - \cot \beta) + \frac{2}{3} H_1 \cot \beta \right\} \right] \\
& + \gamma_2 \left[ \frac{H_2^2}{2} (\cot \lambda_2 - \cot \beta) \left\{ R_2 \cos \alpha_2 - \frac{1}{3} (2H_2 \cot \lambda_2 \right. \right. \\
& \left. \left. - H_2 \cot \beta) \right\} + \frac{2R_2^3}{3} \sin^3 \alpha' \cos (\alpha_2 + \alpha') \right].
\end{aligned}$$

From the geometry of the problem,

$$R_2 = \frac{H_2}{2 \sin \alpha' \sin \lambda_2},$$

$$\alpha_2 = 90 - (\lambda_1 + \alpha),$$

$$\frac{H_2}{H_1} = n$$

$$\frac{\gamma_2}{\gamma_1} = m.$$

Substituting these values in the above equation and simplifying,

$$\begin{aligned}
M_{D_{II}} = & \gamma_2 H_2^3 \left[ \frac{1}{2mn} (\cot \lambda_2 - \cot \beta) (\cot \alpha' + \cot \beta) \right. \\
& + \frac{1}{4mn} \cot \beta (\cot \lambda_2 - \cot \alpha' - 4 \cot \beta + 2 \cot^2 \beta) \\
& - \frac{1}{6mn} \cot^2 \beta + \frac{1}{12} (1 - 2 \cot^2 \beta + 3 \cot \lambda_2 \cot \alpha' \\
& \left. + 3 \cot \beta \cot \lambda_2 - 3 \cot \beta \cot \alpha') \right].
\end{aligned}$$

The resisting moment for this case consists of two parts,  $M_{R_1}$  and  $M_{R_2}$ .

$M_{R_1}$  = Resisting moment due to cohesion along CD,

$M_{R_2}$  = Resisting moment due to cohesion along AD.

$$M_{R_1} = (C_1)(\overline{CD}) \ell_1$$

where  $\ell_1$  = lever arm about  $O_2$ .

$$M_{R_1} = C_1 H_1 R_2 \cos \alpha_2.$$

From the geometry of the problem,

$$R_2 = \frac{H_2}{2 \sin \alpha' \sin \lambda_2}$$

$$\alpha_2 = 90 - (\lambda_2 + \alpha').$$

Substituting these values for  $R_2$  and  $\alpha_2$  in the above equation,

$$M_{R_1} = \frac{C_1 H_1 H_2 \sin (\lambda_2 + \alpha')}{2 \sin \alpha' \sin \lambda_2}.$$

Putting  $H_2/H_1 = n$ , and simplifying,

$$M_{R_1} = \frac{C_1 H_2^2}{n} (\cot \alpha' + \cot \lambda_2).$$

$$\begin{aligned} M_{R_2} &= R_2 \int_{\alpha_2}^{\alpha_3} C(\theta, Z) R_2 d\theta \\ &= R_2^2 \int_{\alpha_2}^{\alpha_3} \left[ C_2' + (C_1' - C_2') \right] \cos^2 i d\theta \\ &= R_2^2 \int_{\alpha_2}^{\alpha_3} C_1' \left[ \frac{C_2'}{C_1'} + \left(1 - \frac{C_2'}{C_1'}\right) \right] \cos^2 i d\theta. \end{aligned}$$

On integrating the above expression, and putting  $C_2'/C_1' = K'$ , the explicit value for  $M_{R_2}$  is obtained as

$$M_{R_2} = R_2^2 \left[ (1 + K') C_1' \alpha' - \frac{1}{2} (1 - K') C_1' \sin 2\alpha' \cos (2\alpha' - 2\lambda_2) \right].$$

On further simplification, the above expression for  $M_{R_2}$  reduces to

$$M_{R_2} = \frac{C_1' H_2^2}{8 \sin^2 \alpha' \sin^2 \lambda_2} \left[ 2(1 + K') \alpha' - (1 - K') \sin 2\alpha' \cos (2\alpha' - 2\lambda_2) \right].$$

The total resisting moment ( $M_{R_{II}}$ ) for this case is the sum of  $M_{R_1}$  and  $M_{R_2}$ .

$$M_{R_{II}} = \frac{C_1' H_2^2}{n} (\cot \alpha' + \cot \lambda_2) + \frac{C_1' H_2^2}{8 \sin^2 \alpha' \sin^2 \lambda_2} \times \left[ 2\alpha' (1 + K') - (1 - K') \sin^2 \alpha' \cos (2\alpha' - 2\lambda_2) \right].$$

Putting  $C_1' = (p + 1) C_1$ ,

$$M_{R_{II}} = C_1 H_2^2 \left[ \frac{1}{n} (\cot \alpha' + \cot \lambda_2) + \frac{(p + 1)}{8 \sin^2 \alpha' \sin^2 \lambda_2} \times \left\{ 2\alpha' (1 + K') - (1 - K') \sin 2\alpha' \cos (2\alpha' - 2\lambda_2) \right\} \right].$$

The factor of safety,  $F$ , is given by

$$F = \frac{\text{Resisting Moment } (M_{R_{II}})}{\text{Disturbing Moment } (M_{D_{II}})}.$$



$$\begin{aligned}
F = & C_1 H_2^2 \left[ \frac{1}{n} (\cot \alpha' + \cot \lambda_2) + \frac{(p+1)}{8 \sin^2 \alpha' \sin^2 \lambda_2} \right. \\
& \times \left. \left\{ 2\alpha' (1 + K') - (1 - K') \sin 2\alpha' \cos (2\alpha' - 2\lambda_2) \right\} \right] \\
& \div \gamma_2 H_2^3 \left[ \frac{1}{2mn} (\cot \lambda_2 - \cot \beta) (\cot \alpha' + \cot \beta) \right. \\
& + \frac{1}{4mn^2} \cot \beta (\cot \lambda_2 - \cot \alpha' - 4 \cot \beta + 2 \cot^2 \beta) \\
& - \frac{1}{6mn^3} \cot^2 \beta + \frac{1}{12} (1 - 2 \cot^2 \beta + 3 \cot \lambda_2 \cot \alpha' \\
& \left. + 3 \cot \beta \cot \lambda_2 - 3 \cot \beta \cot \alpha') \right].
\end{aligned}$$

The above expression for the factor of safety may be conveniently expressed as

$$F = \frac{C_1}{\gamma_2 H_2} N_2$$

where  $N_2$  is termed a stability number.

$$\begin{aligned}
N_2 = & \frac{\cot \alpha' + \cot \lambda_2}{2n} + \frac{(p+1)}{8 \sin^2 \alpha' \sin^2 \lambda_2} \left\{ 2\alpha' (1 + K') \right. \\
& \left. - (1 - K') \sin 2\alpha' \cos (2\alpha' - 2\lambda_2) \right\} \bigg/ \frac{1}{2mn} \\
& \times (\cot \lambda_2 - \cot \beta) (\cot \alpha' + \cot \beta) + \frac{1}{4mn^2} \cot \beta \\
& \times (\cot \lambda_2 - \cot \alpha' - 4 \cot \beta + 2 \cot^2 \beta) - \frac{1}{6mn^3} \\
& \times \cot^2 \beta + \frac{1}{12} (1 - 2 \cot^2 \beta + 3 \cot \lambda_2 \cot \alpha' \\
& + 3 \cot \beta \cot \lambda_2 - 3 \cot \beta \cot \alpha').
\end{aligned}$$

It is obvious that the minimum factor of safety is obtained by minimizing the stability number,  $N_2$ , with respect to  $\alpha'$  and  $\lambda_2$ , so that

$$\frac{\partial N_2}{\partial \alpha'} = 0$$

$$\frac{\partial N_2}{\partial \lambda_2} = 0.$$

The foregoing operations may be carried out with the aid of a computer, and  $N_2$  minimum would be a function of  $K'$ ,  $m$ ,  $n$ ,  $p$ , and  $\beta$ . For given values of  $K'$ ,  $m$ ,  $n$ , and  $p$ , the stability number is a function of  $\beta$  alone.

Computations have been carried out for values of  $K'$  ranging from 0.5 to 1.0,  $n$  ranging from 1.0 to 3.0, and  $p$  ranging from -0.5 to +0.5. Since the unit weights of soils in the two layers would not ordinarily differ greatly,  $m = \gamma_2/\gamma_1$  is assumed to be unity.

Case (b). The approach followed in arriving at the expressions for disturbing moment, resisting moment, and the factor of safety is the same as that for Case (a).

The total disturbing moment,  $M_{D_{II}}$ , is the sum of the disturbing moments  $M_{D_1}$  and  $M_{D_2}$ .

$$M_{D_1} = \gamma_1 \left\{ \frac{H_2^2}{2} (\cot \lambda_2 - \cot \beta) (\cot \lambda_2 \tan \beta - 1) \right\} \\ \times \left\{ R_2 \cos \alpha_2 - \frac{1}{3} (\cot \lambda_2 - \cot \beta) \right\};$$

$$M_{D_2} = \gamma_2 \left[ \frac{H_2^2}{2} (\cot \lambda_2 - \cot \beta) \left\{ R_2 \cos \alpha_2 - \frac{1}{3} (2H_2 \cot \lambda_2 \right. \right. \\ \left. \left. - H_2 \cot \beta) \right\} + \frac{2}{3} R_2^3 \sin^3 \alpha' \cos (\alpha^2 + \alpha') \right].$$

The total disturbing moment ( $M_{D_{II}}$ ) is found by

$$M_{D_{II}} = M_{D_1} + M_{D_2}.$$

$$\begin{aligned}
M_{D_{II}} = & \gamma_1 \left\{ \frac{H_2^2}{2} (\cot \lambda_2 - \cot \beta) (\cot \lambda_2 \tan \beta - 1) \right\} \\
& \times \left\{ R_2 \cos \alpha_2 - \frac{1}{3} (\cot \lambda_2 - \cot \beta) + \gamma_2 \left[ \frac{H_2^2}{2} (\cot \lambda_2 \right. \right. \\
& - \cot \beta) \left\{ R_2 \cos \alpha_2 - \frac{1}{3} (2H_2 \cot \lambda_2 - H_2 \cot \beta) \right\} \\
& \left. \left. + \frac{2}{3} R_2^3 \sin^3 \alpha' \cos (\alpha_2 + \alpha') \right] \right\}.
\end{aligned}$$

From the geometry of the problem,

$$\begin{aligned}
R_2 &= \frac{H_2}{2 \sin \alpha' \sin \lambda_2} \\
\alpha_2 &= 90 - (\lambda_2 + \alpha')
\end{aligned}$$

and as per the notation

$$\begin{aligned}
n &= \frac{H_2}{H_1} \\
m &= \frac{\gamma_2}{\gamma_1}.
\end{aligned}$$

Substituting these values in the above equation for the disturbing moment, and on simplifying,

$$\begin{aligned}
M_{D_{II}} = & \frac{\gamma_2 H_2^3}{12} \left[ \frac{1}{m} (\cot \lambda_2 - \cot \beta) (\cot \lambda \tan \beta - 1) (3 \cot \alpha' \right. \\
& + \cot \lambda_2 + 2 \cot \beta) + (\cot \lambda_2 - \cot \beta) (3 \cot \alpha' - \cot \lambda_2 \\
& \left. + 2 \cot \beta) + \operatorname{cosec}^2 \lambda_2 \right].
\end{aligned}$$

The resisting moment consists of two parts,  $M_{R_1}$  and  $M_{R_2}$ , as in Case (a). The total resisting moment ( $M_{R_{II}}$ ) is the sum of  $M_{R_1}$  and  $M_{R_2}$ .

$$M_{R_1} = \frac{C_1 H_2^2}{2} (\cot \lambda_2 \tan \beta - 1) (\cot \lambda_2 + \cot \alpha');$$

$$M_{R_2} = \frac{C_1' H_2^2}{8 \sin^2 \alpha' \sin^2 \lambda_2} \left[ 2\alpha' (1 + K') - (1 - K') \right. \\ \left. \times \sin 2\alpha' \cos (2\alpha' - 2\lambda_2) \right].$$

$$M_{R_{II}} = M_{R_1} + M_{R_2}$$

$$M_{R_{II}} = \frac{C_1 H_2^2}{2} (\cot \lambda_2 \tan \beta - 1) (\cot \lambda_2 + \cot \alpha') \\ + \frac{C_1' H_2^2}{8 \sin^2 \alpha' \sin^2 \lambda_2} \left[ 2\alpha' (1 + K') - (1 - K') \right. \\ \left. \times \sin^2 \alpha' \cos (2\alpha' - 2\lambda_2) \right].$$

Putting  $C_1' = (p + 1) C_1$ ,

$$M_{R_{II}} = C_1 H_2^2 \left[ \frac{1}{2} (\cot \lambda_2 \tan \beta - 1) (\cot \lambda_2 + \cot \alpha') \right. \\ \left. + \frac{(p + 1)}{8 \sin^2 \alpha' \sin^2 \lambda_2} \left\{ 2\alpha' (1 + K') - (1 - K') \right. \right. \\ \left. \left. \times \sin 2\alpha' \cos (2\alpha' - 2\lambda_2) \right\} \right].$$

The factor of safety,  $F$ , is given by

$$F = \frac{\text{Resisting Moment } (M_{R_{II}})}{\text{Disturbing Moment } (M_{D_{II}})}$$

$$\begin{aligned}
F = & C_1 H_2^2 \left[ \frac{1}{2} (\cot \lambda_2 \tan \beta - 1) (\cot \lambda_2 + \cot \alpha') \right. \\
& + \frac{(p+1)}{8 \sin^2 \alpha' \sin^2 \lambda_2} \left\{ 2\alpha' (1+K') - (1-K') \sin 2\alpha' \right. \\
& \times \left. \left. \cos (2\alpha' - 2\lambda_2) \right\} \right] \bigg/ \frac{\gamma_2 H_2^3}{12} \left[ \frac{1}{m} (\cot \lambda_2 - \cot \beta) \right. \\
& \times (\cot \lambda_2 \tan \beta - 1) (3 \cot \alpha' + \cot \lambda_2 + 2 \cot \beta) \\
& + (\cot \lambda_2 - \cot \beta) (3 \cot \alpha' - \cot \lambda_2 + 2 \cot \beta) \\
& \left. + \operatorname{cosec}^2 \lambda_2 \right].
\end{aligned}$$

The above expression for the factor of safety may be conveniently expressed as

$$F = \frac{C_1}{\gamma_2 H_2} N_2$$

where  $N_2$  is termed a stability number.

$$\begin{aligned}
N_2 = & \frac{1}{2} (\cot \lambda_2 \tan \beta - 1) (\cot \lambda_2 + \cot \alpha') + \frac{(p+1)}{8 \sin^2 \alpha' \sin^2 \lambda_2} \\
& \times \left\{ 2\alpha' (1+K') - (1-K') \sin 2\alpha' \cos (2\alpha' - 2\lambda_2) \right\} \\
& \div \frac{1}{12} \left[ \frac{1}{m} (\cot \lambda_2 - \cot \beta) (\cot \lambda_2 \tan \beta - 1) (3 \cot \alpha' \right. \\
& + \cot \lambda_2 + 2 \cot \beta) + (\cot \lambda_2 - \cot \beta) (3 \cot \alpha' - \cot \lambda_2 \\
& \left. + 2 \cot \beta) + \operatorname{cosec}^2 \lambda_2 \right].
\end{aligned}$$

It is obvious that the minimum factor of safety is obtained by minimizing the stability number,  $N_2$ , with respect to  $\alpha'$  and  $\lambda_2$ , so that

$$\frac{\partial N_2}{\partial \alpha'} = 0$$

$$\frac{\partial N_2}{\partial \lambda_2} = 0 .$$

The foregoing operations may be carried out with the aid of a computer; and  $N_2$  minimum would be a function of  $K'$ ,  $m$ ,  $p$ , and  $\beta$ . For given values of  $K'$ ,  $m$ , and  $p$ , the stability number is a function of  $\beta$  alone.

Computations have been carried out for a given set of  $K'$ ,  $m$ , and  $p$ , as listed in Case (a).

### Layer 1

The failure surface is circular, as shown in Figure 5. The solution for this case is available (Lo, 1965), but to keep the clarity and continuity of the present analysis, this is listed in detail.

For limiting equilibrium of the mass above the potential surface of rupture EF, the total disturbing moment about  $O_1$  must be equal to the total resisting moment about the same point.

The disturbing moment of the mass of soil above EF is equal to  $M_{D_I}$ .

$$M_{D_I} = \frac{\gamma_1 H_1^3}{12} (1 - 2 \cot^2 \beta + 3 \cot \lambda_1 \cot \beta + 3 \cot \alpha$$

$$\times \cot \lambda_1 - 3 \cot \alpha \cot \beta).$$

The resisting moment ( $M_{R_I}$ ) is given by

$$\begin{aligned} M_{R_I} &= R_1 \int_{\alpha_1}^{\alpha_2} C(\theta, Z) R_1 d\theta \\ &= R_1^2 \int_{\alpha_1}^{\alpha_2} \{C_2 + (C_1 - C_2)\} \cos^2 i d\theta. \end{aligned}$$

On integration and simplification, the expression for  $M_{R_I}$  would be

$$\begin{aligned} M_{R_I} &= \frac{C_1 H_1^2}{4 \sin^2 \alpha \sin^2 \lambda_1} \left[ (1 + K) \alpha + \frac{1}{2} (1 - K) \sin 2\alpha \right. \\ &\quad \left. \times \cos (2f - 2\lambda) \right]. \end{aligned}$$

The factor of safety,  $F$ , is given by

$$F = \frac{\text{Resisting Moment } (M_{R_I})}{\text{Disturbing Moment } (M_{D_I})}$$

$$\begin{aligned} F &= \frac{C_1 H_1^2}{4 \sin^2 \alpha \sin^2 \lambda_1} \left[ (1 + K) \alpha + \frac{1}{2} (1 - K) \sin^2 \alpha \right. \\ &\quad \left. \times \cos (2f - 2\lambda) \right] \Bigg/ \frac{\gamma_1 H_1^3}{12} (1 - 2 \cot^2 \beta + 3 \cot \lambda_1 \\ &\quad \times \cot \beta + 3 \cot \alpha \cot \lambda_1 - 3 \cot \alpha \cot \beta). \end{aligned}$$

For isotropic material,  $C_1 = C_i = C_2$ , and the above expression for the factor of safety reduces to

$$\begin{aligned} F &= \frac{6 \alpha C_1}{\gamma_1 H_1 \sin^2 \alpha \sin \lambda_1} \left[ (1 + K) \alpha + \frac{1}{2} (1 - K) \sin^2 \alpha \right. \\ &\quad \left. \times \cos (2f - 2\lambda) \right] \Bigg/ \frac{\gamma_1 H_1^3}{12} (1 - 2 \cot^2 \beta + 3 \cot \lambda_1 \\ &\quad \times \cot \beta + 3 \cot \alpha \cot \lambda_1 - 3 \cot \alpha \cot \beta). \end{aligned}$$

This equation is identical to Taylor's solution for the case  $\varphi = 0$  (Taylor, 1937).

The above equation for  $F$  may be conveniently written as

$$F = \frac{C_1}{\gamma_1 H_1} N_1$$

where  $N_1$  is termed a stability number.

$$\begin{aligned} N_1 = & 3 \left[ (1 + K) \alpha + \frac{1}{2} (1 - K) \sin 2\alpha \cos (2f - 2\lambda_1) \right] \\ & \div \sin^2 \alpha \sin^2 \lambda_1 (1 - 2 \cot^2 \beta + 3 \cot \lambda_1 \cot \beta \\ & + 3 \cot \alpha \cot \lambda_1 - 3 \cot \alpha \cot \beta). \end{aligned}$$

It is obvious that the minimum factor of safety is obtained by minimizing the stability number  $N_1$  with respect to  $\alpha$  and  $\lambda_1$ , so that

$$\frac{\partial N_1}{\partial \alpha} = 0$$

$$\frac{\partial N_1}{\partial \lambda_1} = 0.$$

The foregoing operations may be carried out with the aid of a computer; and  $N_1$  is solely a function of  $K$ ,  $f$ , and  $\beta$ . For given values of  $K$  and  $f$ , the stability number is a function of  $\beta$  alone.

Computations have been carried out for  $f = 55^\circ$  and  $K$  ranging from 0.5 to 1.0. The value of  $55^\circ$  for  $f$  is based on experimental data (Lo, 1965).

### Charts

Numerical results, which are graphically presented in the following pages (Figures 7 through 29), are obtained with the aid of an IBM 360/65 computer available at Oklahoma State University.



Figure 30 is a reproduction of the chart presented by Lo.(1965). These charts can be used to solve slope stability problems in a two-layered system of anisotropic soils.

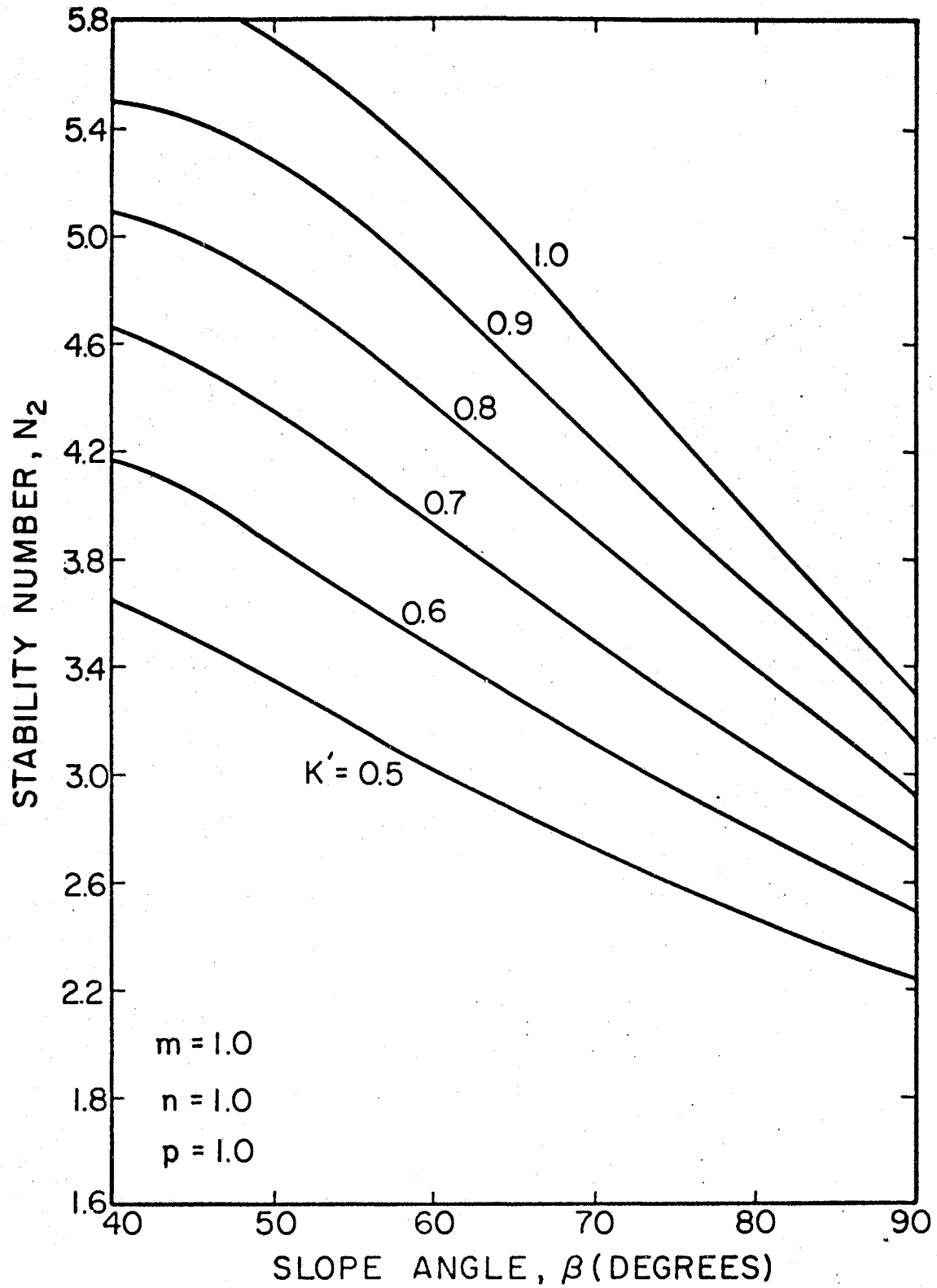


Figure 7. Slope Angle ( $\beta$ ) versus Stability Number ( $N_2$ ),  
Layer 2 ( $m = 1.0$ ,  $n = 1.0$ , and  $p = 1.0$ )

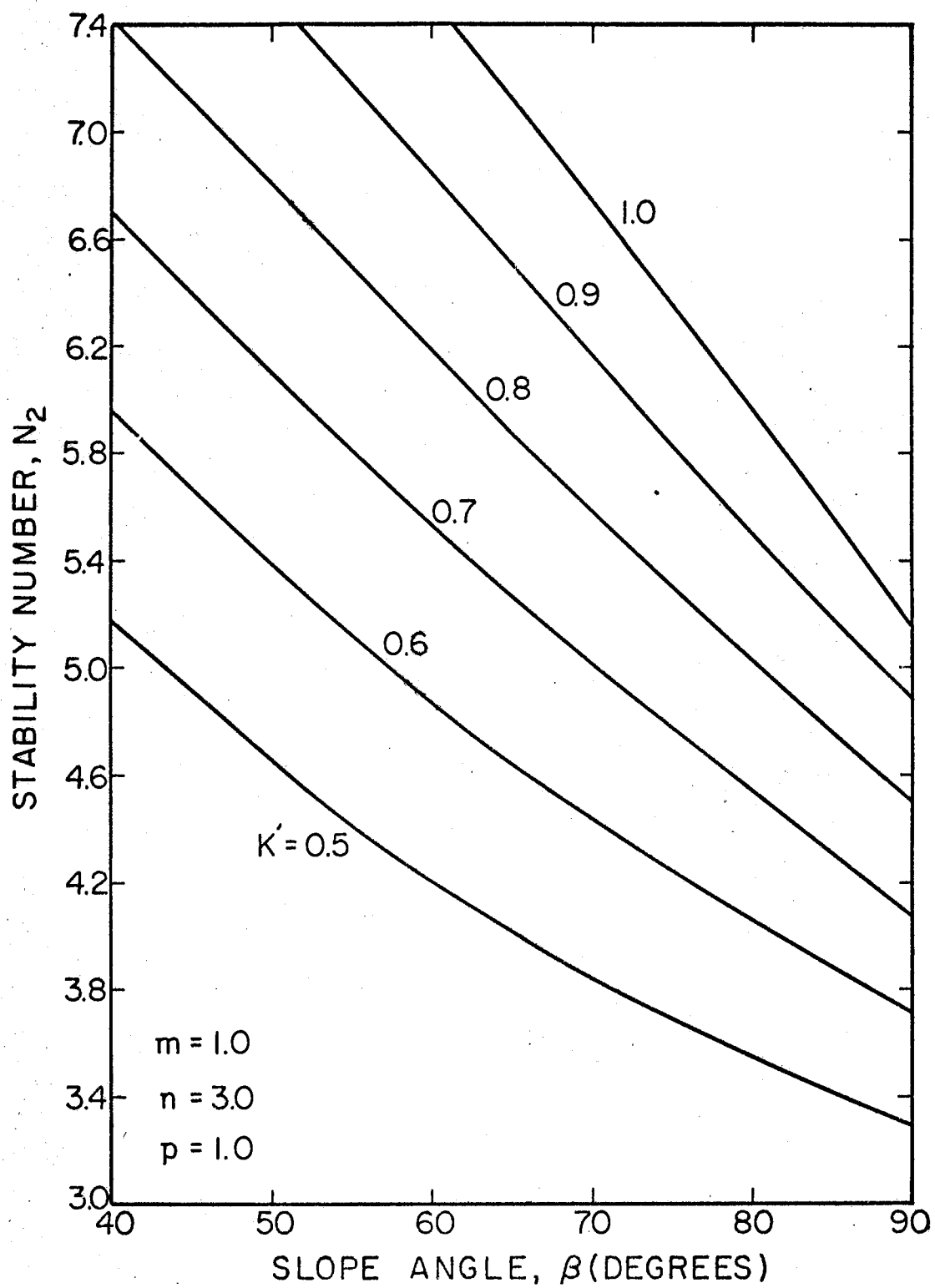


Figure 8. Slope Angle ( $\beta$ ) versus Stability Number ( $N_2$ ), Layer 2 ( $m = 1.0$ ,  $n = 3.0$ , and  $p = 1.0$ )<sup>2</sup>

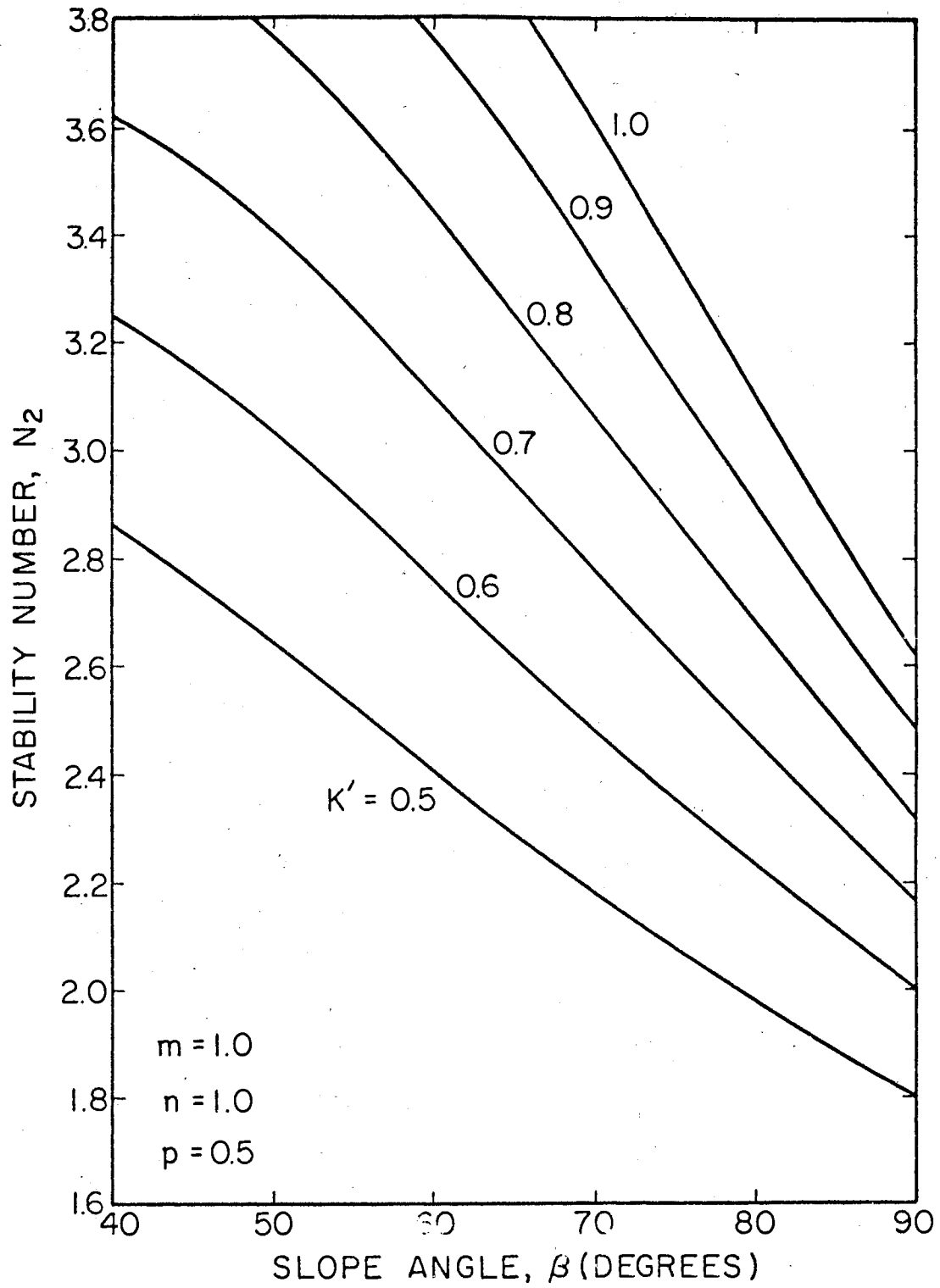


Figure 9. Slope Angle ( $\beta$ ) versus Stability Number ( $N_2$ ),  
Layer 2 ( $m = 1.0$ ,  $n = 1.0$ , and  $p = 0.5$ )

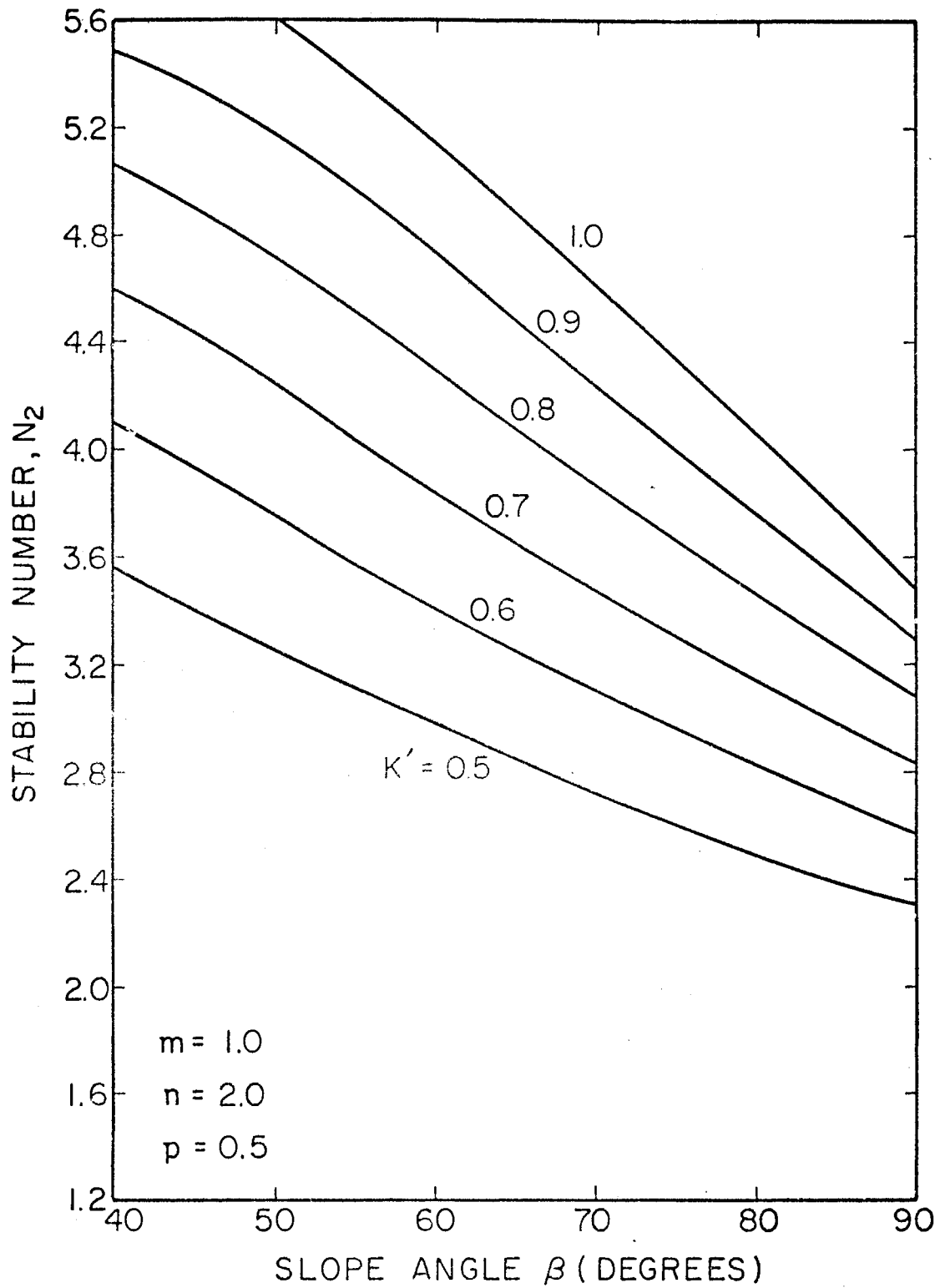


Figure 10. Slope Angle ( $\beta$ ) versus Stability Number ( $N_2$ ), Layer 2 ( $m = 1.0$ ,  $n = 2.0$ , and  $p = 0.5$ )

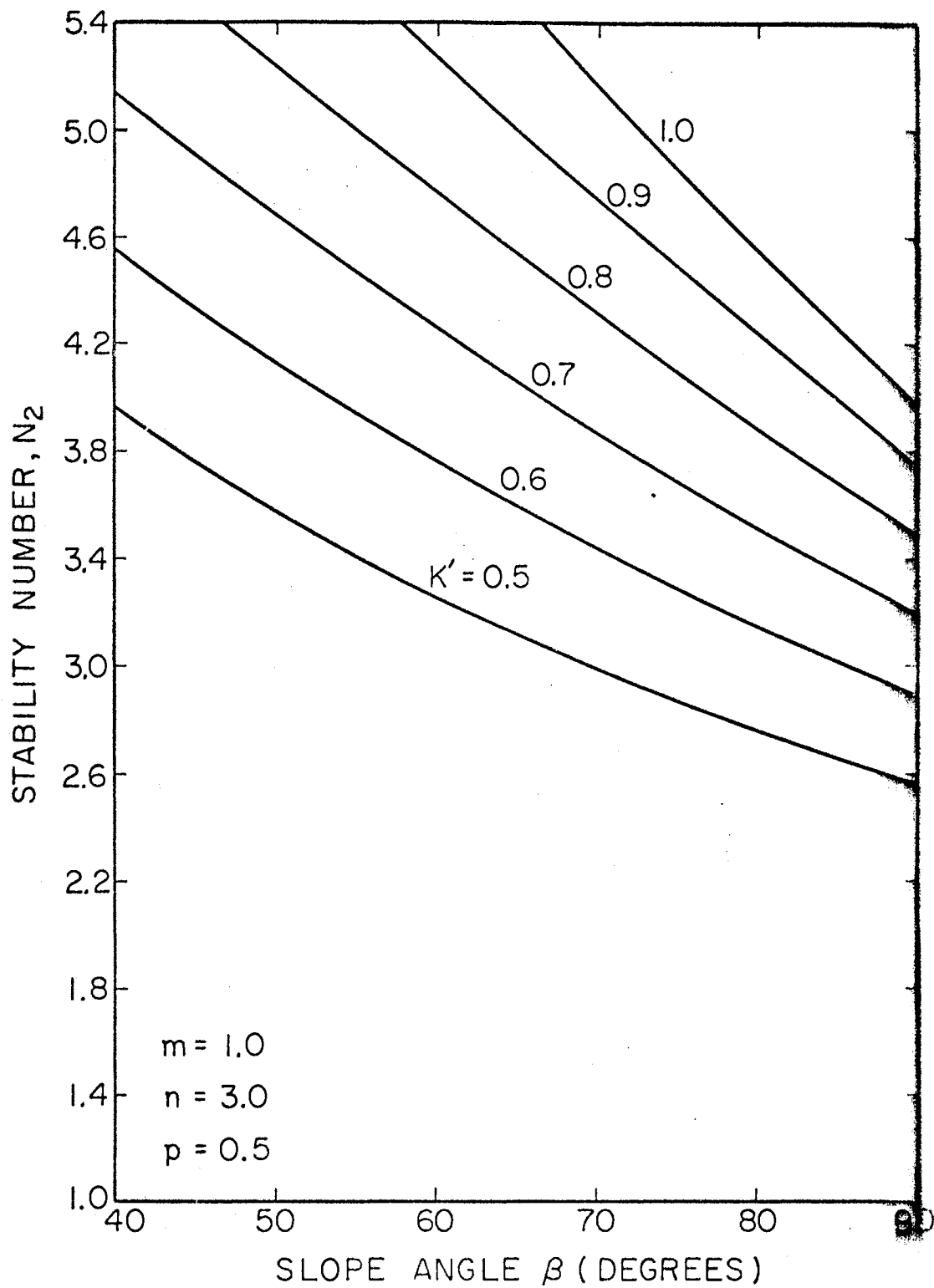


Figure 11. Slope Angle ( $\beta$ ) versus Stability Number ( $N_2$ ), Layer 2 ( $m = 1.0$ ,  $n = 3.0$ , and  $p = 0.5$ )

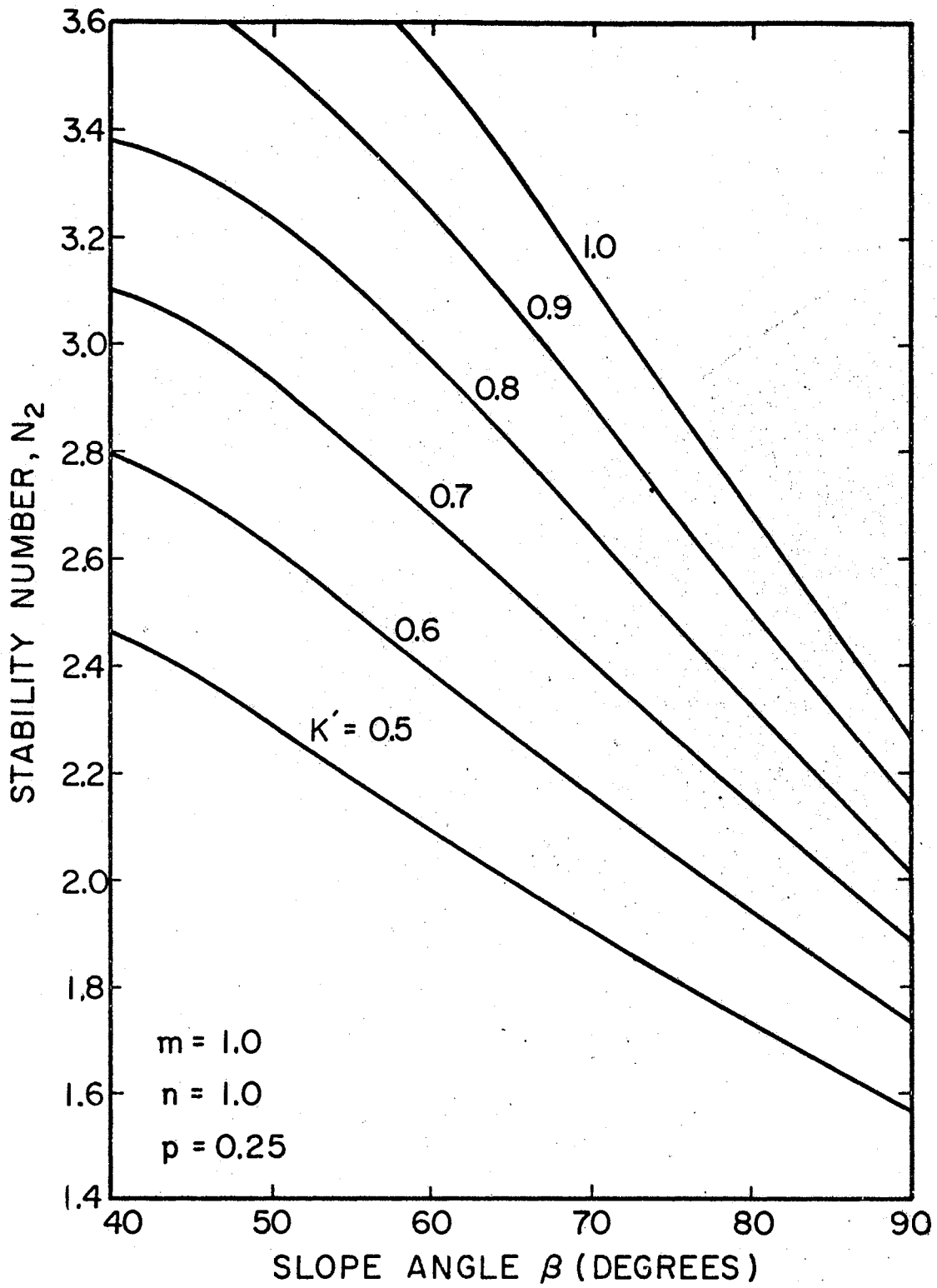


Figure 12. Slope Angle ( $\beta$ ) versus Stability Number ( $N_2$ ),  
 Layer 2 ( $m = 1.0$ ,  $n = 1.0$ , and  $p = 0.25$ )

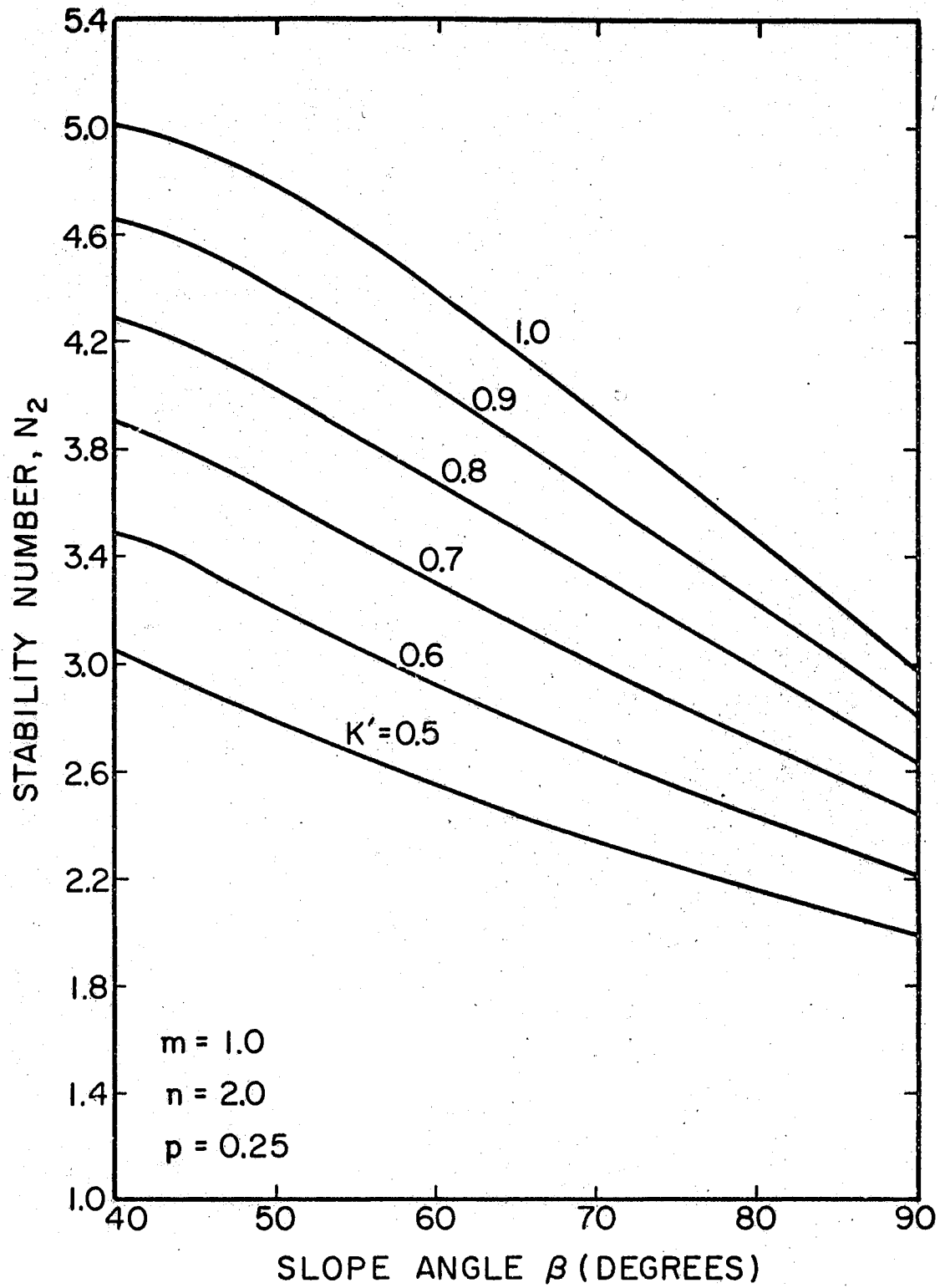


Figure 13. Slope Angle ( $\beta$ ) versus Stability Number ( $N_2$ ),  
Layer 2 ( $m = 1.0$ ,  $n = 2.0$ , and  $p = 0.25$ )



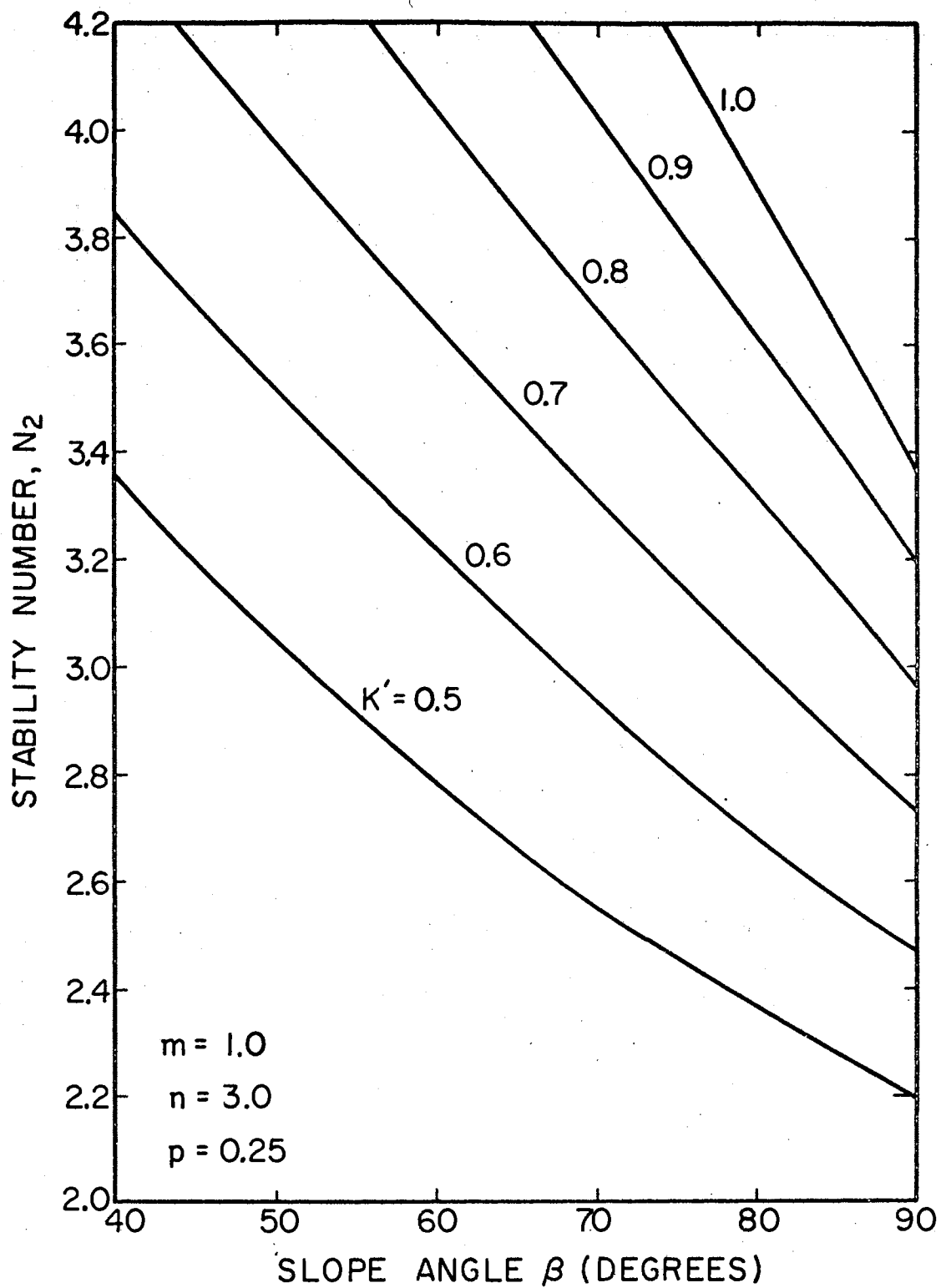


Figure 14. Slope Angle ( $\beta$ ) versus Stability Number ( $N_2$ ),  
Layer 2 ( $m = 1.0$ ,  $n = 3.0$ , and  $p = 0.25$ )

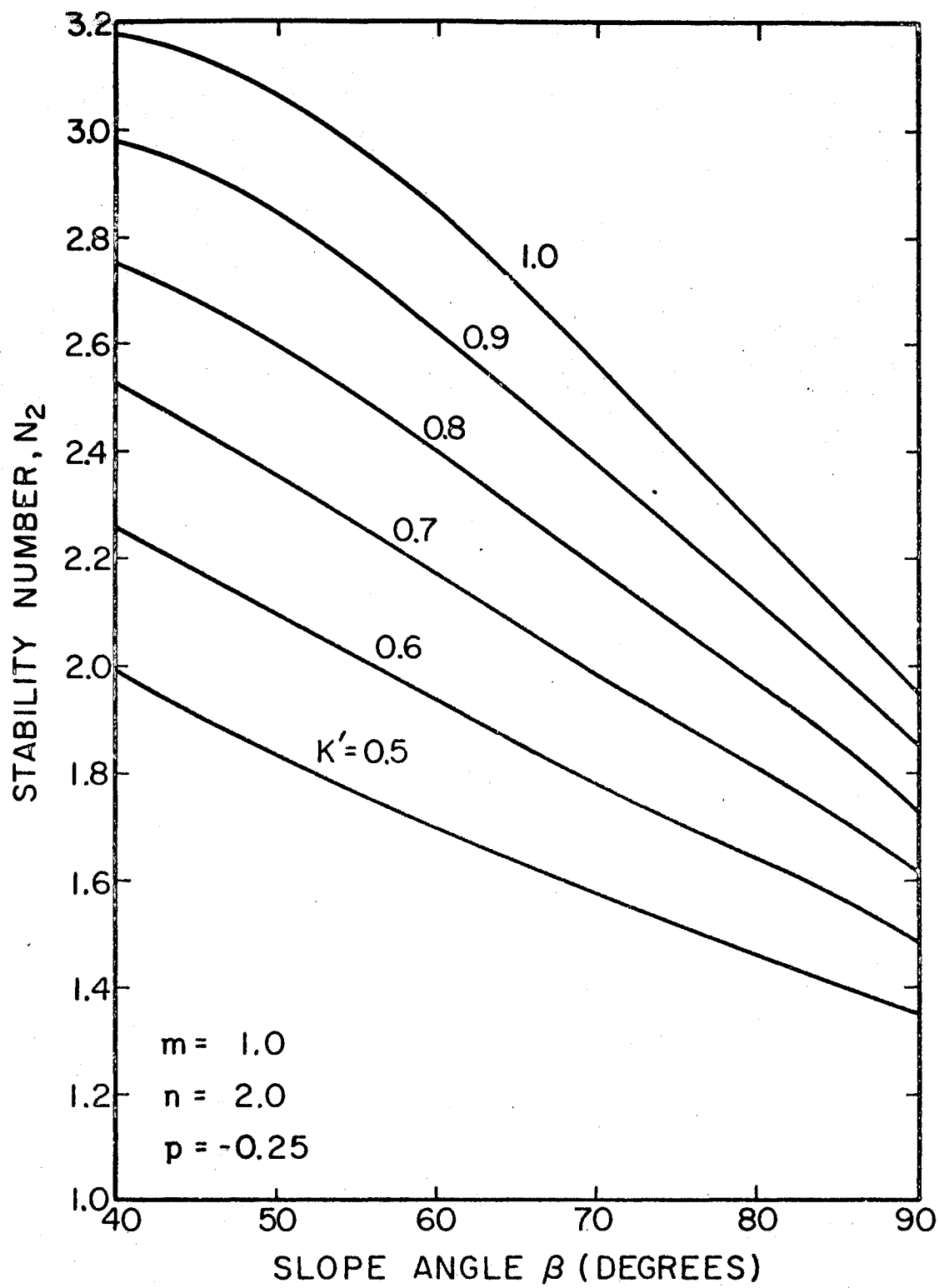


Figure 15. Slope Angle ( $\beta$ ) versus Stability Number ( $N_2$ ), Layer 2, ( $m = 1.0$ ,  $n = 2.0$ , and  $p = -0.25$ )<sup>2</sup>

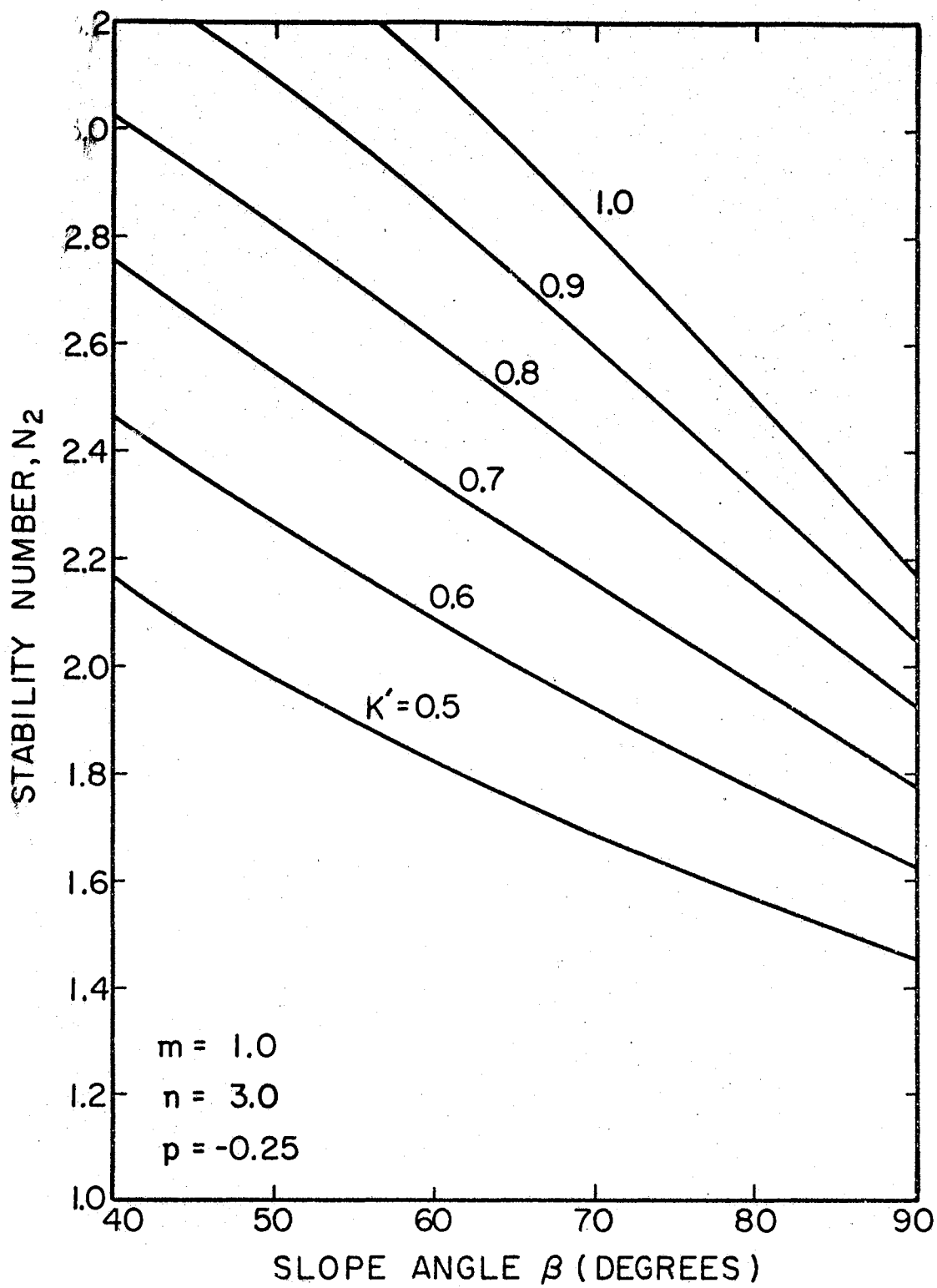


Figure 16. Slope Angle ( $\beta$ ) versus Stability Number ( $N_2$ ), Layer 2 ( $m = 1.0$ ,  $n = 3.0$ , and  $p = -0.25$ )

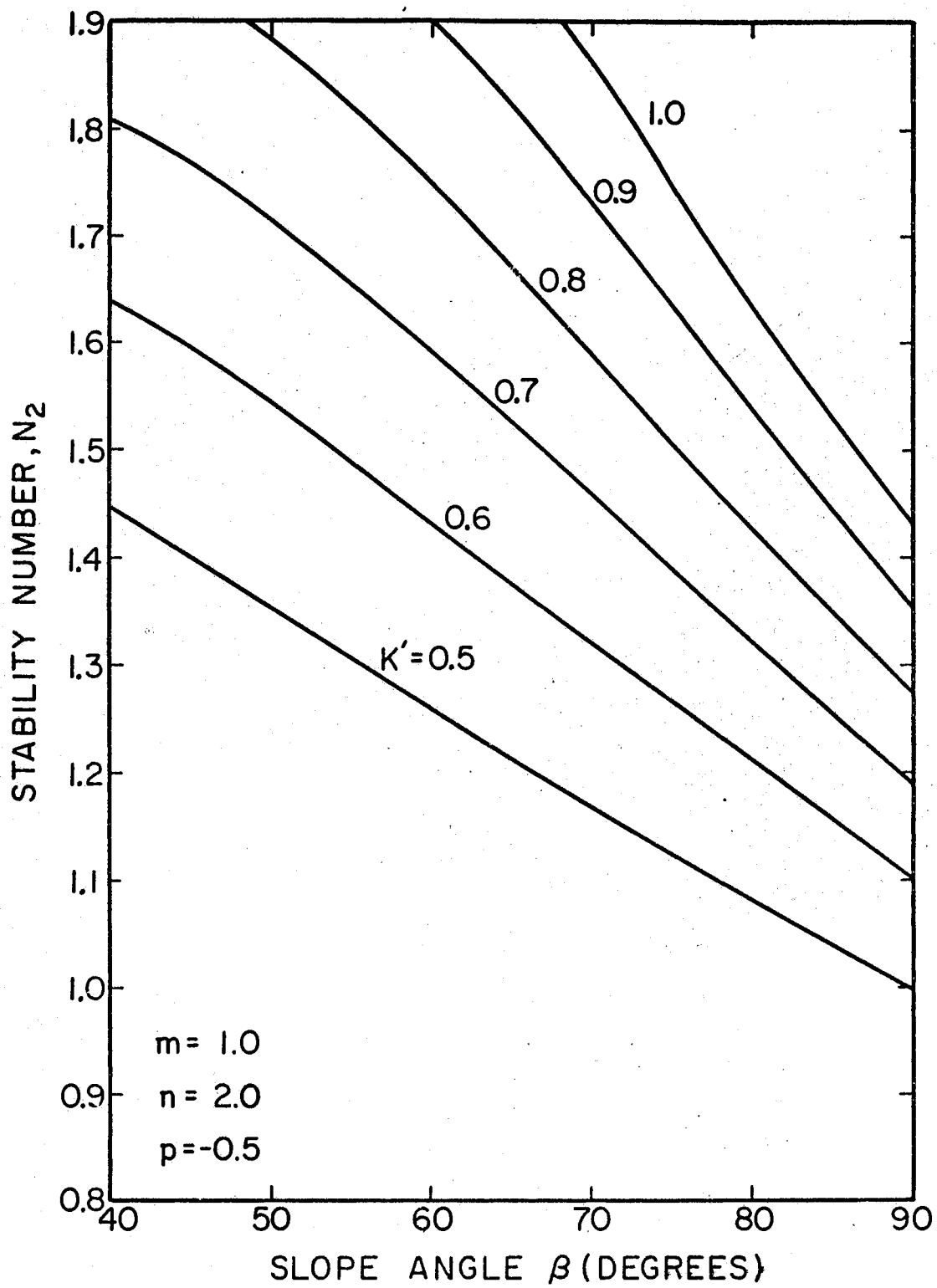


Figure 17. Slope Angle ( $\beta$ ) versus Stability Number ( $N_2$ ),  
 Layer 2 ( $m = 1.0$ ,  $n = 2.0$ , and  $p = -0.5$ )

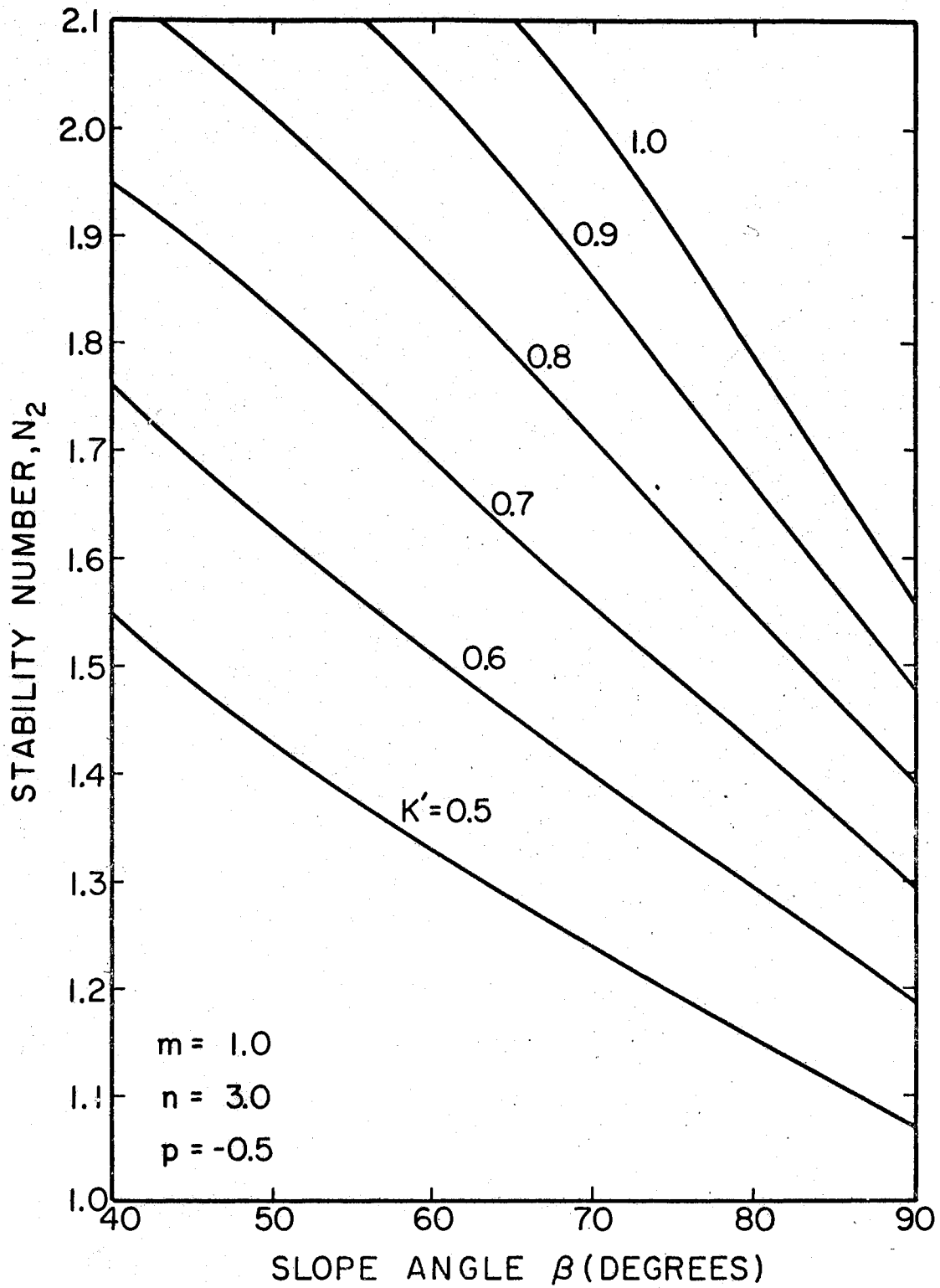


Figure 18. Slope Angle ( $\beta$ ) versus Stability Number ( $N_2$ ),  
 Layer 2 ( $m = 1.0$ ,  $n = 3.0$ , and  $p = -0.5$ )

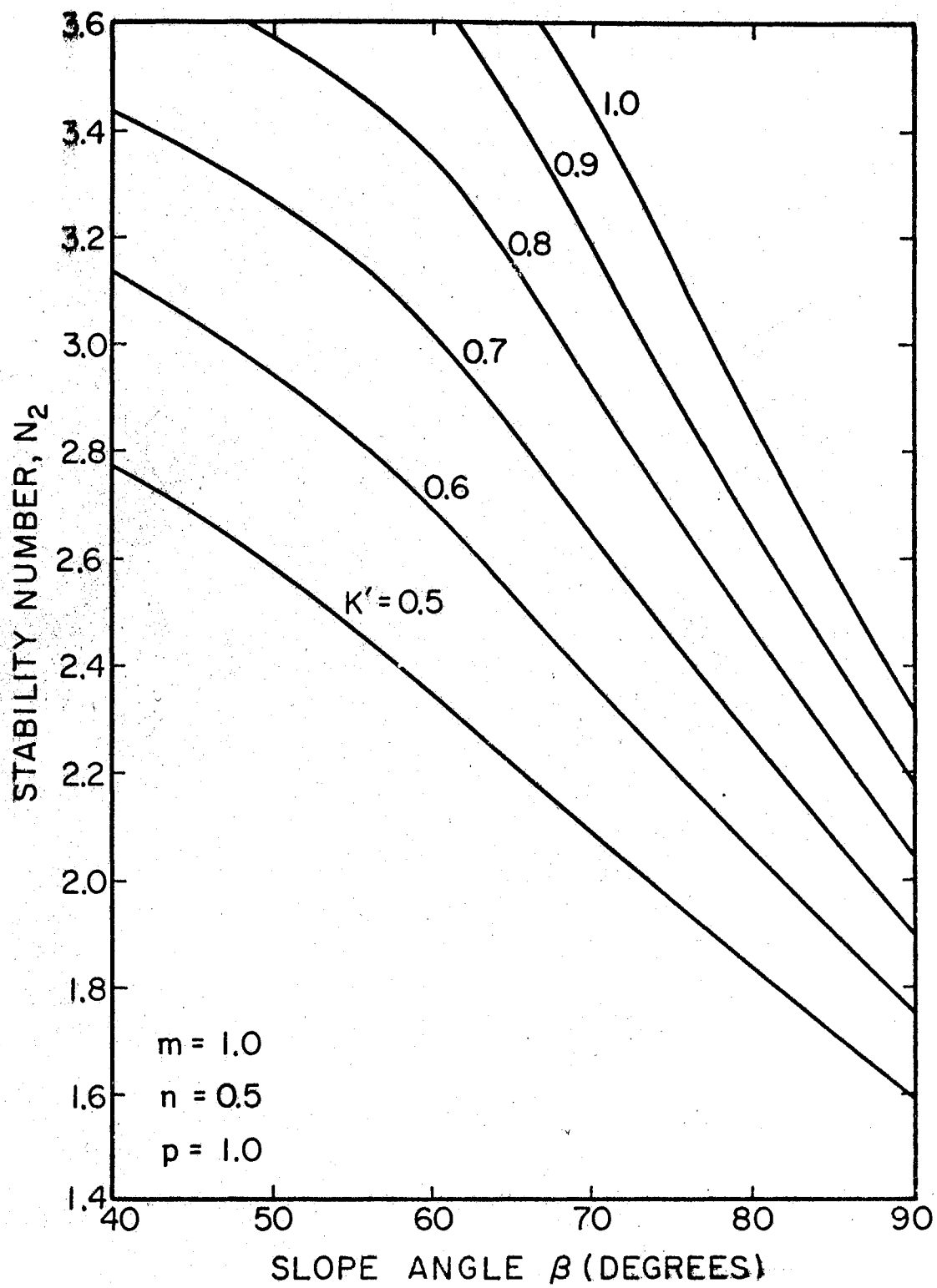


Figure 19. Slope Angle ( $\beta$ ) versus Stability Number ( $N_2$ ), Layer 2 ( $m = 1.0$ ,  $n = 0.5$ , and  $p = 1.0$ )

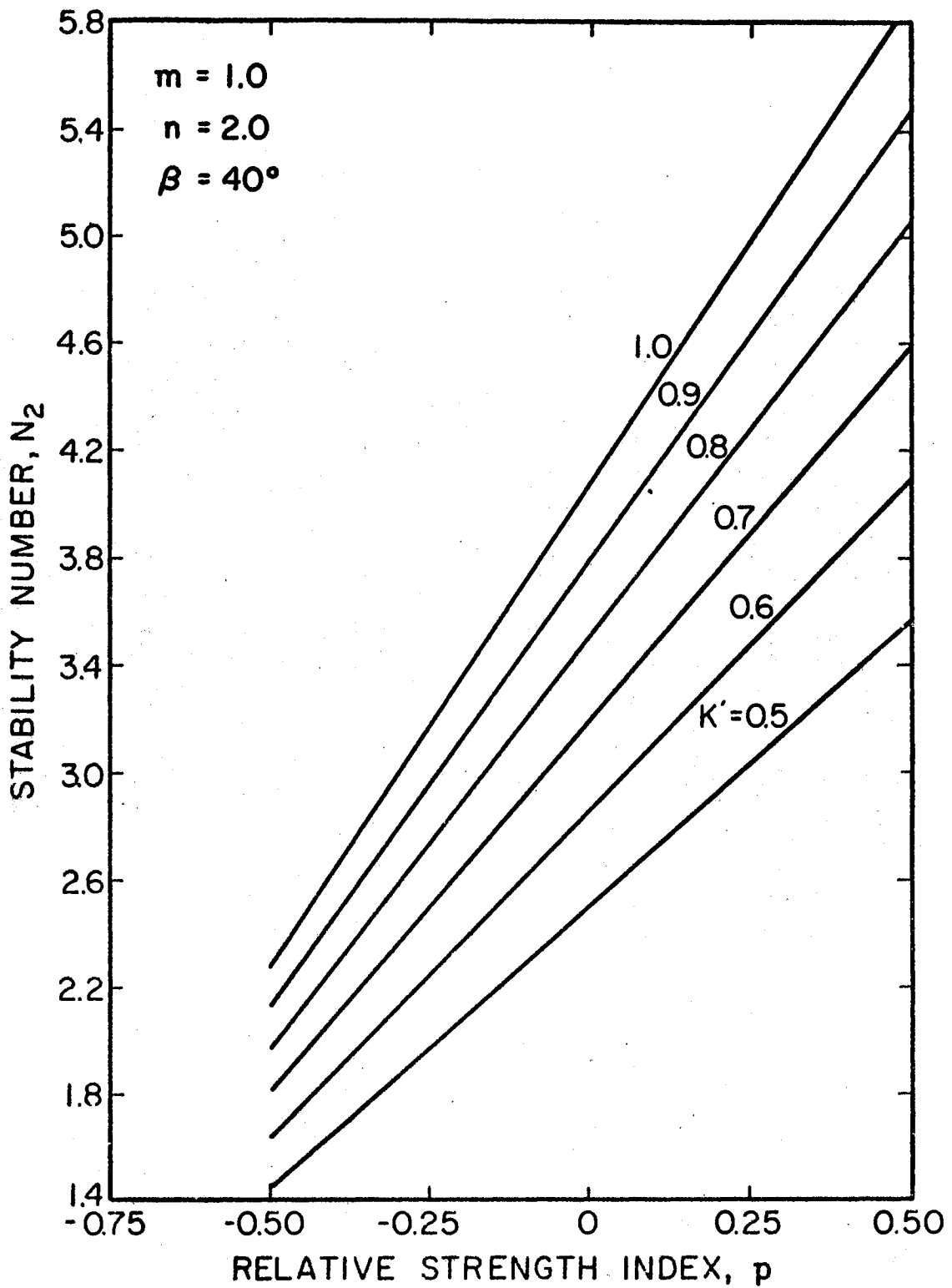


Figure 20. Relative Strength Index ( $p$ ) versus Stability Number ( $N_2$ ) ( $m = 1.0$ ,  $n = 2.0$ , and  $\beta = 40^\circ$ )

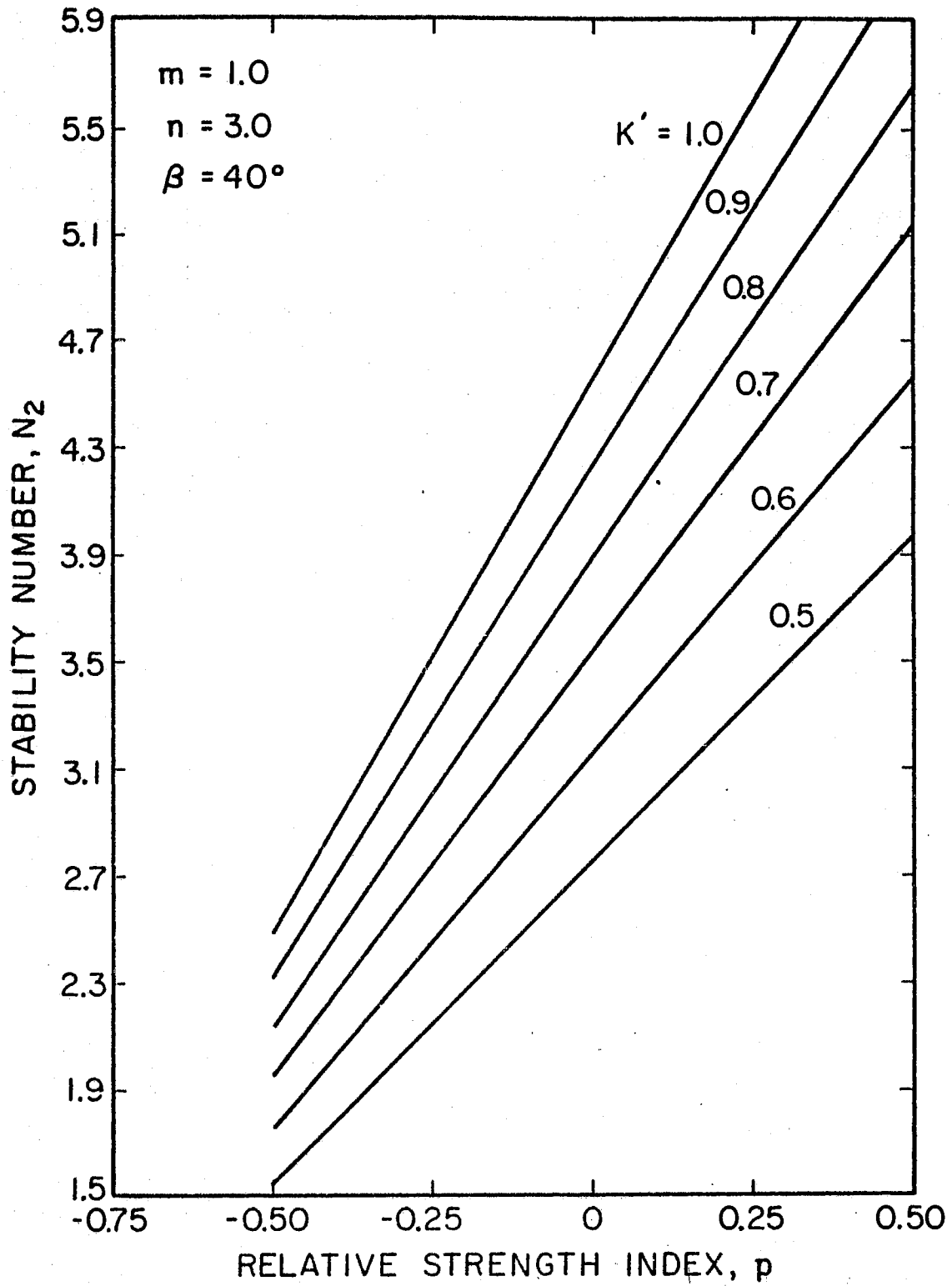


Figure 21. Relative Strength Index ( $p$ ) versus Stability Number ( $N_2$ ) ( $m = 1.0$ ,  $n = 3.0$ , and  $\beta = 40^\circ$ )



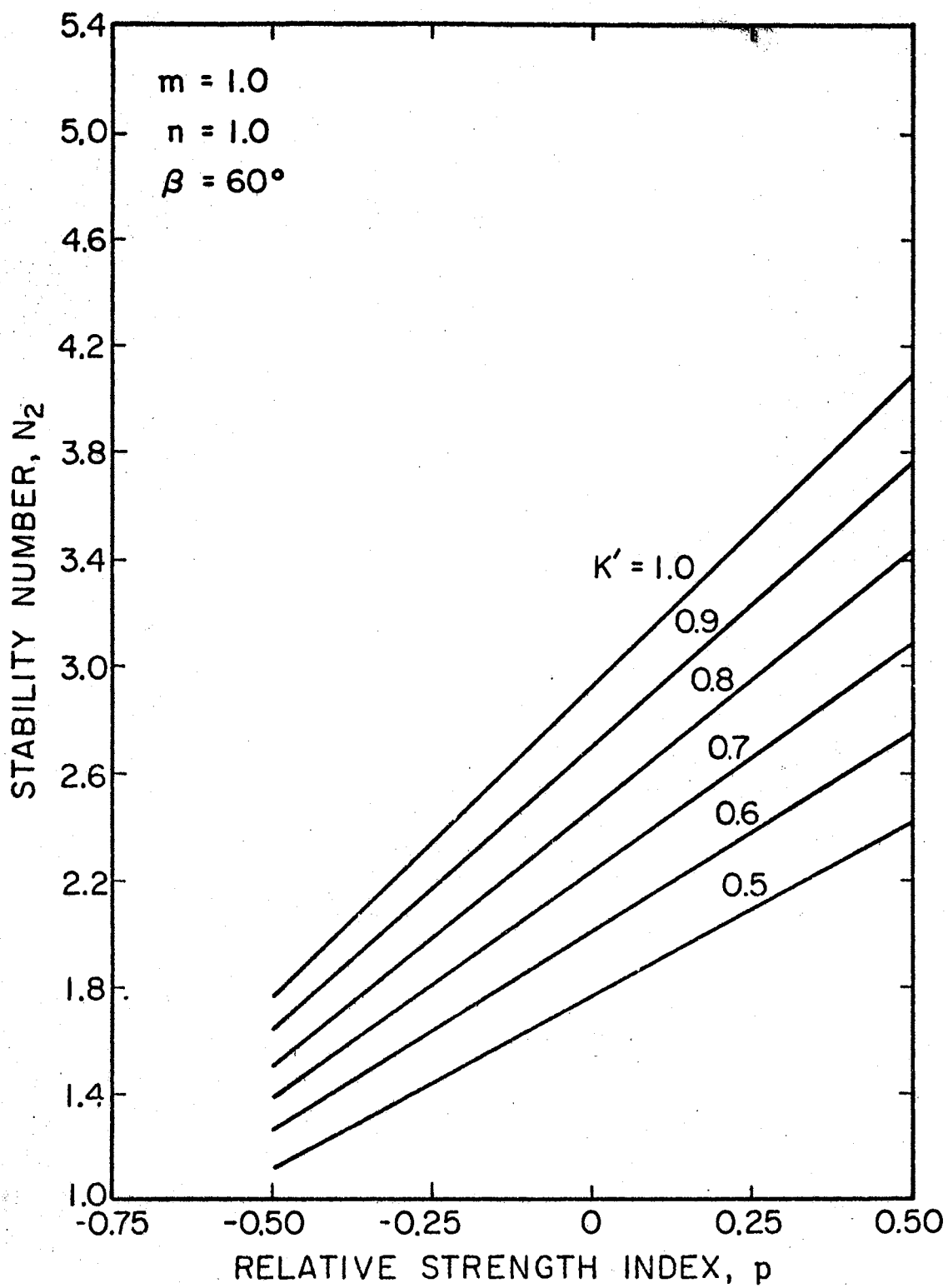


Figure 22. Relative Strength Index ( $p$ ) versus Stability Number ( $N_2$ ) ( $m = 1.0$ ,  $n = 1.0$ , and  $\beta = 60^\circ$ )

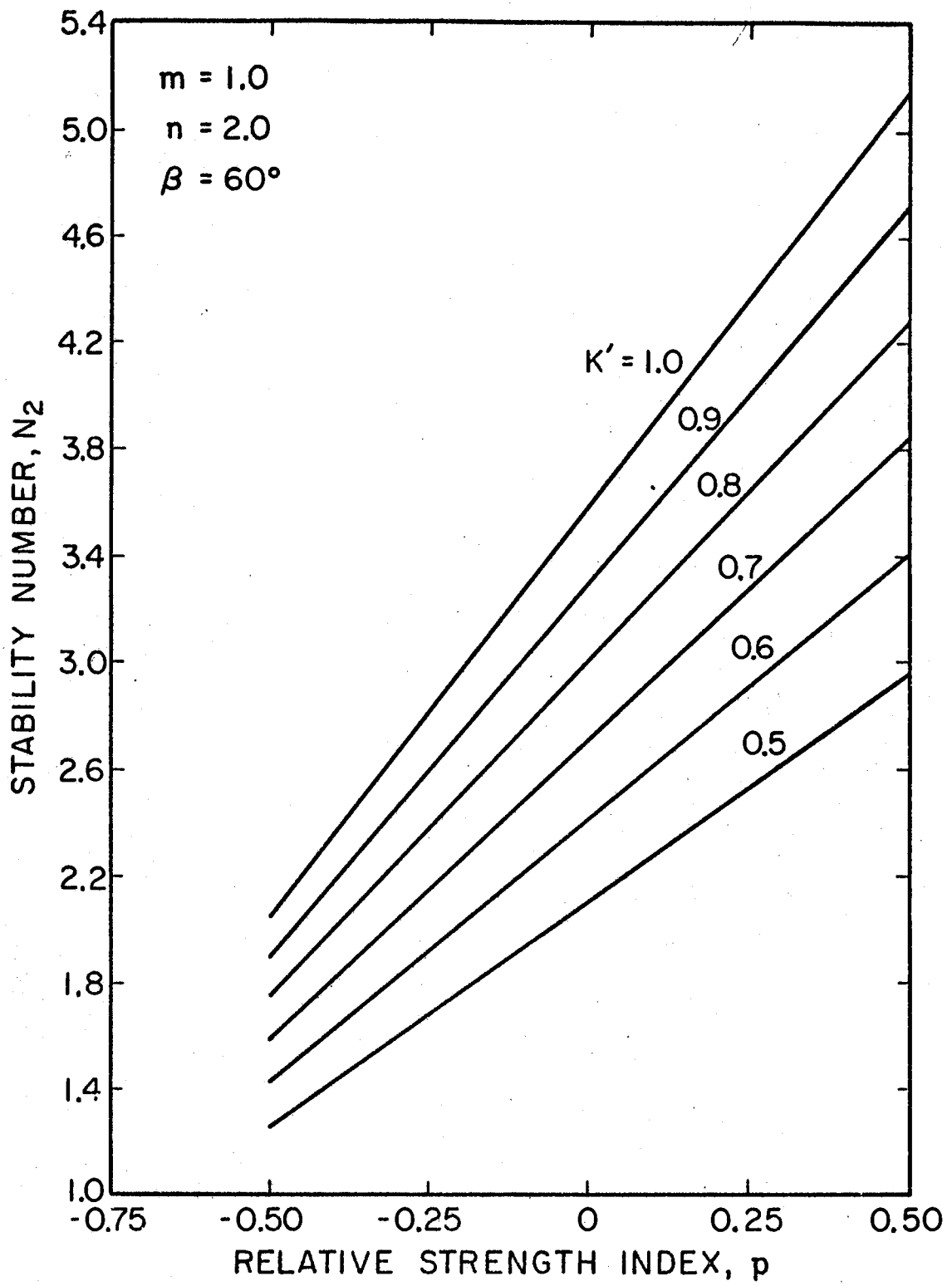


Figure 23. Relative Strength Index ( $p$ ) versus Stability Number ( $N_2$ ) ( $m = 1.0$ ,  $n = 2.0$ , and  $\beta = 60^\circ$ )

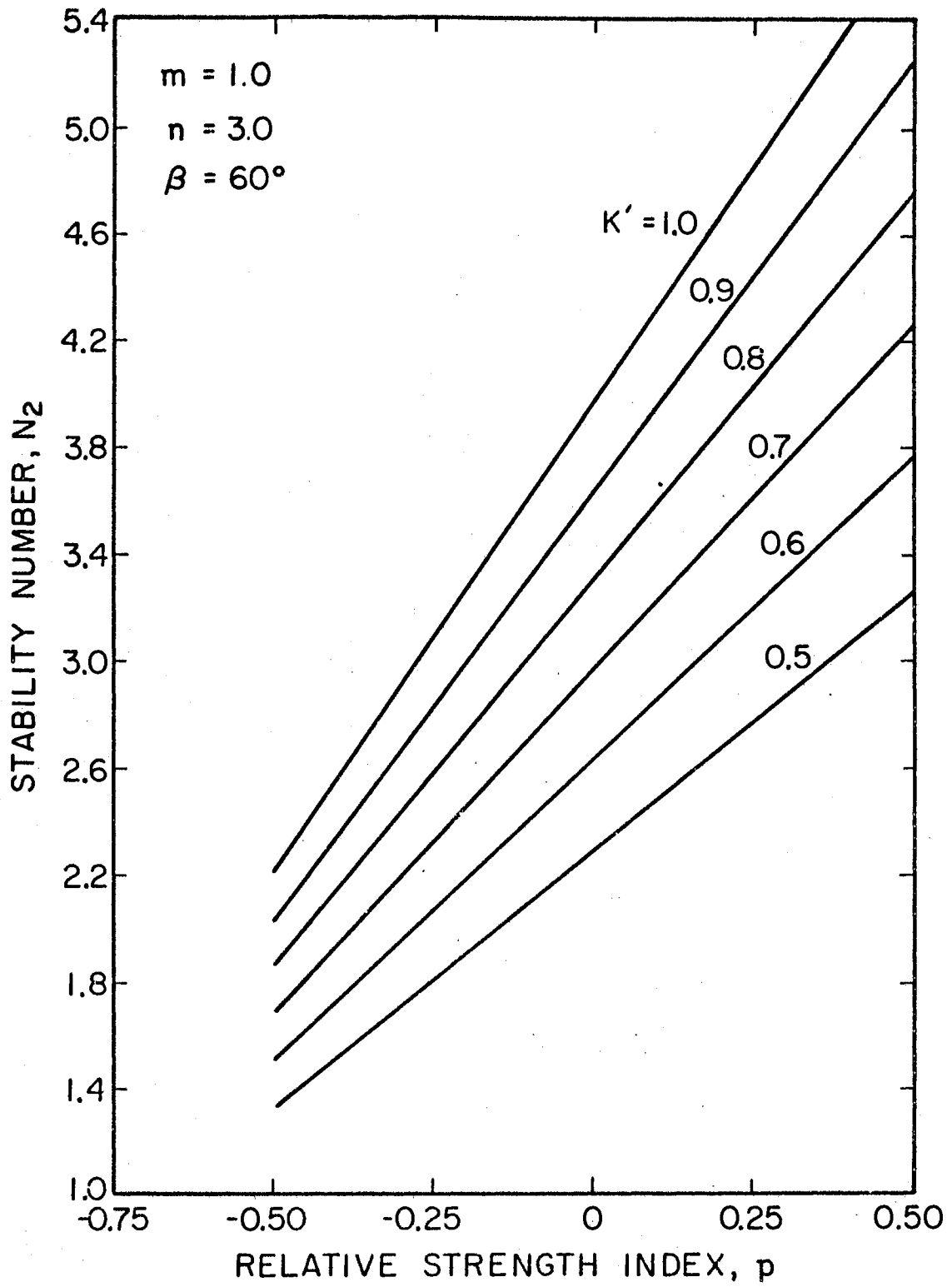


Figure 24. Relative Strength Index ( $p$ ) versus Stability Number ( $N_2$ ) ( $m = 1.0$ ,  $n = 3.0$ , and  $\beta = 60^\circ$ )

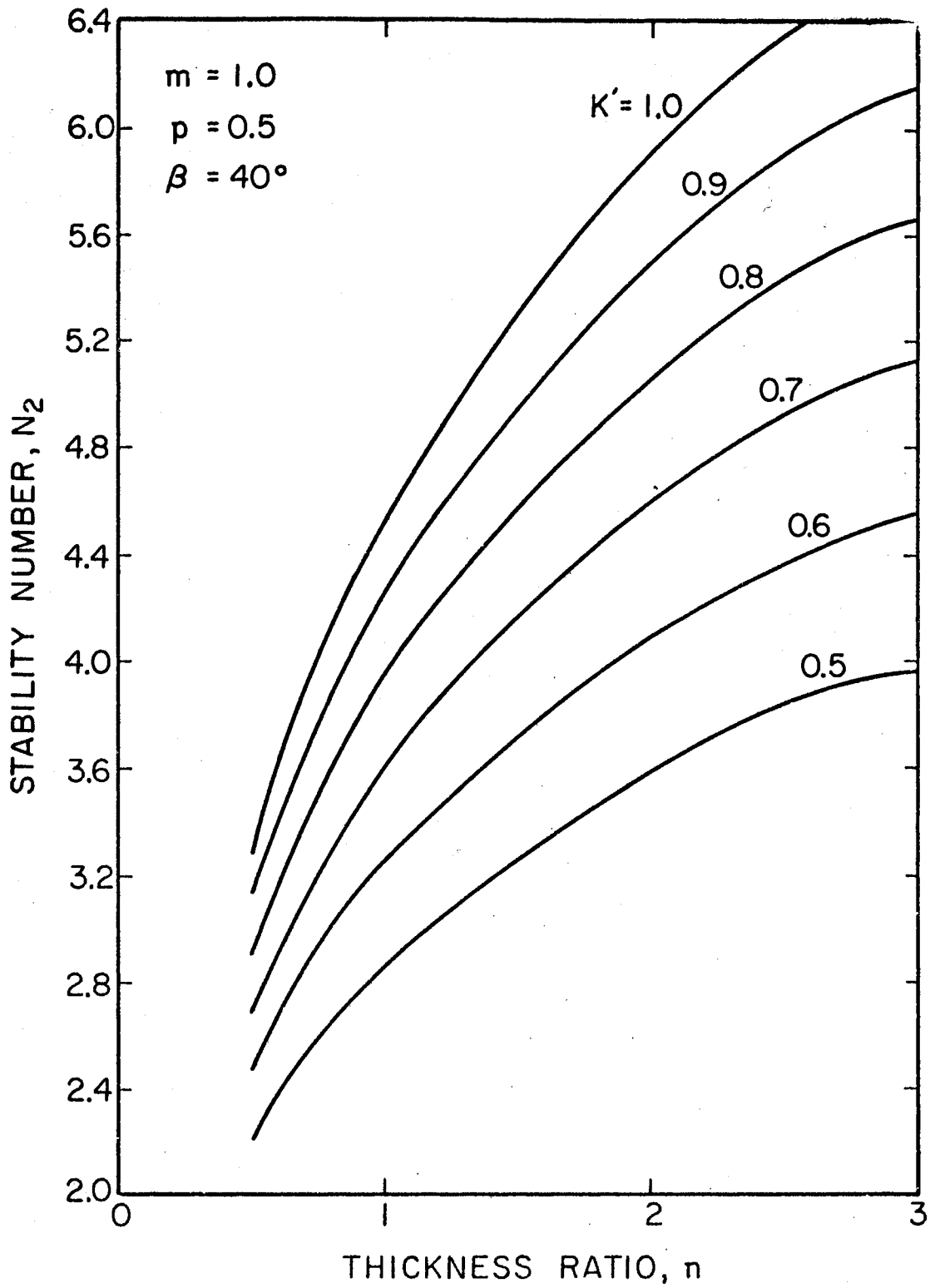


Figure 25. Thickness Ratio ( $n$ ) versus Stability Number ( $N_2$ ), ( $m = 1.0$ ,  $p = 0.5$ , and  $\beta = 40^\circ$ )

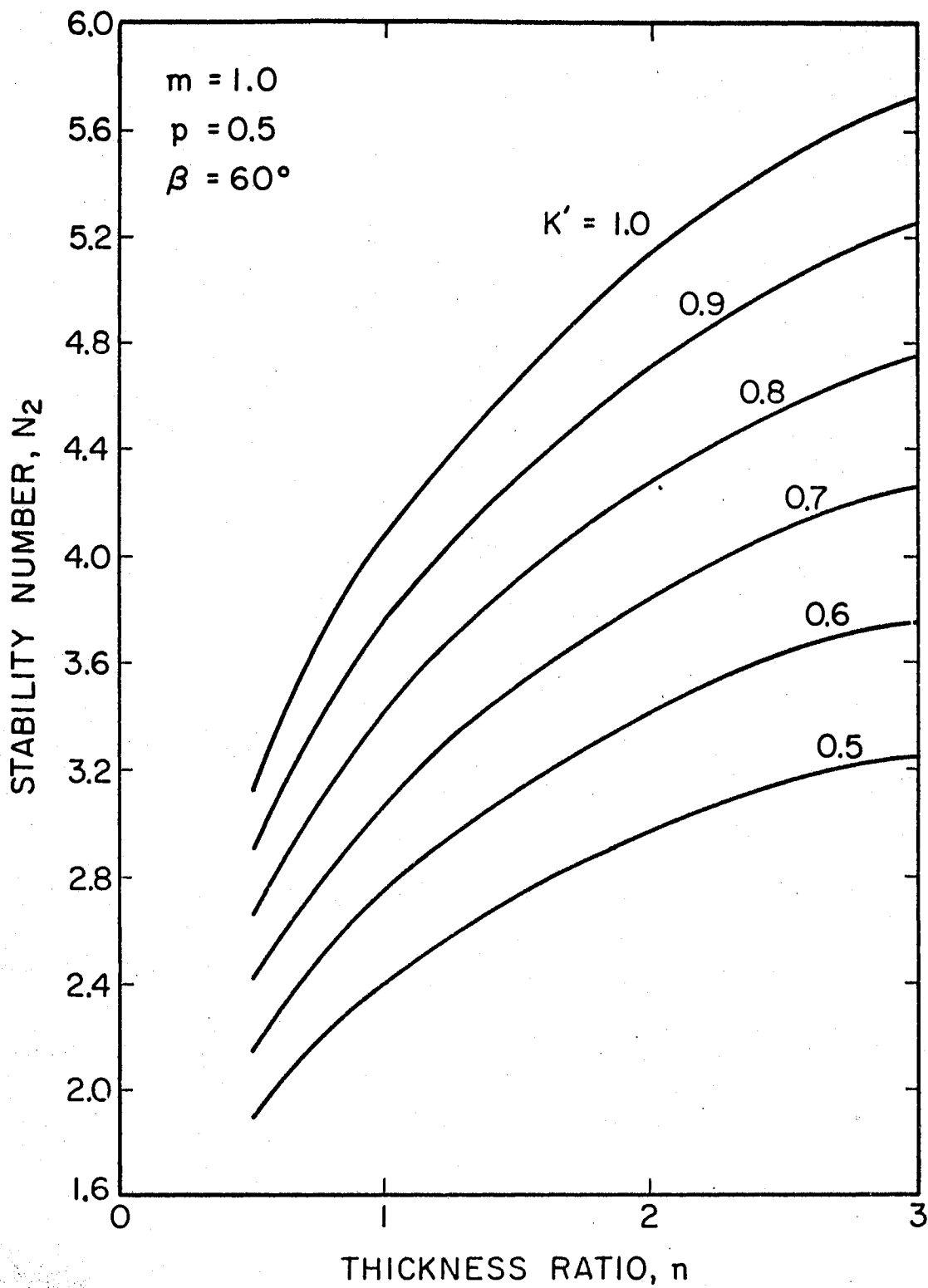


Figure 26. Thickness Ratio ( $n$ ) versus Stability Number ( $N_2$ ), ( $m = 1.0$ ,  $p = 0.5$ , and  $\beta = 60^\circ$ )

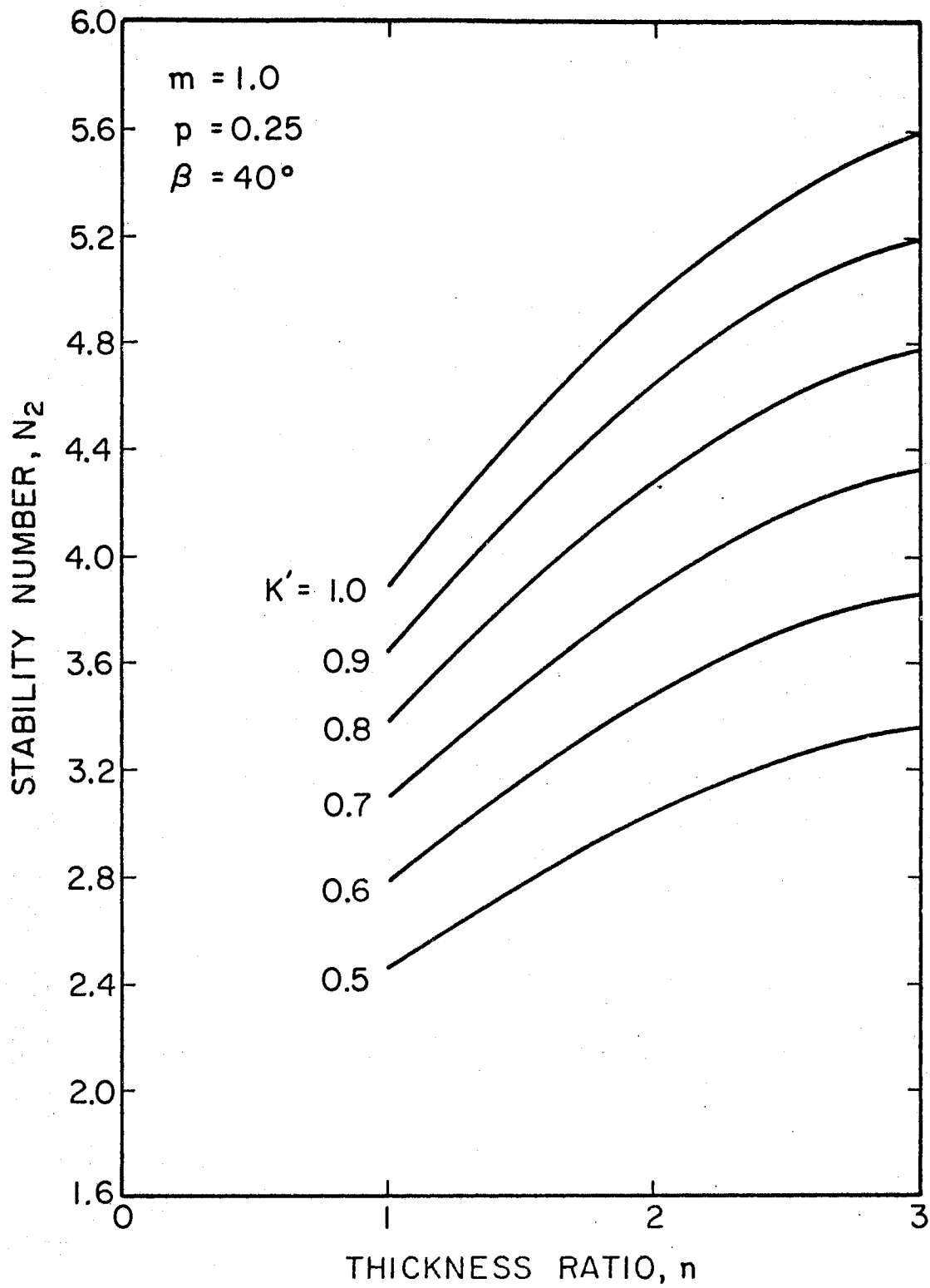


Figure 27. Thickness Ratio ( $n$ ) versus Stability Number ( $N_2$ ), ( $m = 1.0$ ,  $p = 0.25$ , and  $\beta = 40^\circ$ )

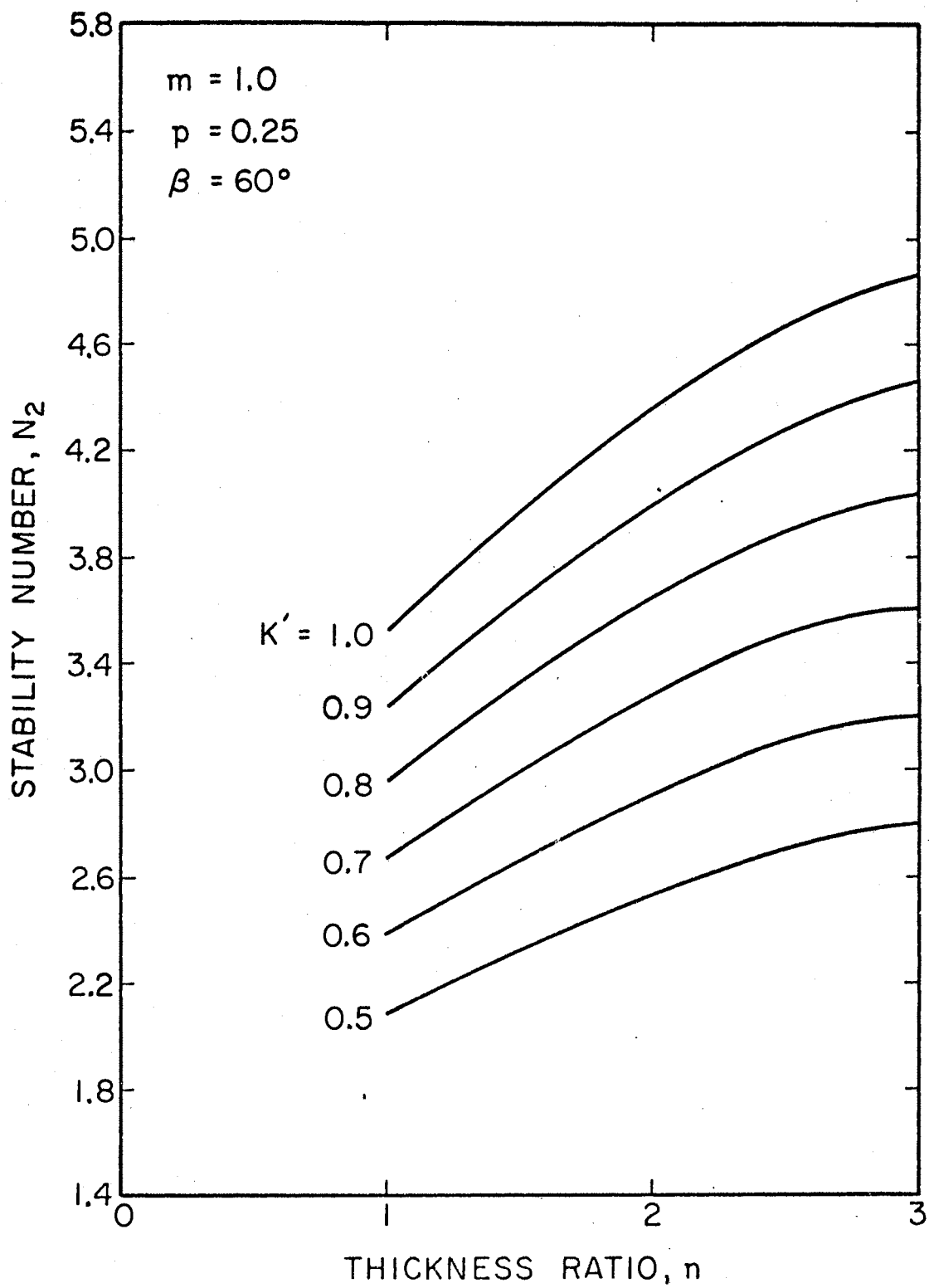


Figure 28. Thickness Ratio ( $n$ ) versus Stability Number ( $N_2$ ), ( $m = 1.0$ ,  $p = 0.25$ , and  $\beta = 60^\circ$ )

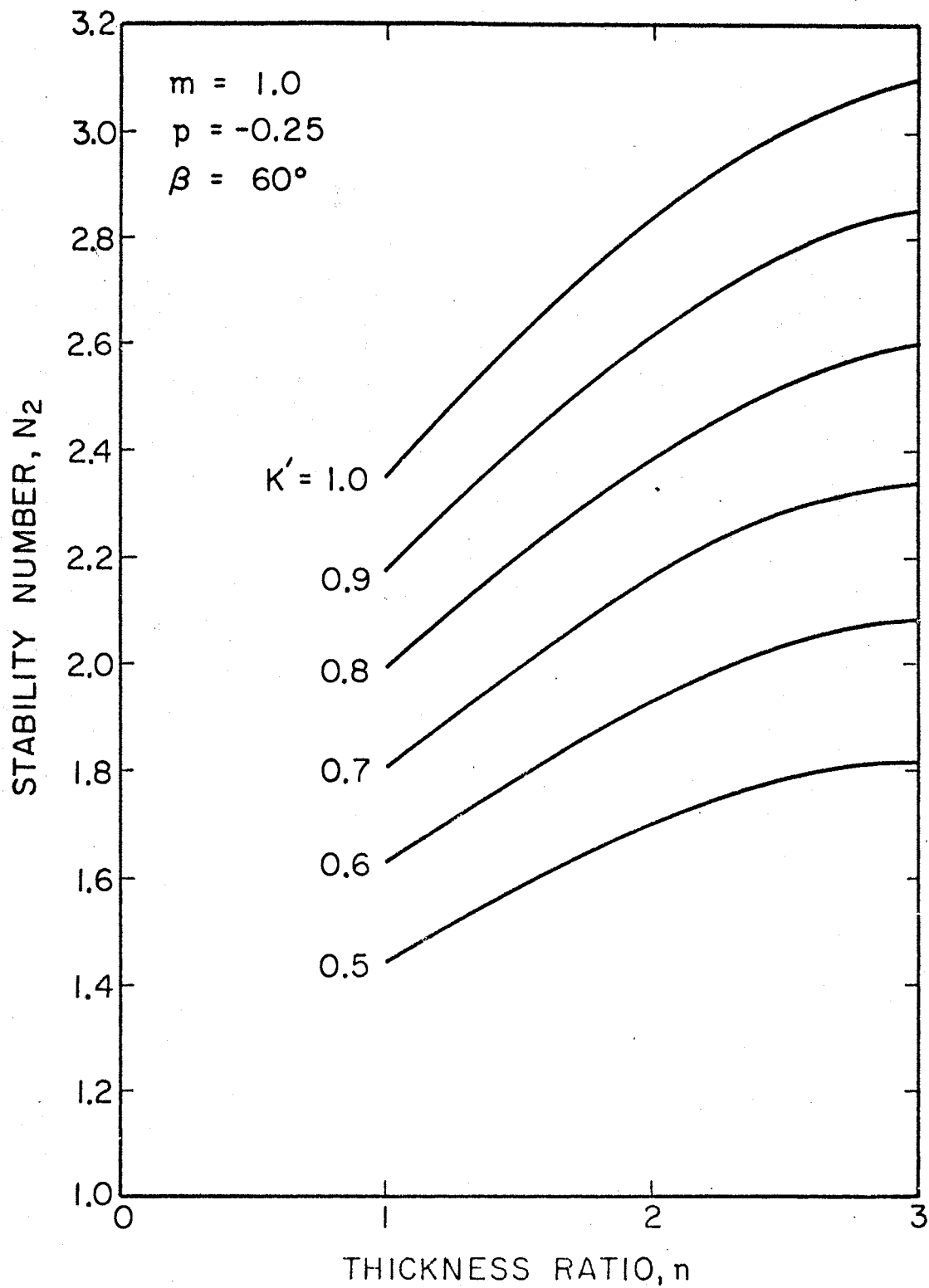


Figure 29. Thickness Ratio ( $n$ ) versus Stability Number ( $N_2$ ), ( $m = 1.0$ ,  $p = -0.25$ , and  $\beta = 60^\circ$ )



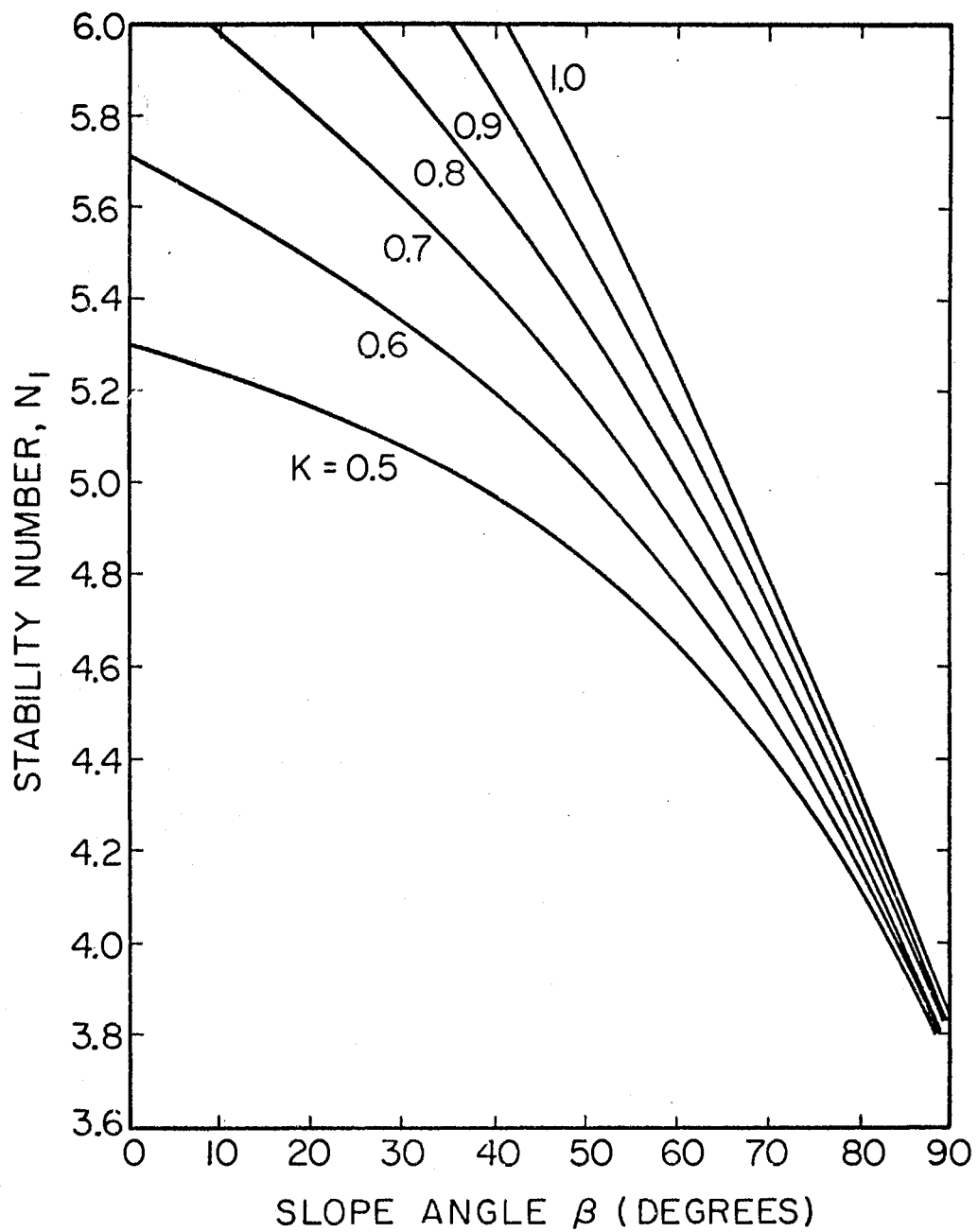


Figure 30. Slope Angle ( $\beta$ ) versus Stability Number ( $N_2$ ), Layer 1, (After Lo, 1965)

## CHAPTER IV

### DISCUSSION OF RESULTS AND CONCLUSIONS

While estimating the stability of slopes, it is often assumed that soil is homogeneous and isotropic. However, it is evident from the review of the published literature (Chapter II) that soil is rarely homogeneous and isotropic. Nonhomogeneity and anisotropy in natural soil deposits affect the stability of slopes in such deposits. There are two major kinds of deviation from the ideal homogeneous material. The first is the case in which the soil consists of layers of distinctly different soils (for example, layered clays), the second being a soil deposit which lacks any distinct stratification but whose properties vary from one point to another over a wide range.

In the present study, it is the former kind of nonhomogeneity that is studied. An analytical method for the evaluation of stability of earth slopes in a two-layered system of anisotropic soils is presented. The approach to this problem is based on the intuition that the overall stability of the slope is governed by the individual stability of the layers. Hence, the two layers are analyzed separately for their stability numbers (from which the factor of safety is obtained,  $F = \frac{C}{\gamma H} N$ ). The stability of a given slope is then dependent on the layer having the lower factor of safety.

It is logical to expect that the ratio of thickness of layers ( $n = H_2/H_1$ ), the anisotropy index ( $K = C_2/C_1$ ,  $K' = C'_2/C'_1$ ), and the

relative strength ( $p$ ) of the layers would influence the stability of the slope. To study these effects, numerical results are obtained for the following values of the above parameters:

$$n = 0.5, 1.0, 2.0, \text{ and } 3.0$$

$$K \text{ and } K' = 0.5, 0.6, 0.7, 0.8, 0.9, \text{ and } 1.0$$

$$p = -0.5, -0.25, 0.25, 0.5, \text{ and } 1.0.$$

Since there would not be much difference in the unit weights of soil in the two layers,  $m (\gamma_2/\gamma_1)$  is taken to be unity. All the numerical results which are graphically presented (Figures 7 through 29) were obtained with the aid of IBM 360/65 computer available at Oklahoma State University. The computer program used in obtaining the minimum stability number is listed in Appendix I. The analytical method suggested in this report is valid for slopes steeper than  $40^\circ$ .

The stability number  $N_2$  for the bottom layer is dependent on slope angle ( $\beta$ ), coefficient of anisotropy ( $K'$ ), coefficient of nonhomogeneity ( $m$ ), thickness ratio ( $n$ ), and relative strength index ( $p$ ). Charts (Figures 7 through 19) are presented to show slope angle ( $\beta$ ) versus stability number ( $N_2$ ) for various values of  $K'$ ,  $n$ , and  $p$ . For all these charts,  $m$  is taken to be unity.

The stability number  $N_1$  for the top layer is a function of slope angle ( $\beta$ ) and coefficient of anisotropy ( $K$ ). A chart (Figure 30) showing slope angle ( $\beta$ ) versus stability number ( $N_1$ ) is presented for various values of  $K$  (Lo, 1965).

To assess the influence of  $p$  (relative strength index) on the stability of the second layer, the stability number ( $N_2$ ) is plotted against  $p$  varying from  $-0.50$  to  $0.50$  in Figures 20 through 24. It is evident from these charts that the stability number increases linearly with  $p$ .

This is in accordance with the expectations that the stronger the stratum the more stable it will be.

Charts (Figures 25 through 29) demonstrate the influence of thickness ratio ( $n$ ) on stability number ( $N_2$ ). For a given slope ( $\beta$ ) and strength ( $p$ ), the stability number ( $N_2$ ) increases with  $n$ . Nevertheless, there is a trend indicating that  $N_2$  is less and less influenced with an increase in  $n$  from 0.5 to 3.0. For clarity in understanding this statement, four cases are tabulated below. Perhaps, for values of  $n$  greater than 3, its influence on  $N_2$  is negligible.

TABLE III  
INFLUENCE OF THICKNESS RATIO ( $n$ ) ON  
STABILITY NUMBER ( $N_2$ )

m	p	$\beta$	K'	percent increase in $N_2$ as $n$ increases from		
				0.5 to 1.0	1.0 to 2.0	2.0 to 3.0
1.0	0.5	60°	0.5	27.50	22.82	10.14
1.0	0.5	60°	1.0	30.24	25.92	11.40
1.0	0.5	40°	0.5	30.60	24.12	11.83
1.0	0.5	40°	1.0	38.85	29.95	12.37

### Conclusions

From the above study, the following conclusions may be drawn with regard to stability slopes in a layered system:

1. Charts (Figures 7 through 30) enable the analysis of earth slopes (slopes steeper than  $40^\circ$ ) in a two-layered system of anisotropic soils.

2. The overall stability of a slope is dependent on the individual stability of the layers.

3. The stability of the bottom layer is dependent on the thickness ratio ( $n$ ), the coefficient of anisotropy ( $K'$ ), the relative strength index ( $p$ ),  $m(\gamma_2/\gamma_1)$ , and slope angle ( $\beta$ ), whereas the stability of the top layer is a function of the coefficient of anisotropy ( $K$ ) and slope angle ( $\beta$ ).

4. The stability number ( $N_2$ ) of the bottom layer for a given slope increases linearly with the relative strength index ( $p$ ).

5. The influence of the thickness ratio ( $n$ ) on the stability number ( $N_2$ ) for a given slope reduces gradually as  $n$  increases from 0.5 to 3.0. Perhaps, for higher values ( $n > 3.0$ ), its influence on  $N_2$  is negligible. So, for thickness ratios greater than 3.0, the charts for  $n = 3.0$  could be used to analyze the stability of a given slope. These would give a conservative estimate of the factor of safety.

6. The method presented in this thesis assumes the following variation for shear strength:

$$C_i = C_h + (C_v - C_h) \cos^2 i,$$

where

$C_i$  = shear strength along a plane inclined at an angle,  $i$ , to the vertical, and

$C_h$  and  $C_v$  = shear strengths along the horizontal and vertical planes, respectively.

However, this method could be extended for any other assumed variation for shear strength.

### Recommendations for Further Research

During this study, some interesting topics were noted which merit further investigation. Some suggestions in this direction are listed in the following paragraphs.

1. It is not uncommon for a soils engineer to encounter  $C - \phi$  soils in nature. Therefore, it may be worthwhile to develop an analytical method to solve stability problems in such soils.

2. Pore pressure effects and earthquake effects have not been considered in the present work. It is suggested that a theoretical method could be developed taking these factors into account to assess their influence on stability of earth slopes in layered soils.

3. The application of the finite element method of analysis, which has been found to be versatile in solving problems in some areas of soil mechanics, could be studied to analyze slope stability problems in a layered system of nonhomogeneous and anisotropic soils.

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APPENDIX A  
HYPOTHETICAL PROBLEMS

1. Analyze the slope cut shown in Figure 31 for its stability. The cut is of 30' in height and is on a  $40^\circ$  slope in a layered system of non-homogeneous and anisotropic soils. The properties of the soil in the layers are as follows:

$$\text{Top Layer: } C_1 = 800 \text{ psf} \quad C_2 = 480 \text{ psf} \quad \gamma_1 = 120 \text{ pcf}$$

$$\text{Bottom Layer: } C'_1 = 600 \text{ psf} \quad C'_2 = 360 \text{ psf} \quad \gamma_2 = 120 \text{ pcf.}$$

Assume that the critical surface corresponds to a toe failure.

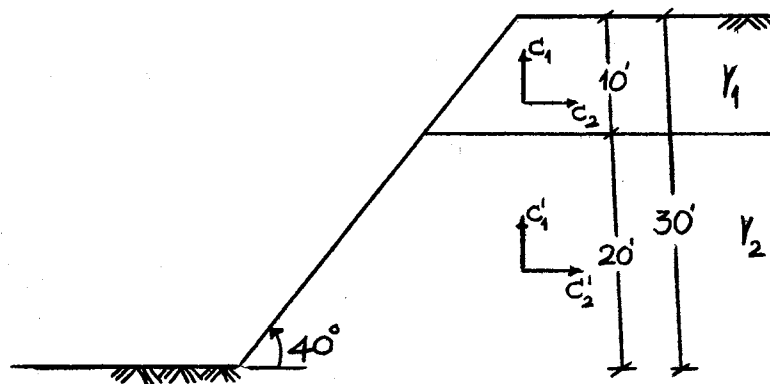


Figure 31. Slope Cut in a Layered System of Anisotropic Soils

$$\text{Data: } C_1 = 800 \text{ psf}, C_2 = 480 \text{ psf}, K = \frac{C_2}{C_1} = \frac{480}{800} = 0.6$$

$$C'_1 = 600 \text{ psf}, C'_2 = 360 \text{ psf}, K' = \frac{C'_2}{C'_1} = \frac{360}{600} = 0.6.$$

As per the approach followed in this thesis,

$$C'_1 = (p + 1) C_1$$

$$600 = (p + 1) 800$$

$$(p + 1) = \frac{600}{800}$$

$$p = 0.75 - 1.0$$

$$= -0.25.$$

$$m = \frac{\gamma_2}{\gamma_1} = \frac{120}{120} = 1.0$$

$$n = \frac{H_2}{H_1} = \frac{20}{10} = 2.0$$

$$\beta = 40^\circ$$

$$M = 1.0$$

$$n = 2.0$$

$$p = -0.25$$

$$K = K' = 0.6.$$

Layer 1:

$$\text{Stability Number } (N_1) = 5.198 \text{ (From Figure 30)}$$

$$\begin{aligned} \text{Factor of Safety} &= \frac{C_1}{\gamma_1 H_1} N_1 \\ &= \frac{800}{120 \times 10} \times 5.198 \\ &= 3.446. \end{aligned}$$

Layer 2:

$$\text{Stability Number } (N_2) = 2.26 \text{ (From Figure 15)}$$

$$\begin{aligned} \text{Factor of Safety} &= \frac{C_1}{\gamma_2 H_2} N_2 \\ &= \frac{800}{120 \times 20} \times 2.26 \\ &= 0.753. \end{aligned}$$

The stability of the cut is governed by the bottom layer (weaker), since the factor of safety for this layer is less than that for the top layer. In the present case, the cut is unstable, since the factor of safety is less than one.

2. An embankment 30 feet high (Figure 32) is made up of two soils,  $S_1$  and  $S_2$ , whose properties are given below. The soil  $S_1$  is used for constructing the lower 10 feet of the embankment, and soil  $S_2$  is used for the rest.

Soil  $S_1$ :  $C_1 = 500$  psf  $C_2 = 400$  psf  $\gamma_1 = 120$  pcf

Soil  $S_2$ :  $C'_1 = 1000$  psf  $C'_2 = 800$  psf  $\gamma_2 = 120$  pcf.

Analyze this embankment for its stability. Assume that the critical failure surface corresponds to toe failure.

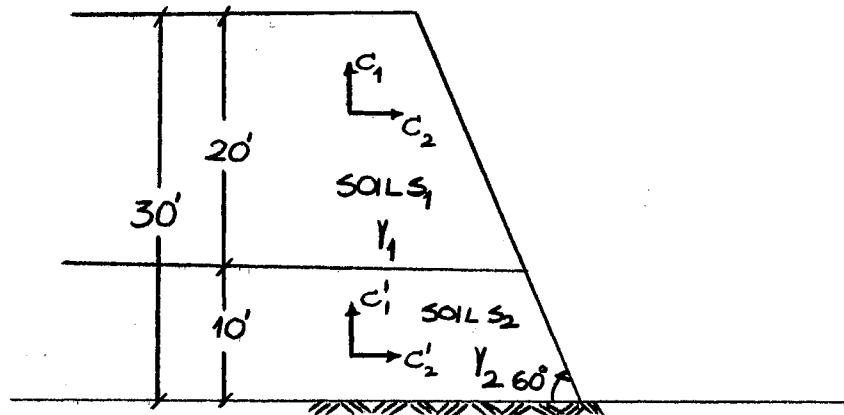


Figure 32. An Embankment with a Layered System of Anisotropic Soils

Data:

$$\text{Soil } S_1: C_1 = 500 \text{ psf}, C_2 = 400 \text{ psf}, K = \frac{C_2}{C_1} = \frac{400}{500} = 0.8$$

$$\text{Soil } S_2: C'_1 = 1000 \text{ psf}, C'_2 = 800 \text{ psf}, K' = \frac{C'_2}{C'_1} = \frac{800}{1000} = 0.8$$

As per the notation followed in this thesis,

$$C'_1 = (p + 1) C_1$$

$$1000 = (p + 1) 500$$

$$(p + 1) = 2.0$$

$$p = 1.0$$

$$m = \frac{\gamma_2}{\gamma_1} = \frac{120}{120} = 1.0$$

$$n = \frac{H_2}{H_1} = \frac{10}{20} = 0.5.$$

$$\beta = 60^{\circ} \qquad m = 1.0 \qquad n = 0.5$$

$$p = 1.0 \qquad K = K' = 0.8.$$

Top Layer:

$$\text{Stability Number} = 5.023 \text{ (From Figure 30)}$$

$$\begin{aligned} \text{Factor of Safety} &= \frac{C_1}{\gamma_1 H_1} N_1 \\ &= \frac{500}{120 \times 20} \times 5.023 \\ &= 1.0465. \end{aligned}$$

Bottom Layer:

$$\text{Stability Number} = 3.3415 \text{ (From Figure 19)}$$

$$\begin{aligned} \text{Factor of Safety} &= \frac{C_1}{\gamma_2 H_2} N_2 \\ &= \frac{500}{120 \times 10} \times 3.3415 \\ &= 1.3920. \end{aligned}$$

The stability of the embankment is controlled by the top layer (weaker), as the factor of safety for this is less than that for the bottom layer.

APPENDIX B  
COMPUTER PROGRAM



## 80/80 LIST

00000000111111112222222233333333444444445555555566666666777777778  
 1234567890123456789012345678901234567890123456789012345678901234567890

```

CARD
0001 C      OKLAHOMA STATE UNIVERSITY          D.DHAVALA
0002 C
0003 C
0004 C      STABILITY ANALYSIS OF SLOPES IN A TWO-LAYER SYSTEM OF
0005 C      ANISOTROPIC SOILS
0006 C
0007 C
0008 C      DESCRIPTION OF PARAMETERS
0009 C
0010 C
0011 C      B          =SLOPE ANGLE
0012 C      M          =COEFFICIENT OF NON-HOMOGENEITY
0013 C      T          =THICKNESS RATIO
0014 C      P          =RELATIVE STRENGTH INDEX
0015 C      RP         =COEFFICIENT OF ANISOTROPY
0016 C      SN2       =STABILITY NUMBER-LAYER 2
0017 C      AP        =GEOMETRICAL PARAMETER DEFINING CRITICAL SLIP SURFACE.
0018 C              ALPHA PRIME
0019 C      BP        =GEOMETRICAL PARAMETER DEFINING CRITICAL SLIP SURFACE.
0020 C              LAMDA TWO
0021 C      XL(1)     =LOWER LIMIT ON AP
0022 C      XL(2)     =LOWER LIMIT ON BP
0023 C      XR(1)     =UPPER LIMIT ON AP
0024 C      XR(2)     =UPPER LIMIT ON BP
0025 C
0026 C
0027 C      SUBROUTINES REQUIRED
0028 C
0029 C      PATRN, EXPLOR, AND MERIT
0030 C
0031 C
0032 C      IMPLICIT REAL*8 (A-H,O-Z)
0033 C      COMMON /MNMERT/ AA,BB,B,R,RP,SN?
0034 C      DIMENSION X(9),XL(9),XR(9)
0035 C      N=2
0036 C      NP=2
0037 C      DELTA=0.01
0038 C      ROW=0.5
0039 C      F=0.001
0040 C      Q=3.141592600/18.000
0041 C      200 CONTINUE
0042 C      READ 201,RP
0043 C      201 FORMAT(1E10.3)
0044 C      IVAL=10.0*RP
0045 C      IF (IVAL .EQ. 0) GO TO 999
0046 C      DO 25 I=5,7
0047 C      B=1*Q
0048 C      XL(1)=0.1745
0049 C      XL(2)=0.1745
0050 C      XR(1)=0.8855
0051 C      XR(2)=0.5655
0052 C      X(1)=0.5236
0053 C      X(2)=0.4363
0054 C      CALL PATRN(N,NP,DELTA,F,XL,XR,Y,X,ROW,NN)

```

## 80/80 LIST

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```

CARD
0055      CALL MERIT(X,Y)
0056      WRITE(6,3) AA,BB,B,R,RP,SN2
0057      3 FORMAT(/,1X,'AA=',F15.8,/,1X,
0058          1'BB=',F15.8,/,1X,
0059          1'B=',F15.8,/,1X,
0060          1'R=',F11.4,/,1X,
0061          1'RP=',F11.4,/,1X,
0062          1'SN2=',F11.4,/)
0063      25 CONTINUE
0064          GO TO 200
0065      999 STOP
0066          END
0067
0068
0069
0070
0071
0072
0073      SUBROUTINE PATRN(N,NP,DELTA,F,XL,XR,Y,X,ROW,NN)
0074      IMPLICIT REAL*8 (A-H,O-Z)
0075      C      PATTERN SEARCH FOR MULTIVARIABLES
0076      C      THIS SUBROUTINE CONDUCTS A PATTERN SEARCH
0077      C      WITHIN REGIONAL CONSTRAINTS IN A HIPERSURFACE
0078      C      OF UPTO NINE INDEPENDENT VARIABLES.
0079      C
0080      C      CALLING PROGRAM REQUIREMENTS.
0081      C
0082      C      PROVIDE A SUBROUTINE MERIT FROMWHICH AN ORDINATE Y IS RETURNED
0083      C      WHEN COLUMN VECTOR ABSCISSA X IS RETURNED.
0084      C
0085      C      VARIABLES.
0086      C      N=NUMBER OF INDEPENDENT VARIABLES.
0087      C      NP=CONVERGENCE MONITOR.
0088      C      NP=C WILL NOT PRINT.
0089      C      NP 1 WILL PRINT EVERY ITERATION.
0090      C      NP 2 WILL PRINT EVERY 2ND ITERATION.
0091      C      DELTA =CURRENT STEP SIZE.
0092      C      F=MINIMUM STEP SIZE
0093      C      XL=LOWER BOUND OF SEARCH DOMAIN
0094      C      XR=HIGHER BOUND OF SEARCH DCMAIN
0095      C      Y=FUCTIONAL VALUE RESULTING FROM CURRENT MOVE
0096      C      YY=FUNCTIONAL VALUE AT BASE POINT
0097      C      YYY=FUNCTIONAL VALUE AT CURRENT BASE POINT
0098      C      XXX=PREVIOUS BASE POINT
0099      C      XX=BASE POINT RESULTING FROM CURRENT MOVE
0100      C      DIMENSION X(9),XX(9),XXX(9),XL(9),XR(9),NG(9)
0101      C      NF=0
0102      C      N1=0
0103      C      N2=0
0104      C      NN=0
0105      C      DELTA1=DELTA
0106      C      DO 45 I=1,N
0107      45  NG(I)=1
0108      C      IF (NP)5,5,6

```

## 80/80 LIST

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```

CARD
0109      6  WRITE(6,7)(NG(I),I=1,N)
0110      7  FORMAT(2X,'NN',6X,'DELTA',9X,'Y',2X,9(7X,'(',12,')'))
0111      5  CALL MERIT(X,Y)
0112          NF=NF+1
0113          NN=NN+1
0114          IF(NP)31,31,32
0115      32  WRITE(6,33)NN,DELTA,Y,(X(I),I=1,N)
0116      33  FORMAT(1X,14,9(2X,E11.4))
0117      31  CONTINUE
0118  C     STRAT AT BASE POINT
0119      1  YY=Y
0120          DO 10 K=1,N
0121          XX(K)=X(K)
0122      10  CONTINUE
0123  C     MAKE EXPLORATORY MOVES
0124          CALL EXPLOR(N,XX,YY,XL,XR,DELTA,ROW,NF)
0125  C     IS PRESENT FUNCTIONAL VALUE BELOW THAT AT BASE POINT?
0126  C
0127          IF(YY - Y)3,3,2
0128  C     SET NEW BASE POINT
0129      2  DO 12 K=1,N
0130          XXX(K)=X(K)
0131          X(K)=XX(K)
0132      12  CONTINUE
0133          Y=YY
0134  C     MAKE PATTERN MOVE
0135          DO 14 K=1,N
0136          XX(K)=2.0*XX(K)-XXX(K)
0137      14  CONTINUE
0138  C     CHECK IF CONSTRAINT IS VIOLATED
0139          DO 20 I=1,N
0140          IF(XX(I)-XL(I))41,42,42
0141      41  XX(I)=XL(I)
0142      42  IF(XX(I)-XR(I))20,20,44
0143      44  XX(I)=XR(I)
0144      20  CONTINUE
0145          CALL MERIT(XX,YYY)
0146          NF=NF+1
0147          NN=NN+1
0148          YY=YYY
0149          IF (N2)21,22,21
0150      21  N2=N2+1
0151          IF(N2-NP)22,23,22
0152      22  WRITE(6,33)NN,DELTA,Y,(X(I),I=1,N)
0153          N2=0
0154      23  CALL EXPLOR(N,XX,YY,XL,XR,DELTA,ROW,NF)
0155          IF(YY-Y)1,1,2
0156      3  DDELTA=DELTA-F
0157          IF (ODELTA)13,15,15
0158  C     DECREASE STEP SIZE
0159      15  DELTA=ROW*DELTA
0160          GO TO 1
0161      13  WRITE(6,100)NF,Y,NN,DELTA1,DELTA,ROW,F
0162      100  FORMAT(/,1X,'TOTAL NUMBER OF FUNCTION EVALUATIONS.....',15,/,1X,

```

## 80780 LTST

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```

CARD
0163      1*LARGEST MERIT ORDINATE.....',F15.8,/,1X,
0164      1*NUMBER OF BASE      EVALUATIONS.....',115,/,1X,
0165      3*ORIGINAL STEP SIZE.....',F15.8,/,1X,
0166      4*FINAL STEP SIZE.....',F15.8,/,1X,
0167      5*REDUCTION FACTOR FOR STEP SIZE.....',F15.8,/,1X,
0168      6*FRACTIONAL REDUCTION OF UNCERTAINTY.....',F15.8,/)
0169      RETURN
0170      END
0171
0172 C
0173      SUBROUTINE EXPLOR(N,XX,YY,XL,XR,DELTA,ROW,NF)
0174      IMPLICIT REAL*8 (A-H,O-Z)
0175      DIMENSION XX(9),XL(9),XR(9)
0176      DO 10 K=1,N
0177 C      INCREASE ORDINATE, CALCULATE ORDINATE
0178      XX(K)=XX(K)+DELTA
0179 C      CHECK IF CONSTRAINT IS VIOLATED
0180      IF(XX(K)-XL(K))21,22,22
0181      21 XX(K)=XL(K)
0182      22 IF(XX(K)-XR(K))24,24,23
0183      23 XX(K)=XR(K)
0184      24 CONTINUE
0185      CALL MERIT(XX,YYY)
0186      NF=NF+1
0187 C      IS MOVE A SUCCESS ?
0188      IF(YYY-YY)1,1,2
0189 C      RETAIN NEW CO ORDINATE AND NEW FUNCTIONAL VALUE
0190      2 YY=YYY
0191      GO TO 10
0192 C      DECREASE ORDINATE,CALCULATE NEW ORDINATE
0193      1 XX(K)=XX(K)-2.0*DELTA
0194      IF(XX(K)-XL(K))25,26,26
0195      25 XX(K)=XL(K)
0196      26 IF(XX(K)-XR(K))28,28,27
0197      27 XX(K)=XR(K)
0198      28 CONTINUE
0199      CALL MERIT(XX,YYY)
0200      NF=NF+1
0201 C      IS MOVE A SUCCESS ?
0202      IF(YYY-YY)3,3,4
0203 C      RETURN CO-ORDINATE & NEW FUNCTIONAL VALUE
0204      4 YY=YYY
0205      GO TO 10
0206 C      RESET CO-ORDINATE
0207      3 XX(K)=XX(K)+DELTA
0208      10 CONTINUE
0209      RETURN
0210      END
0211
0212
0213
0214
0215
0216

```

## 80/80 LIST

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```

CARD
0217      SUBROUTINE MERIT(X,Y)
0218      IMPLICIT REAL*8 (A-H,O-Z)
0219      COMMON /MMERT/ AA,BB,R,R,RP,SN2
0220      REAL*8 DSIN,DCOS,M,N,AA,BB
0221      DIMENSION X(9),F(4)
0222      C    AP=ALPHA PRIME
0223      C    BP=LAMDA TWO
0224      AP=X(1)
0225      BP=X(2)
0226      M=1.000
0227      T=3.000
0228      P=-0.2500
0229      SAP=DSIN(AP)
0230      CAP=DCOS(AP)
0231      C2AP=DCOS(AP+AP)
0232      COTAP=CAP/SAP
0233      SBP=DSIN(BP)
0234      CBP=DCOS(BP)
0235      COTBP=CBP/SBP
0236      COTBP2=COTBP*COTBP
0237      COTBP3=COTBP2*COTBP
0238      SB=DSIN(B)
0239      CB=DCOS(B)
0240      TANB=SB/CB
0241      COTB=CB/SB
0242      COTB2=COTB*COTB
0243      COTB3=COTB2*COTB
0244      S2AP=DSIN(AP+AP)
0245      SAP2=SAP*SAP
0246      CSAP2=1.000/SAP2
0247      SBP2=SBP*SBP
0248      CSBP2=1.000/SBP2
0249      Y1=2.000*(AP-BP)
0250      CY1=DCOS(Y1)
0251      SY1=DSIN(Y1)
0252      R=COTBP*TANB
0253      IF(R.GT.(4.0/3.0)) GO TO 50
0254      AA=0.2500*(1.000+RP)*(1.000+P)*AP*CSAP2*CSBP2-0.12500*(1.000-RP)*
0255      11.000+P)*CSAP2*CSBP2*S2AP*CY1+0.500*(COTBP2*TANB+COTAP*COTBP*TANB-
0256      ICOTBP-COTAP)
0257      BB=(COTAP*COTBP2*TANB/(4.000*M))+(COTBP3*TANB/(12.000*M))-(COTBP2/
0258      112.000)+((M-2.000)/(4.000*M))*COTAP*COTBP+((M-1.000)/(4.000*M))*CO
0259      ITRP*COTB+((1.000-M)/(4.000*M))*COTB*COTAP+((1.000-M)/(6.000*M))*CO
0260      ITB2+(1.000/12.000)*CSBP2
0261      F(1)=(0.2500*(1.000+RP)*(1.000+P)*CSBP2-0.500*(1.000+RP)*(1.000+P)
0262      1*AP*COTAP*CSBP2+0.2500*(1.000-RP)*(1.000+P)*COTAP*CSBP2*S2AP*CY1-0
0263      1.2500*(1.000-RP)*(1.000+P)*CSBP2*C2AP*CY1+0.2500*(1.000-RP)*(1.000
0264      +P)*CSBP2*S2AP*SY1+0.500*(1.000-COTBP*TANB))*BB+(0.2500*COTBP2*TAN
0265      B/T+(M-2.000)*COTBP/(4.000*M)+(1.000-M)*COTB/(4.000*M))*AA
0266      F(2)=(-0.500*(1.000+RP)*(1.000+P)*AP*CSAP2*COTBP+0.2500*(1.000-RP)
0267      1*(1.000+P)*CSAP2*COTBP*S2AP*CY1-0.2500*(1.000-RP)*(1.000+P)*CSAP2*
0268      S2AP*SY1+0.500*(-2.000*COTBP*TANB-COTAP*TANB+1.000))*BB+(0.500*COT
0269      1AP*COTRP+TANB/M+0.2500*COTBP2*TANB/M+(M-2.000)*COTAP/(4.000*M)+(M-
0270      11.000)*COTB/(4.000*M))*AA

```

## 80/80 LIST

000000000111111112222222223333333333444444444555555556666666667777777778  
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CARD  
 0271 SN2=AA/BB  
 0272 Y=F(1)\*F(1)+F(2)\*F(2)  
 0273 Y=-Y  
 0274 RETURN  
 0275 50 AA=0.25D0\*(1.0D0+RP)\*(P+1.0D0)\*AP\*CSAP2\*CSBP2-0.125D0\*(1.0D0-RP)\*(  
 0276 11.0D0+P)\*CSAP2\*CSBP2\*S2AP\*CY1+(COTAP+COTBP)/(2.0D0\*T)  
 0277 BB=(0.5D0/(M\*T)+0.25D0/(M\*T\*T)+0.25D0\*(COTBP-COTAP)\*COTB-(0.5D0/(  
 0278 1M\*T)+1.0D0/(M\*T\*T)+1.0D0/(6.0D0\*M\*T\*T)+1.0D0/6.0D0)\*COTB2+(0.5D0  
 0279 1/(M\*T\*T))\*COTB3+1.0D0/12.0D0+(0.5D0/(M\*T)+0.25D0)\*COTBP\*COTAP  
 0280 F(1)=(0.25D0\*(1.0D0+RP)\*(1.0D0+P)\*CSBP2-0.5D0\*(1.0D0+RP)\*(1.0D0+P)  
 0281 1\*AP\*COTAP\*CSBP2-0.25D0\*(1.0D0-RP)\*(1.0D0+P)\*C2AP\*CY1\*CSBP2+0.25D0\*  
 0282 1\*(1.0D0-RP)\*(1.0D0+P)\*S2AP\*SY1\*CSBP2+0.25D0\*(1.0D0-RP)\*(1.0D0+P)\*S2  
 0283 1AP\*CY1\*COTAP\*CSBP2-0.5D0/T)\*BB+AA\*((0.25D0+0.5D0/(M\*T))\*COTBP-(0.5  
 0284 1D0/(M\*T)+0.25D0/(M\*T\*T)+0.25D0)\*COTB)  
 0285 F(2)=(-0.5D0\*(1.0D0+RP)\*(1.0D0+P)\*AP\*CSAP2\*COTBP-0.25D0\*(1.0D0-RP)  
 0286 1\*(1.0D0+P)\*S2AP\*SY1\*CSAP2+0.25D0\*(1.0D0-RP)\*(1.0D0+P)\*S2AP\*CY1\*CSA  
 0287 1P2\*COTBP-0.5D0/T)\*BB+((0.5D0/(M\*T)+0.25D0)\*COTAP+(0.5D0/(M\*T)+0.25  
 0288 1D0/(M\*T)+0.25D0)\*COTB)\*AA  
 0289 SN2=AA/BB  
 0290 Y=F(1)\*F(1)+F(2)\*F(2)  
 0291 Y=-Y  
 0292 RETURN  
 0293 END

VITA

Dakshniamurty Dhavala

Candidate for the Degree of

Doctor of Philosophy

Thesis: STABILITY OF SLOPES IN A TWO-LAYER SYSTEM OF ANISOTROPIC SOILS

Major Field: Civil Engineering

Biographical:

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