

## **Stability and sustainability assessment for the surface failure of various rock slopes excavated along the Konya-Alanya Road segment, Turkey**

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### **Abstract**

Most of the road networks in Turkey pass through the Hilly region. These roads are constructed through inadequate blasting accompanied by road deterioration or collapse. To date, several researchers in Turkey have not paid adequate attention to the stability analysis of road failures based on the surface orientation of the dominant discontinuity sets. This study, based on the field survey, laboratory measurements, and the use of Dips analyst software, aims to investigate the stability and sustainability of seven different sites that exhibit imminent slope failure along the Konya-Alanya Road (KAR) segment. The sites selected are geologically investigated and geotechnically evaluated using a scan-line survey. We carried out several fields and laboratory measurements. Both Slope Mass Rating (SMR) and Rock Mass Rating (RMR) are quantified and included with rock mass assessment of each slope site. Both the field and laboratory results are integrated by kinematic analysis methods to assess the potential failure of these slopes. The kinematic analysis results demonstrate that the dominant failure forms are planar, wedge, and toppling. RMR<sub>b</sub> results show that some slopes with good rock quality remain unstable and vulnerable to failure despite their fair RMR<sub>b</sub> values. While SMR results show that five of the seven sites are risky, thus, described as partly stable, and the other two are unstable. The results have also implications for the understanding of the causal factors for slope instability, include the discontinuities present in the rock mass, physical, environmental, and meteorological factors influencing them. This study concludes that urgent remedial measures for their long-term stability are recommended.

**Keywords:** Discontinuity; kinematic analysis; slope stability; SMR.

### **1. Introduction**

For safe, sustainable road construction and even in post-construction, slope stability analysis of road cuts is still a crucial stage (Hack, 2018). Slope instability particularly in the hilly terrains is often created from unplanned rock slope excavations for construction, infrastructure development, or

during the expansion of road networks through blasting, regardless of many significant geological and structural factors like discontinuities (Pradhan & Siddique, 2020). The interplay of these factors, and weather events (excessive rainfall and snowmelt), are the primary causes of road

deterioration and many slopes instability issues (Chen *et al.*, 2020).

The major causes of slope instability in the mountainous region generally can be summarized into three causes:

- i) Geological and environmental causes (Feng *et al.*, 2020; Othman *et al.*, 2019).
- ii) Dynamic and morphological causes (Hamza & Raghuvanshi, 2017).
- iii) Anthropogenic activities cause, such as excavation of slope, blasting, land use, etc. (Hack, 2018).

Blasting-triggered excavation can also increase its destabilization for cut slopes' long-term stability.

Slope failures can be classified into three or four forms of failure mechanisms (planar, wedge, toppling, and circular failures) (Hoek & Bray, 1981; Wyllie, 2017). In many cases, instability in rock slopes and rock mass develops through discontinuities, and these instabilities occur along one or two intersecting planes of discontinuity then forms a wedge type of failure (Hack *et al.*, 2003).

The Taurus Mountains are well known for their geomorphological complexity. It has a landscape resulting from tectonic stress in different directions and intensities (Özgül, 1997). In the Taurus Mountains, rock-slope instability is among the most dangerous environmental threats along the hillsides of the cutting road. For instance, in the present

study, the segment of the KAR gives a better connection to the residential area and makes it possible to connect agricultural centers and primary residential areas. However, due to frequent slope instability problems, this road was soon blocked for public use. These problems are mostly due to many causes. Slope instability problems around the study area have generally been studied using GIS, remote sensing, and laser techniques by many researchers (Üstün *et al.*, 2015), but, they have not paid much attention to the geo-structural causes and other causative factors in their investigations. This research comes to:

- i) investigating the instability of slope at seven sites excavated along the road section.
- ii) identifying different failure modes in the rock slopes.
- iii) pinpoints the causative factors for the failures in these slope sites.

## **2. Study area description**

The investigated slope sites along the road segment locate within Taurus Hilly's topographic areas. It begins from Taşkent district (128 km away from Konya city) up to the border of Alanya situated south of Konya city (Figure 1).

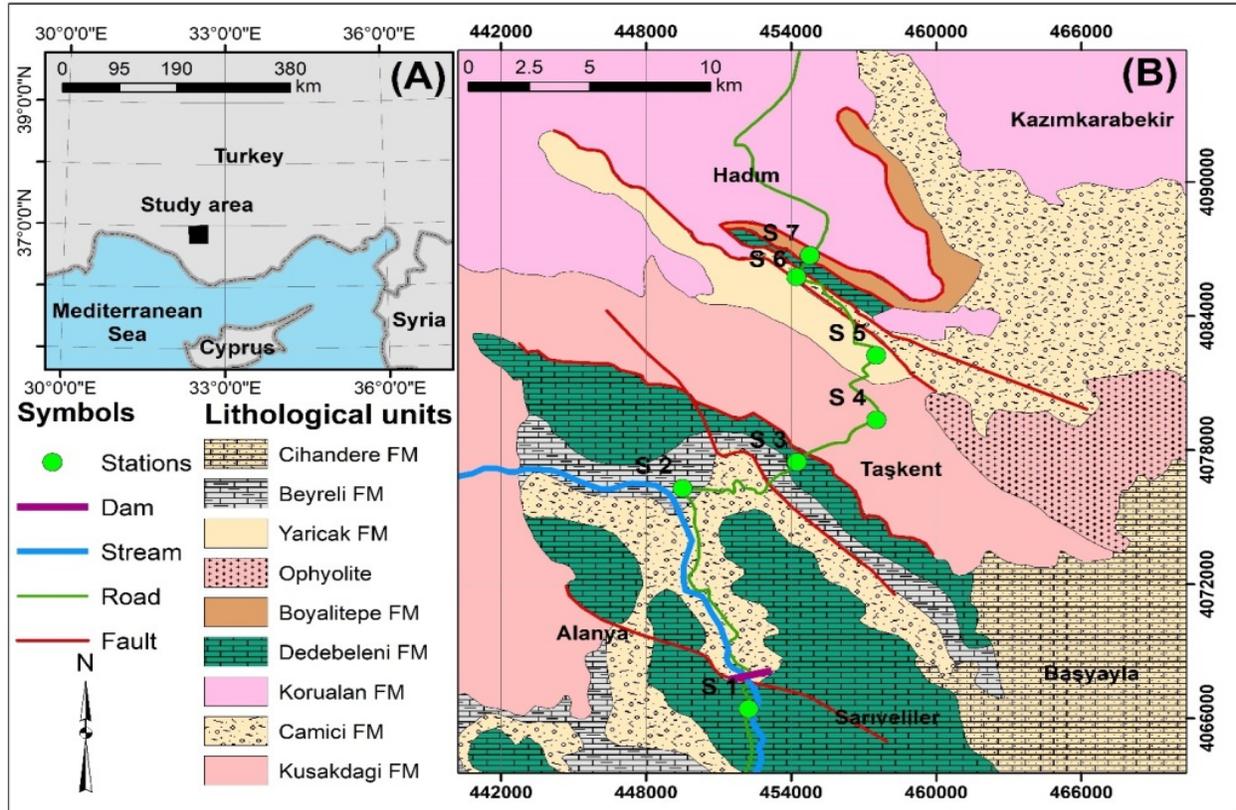


Fig. 1. A) location map, B) geological map of the study area (MTA, 2019).

These slopes are moderate- highly weathered, characterized by narrow valleys with high relief (between 1052 m and 1403 m), irregular and steep slopes with dip angles ranging between  $50^\circ$  and  $68^\circ$ , while its average dip direction is  $124^\circ$  and contain diverse shear zones, and largely man-made rock cut slopes that have been excavated with inadequate blasting, so these slopes pretend to be disturbed and deteriorated. However, these slopes, are recognized at the site as temporarily stabilized.

The annual average rainfall in the road region and around is between 648 mm and 732 mm (OGM, 2019).

Geologically, the dominant lithology of the cut slopes investigated along the road and around are mainly constitutes of jointed limestone alternations represented by the

Yarıcak (Carboniferous), Kuşakdağı (Permian), Beyreli (Middle Triassic), Dedebeleni (Late Jurassic) formations (Turan, 1990) (Figure 1). Bozkır Unit is also exposed in two sites (S-6 and S-7) represent a diverse rock ranged in age from Triassic to Late Cretaceous (Turan, 1990; Özgül, 1997). It contains a highly deformed pelagic and neritic limestone. It differs in appearance and intensity from one location to another and appears in sections repeatedly.

### 3. Methodology and data collection

For this study, both quantitative and qualitative methods are adopted in the data collection and analysis. In quantitative research, an extensive field survey was performed on recording the important

geometric, geo-structural features of the rock mass slope. The qualitative approach includes descriptive data based on field-measured observations and data (dependent variables), such as slope face length, slope height, slope angle, slope width, and slope dip/ dip directions.

### 3.1. Site investigations and field-scale tests

Site investigation and integrated geophysical investigation play a significant role in stability, designing, economic considerations, and even road preservation (Hack *et al.*, 2003; Adesola *et al.*, 2017; Wyllie, 2017). Field experiments are the first step and are important to understand the characteristics of the landslides (Özfirat *et al.*, 2017; Ahmed & Al-Dousari, 2013). In site investigations of this study, data on the seven different sites (S1-S7) in the cut slopes are collected through a detailed field survey. In discontinuity survey, joints properties that include orientation, persistence, aperture, spacing, roughness, infilling material properties, and fault attitudes are investigated at seven scan lines of different sites. A total of 212 measurements of joints are recorded in this survey. Other field-scale measurements carried out over the joint surface on-site are for the compressive strength of joint wall (JCS), and the coefficient of joint roughness (JRC). The wall strength has been directly estimated on the joint surfaces using the Schmidt hammer. The (JRC) is also determined by applying the Barton-type wire profilometer to the joint surfaces directly, with the regular profiles suggested by Barton & Choubey (1977). The related formula for these two important variables along with other shear strength parameters is given in the equation (Eq. 1).

$$\tau = \sigma_n \tan \left[ \phi_b + \text{JRC} \log_{10} \left( \frac{\text{JCS}}{\sigma_n} \right) \right] \dots\dots (1)$$

where  $\tau$  is the shear strength;  $\sigma_n$  is the normal (vertical) stress;  $\phi_b$  is the base friction angle of the planes; JRC is the Joint roughness coefficient, and JCS is referring to joint compressive strength. Finally, representative rock blocks have been collected from the seven slopes sites to carry out laboratory testing.

### 3.2. Laboratory testing

To identify the geotechnical properties of rocks, several laboratory-scales experiments (unit weight, compressive strength ( $\sigma_c$ ), and shear strength parameters ( $c$ ,  $\phi$ ) of rock discontinuity) are undertaken. For these tests, cores are prepared with their appropriate dimensions in the laboratory by I.S.R.M (2007) requirements. The ends of the specimens are sawed as well. Eighteen samples with a diameter of 54 mm are prepared, measured, then checked with a compass. The averages of samples diameters (D) and lengths (L) have been calculated. Moreover, friction angle value along joint surfaces was evaluated (Hencher & Richards, 2015).

### 3.3. Slope mass rating (SMR) approach.

SMR rating approach was adopted for this study. It is mainly obtained based on a simple rock mass rating (RMR<sub>b</sub>; Bieniawski, 1973,1989). RMR<sub>b</sub> is computed according to Bieniawski's (1973) and defined by adding rating values for five parameters are i) Strength of intact rock, ii) Rock Quality Designation (RQD) (measured or estimated),

iii) Spacing of discontinuities, iv) Discontinuities conditions, and v) Water inflow through discontinuities. RMR has a total range of 0 – 100. The SMR rating is estimated mathematically by Eq. 2 (Romana *et al.*, 2003).

$$SMR = RMR_b + (F1 * F2 * F3) + F4..(2)$$

where RMR<sub>b</sub>: basic RMR index. SMR is defined by the five parameters of the RMR<sub>b</sub> rating (Bieniawski, 1989). F1: relates to the parallelism between the dip orientation of joint and slope; F2: defines depending on joint dip within the planar/toppling failure modes; F3: appraises according to the relation between the dips amounts for each the slope and joint; and F4: is a factor relating to the excavation process.

Eq.2 was used to estimate the final SMR rating values in this study. After that, each of the adjustment factors and the RMR<sub>b</sub> rating's basic parameters is determined separately. The investigated slope's locations are described and categorized into different classes of instability with their potential risk degrees based on the anticipated percent possibility of slope stability.

### 3.4. Kinematic analysis approach

The stability analysis with the kinematic method used in this study was usually performed with a lower hemispheric projection of Stereo-net and identified according to instructions given by Hoek and Bray (1981), Goodman (1980), and Wyllie (2017).

The method utilized is mostly based on data acquired from a survey with a scan line, as well as some laboratory testing. Slope-joint orientations, slope geometry, and friction angle along discontinuity planes were among the information collected. In this methodology, at first, poles of the collected slope data are clustered by Stereo-net using DIPS 7.0 analyst software (Rocscience, 2010).

Second, based on the clustered data on the slope, the spatial distribution of all major discontinuity planes is presented. Then, the slope stability analysis is carried out accordingly.

## 4. Results

A total of 212 joints have been documented at seven different locations. Table 1 summarized the prevailing joint orientations data for these locations.

**Table 1.** Joint-slope data measured on slope faces and used as inputs to the kinematic analyzes.

* d-d = dip direct	Slope (d-d/da)	Bedding plane (d-d/da)	Orientation for joints			Residual ( $\phi_r$ ) (°)
			J1 (d-d / da)	J2 (d-d /da)	J3 (d-d / da)	
S-1	100/60	210/34-35	290/47	-	-	29
S-2	300/56	180/30-35	282/10	092/32	196/27	29
S-3	090/57	046/61	224/34	072/33	175/ 24	31
S-4	280/58	270/40	217/35	06/34	167/30	36
S-5	087/63	-	286/06	222/38	165/27	31
S-6	270/48	310/20-40	-	094/04	190/23	37
S-7	185/78	230/40	278/12	-	189/27	30

These data are examined in detail and used later in the analysis using Dips software. Furthermore, the results of field and

laboratory tests associated with the shear strength parameters ( $c, \phi$ ) along with JRC and JCS are summarized in Table 2.

**Table 2.** Field and laboratory findings of shear test of discontinuity conditions in this study.

Sites	Geologic Unit	Rock Type	Field		Laboratory			
			JRC	JCS (MPa)	Cohesion (Kpa)		Friction Angle (deg.)	
			Avg.		Peak ( $c_p$ )	Residual ( $c_r$ )	Peak	Residual ( $\phi_r$ )
S- 1	Dedebeleni	Limestone	8	49.5	648	462	34	30
S- 2	Beyreli	Sandstone	13	47.5	1030	765	31	29
S- 3	Beyreli	Limestone	11	46.5	963	621	31	29
S- 4	Kuşakdağı	Limestone	10	58	1638	1336	33	31
S- 5	Yarıcak	Limestone	12.5	53	2014	1659	38	36
S- 6	Boyaltepe	Limestone	15	63	909	816	33	31
S- 7	Korualan	Limestone	17	64	1419	1099	39	37

Likewise, other results of laboratory tests performed on the rock material (core)

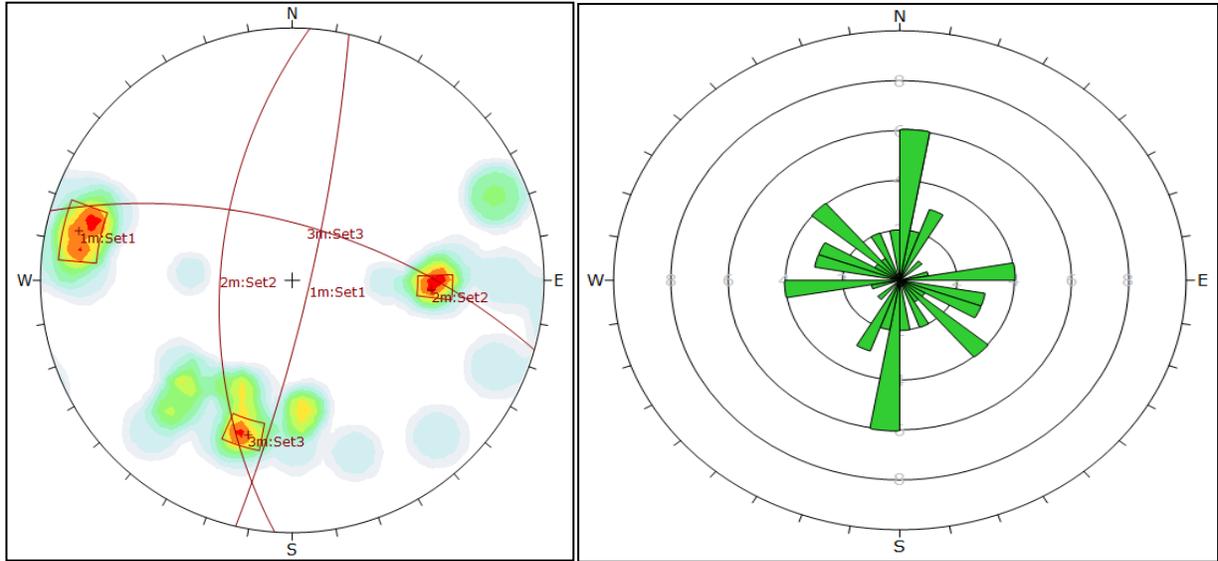
during this analysis, along with their statistics, are shown in Table 3.

**Table 3.** Laboratory tests results of some geomechanical properties for this study.

Property	Tests Number	Min.	Max.	Mean	Sta. Dv.
Compressive strength (Uniaxial) UCS, Mpa	51	1.95	2.98	2.53	0.89
strength (point load test) index (Is(50), MPa)	74	66.2	98.60	70.73	5.21
Unit weight ( $\gamma$ , kN/m <sup>3</sup> )	27	25.59	26.18	25.98	0.45

The data in Tables 1 and 2 were utilized as input parameters in the generation of the data clustering of discontinuity planes, such as a pole, rose map, and contour diagrams (Figure

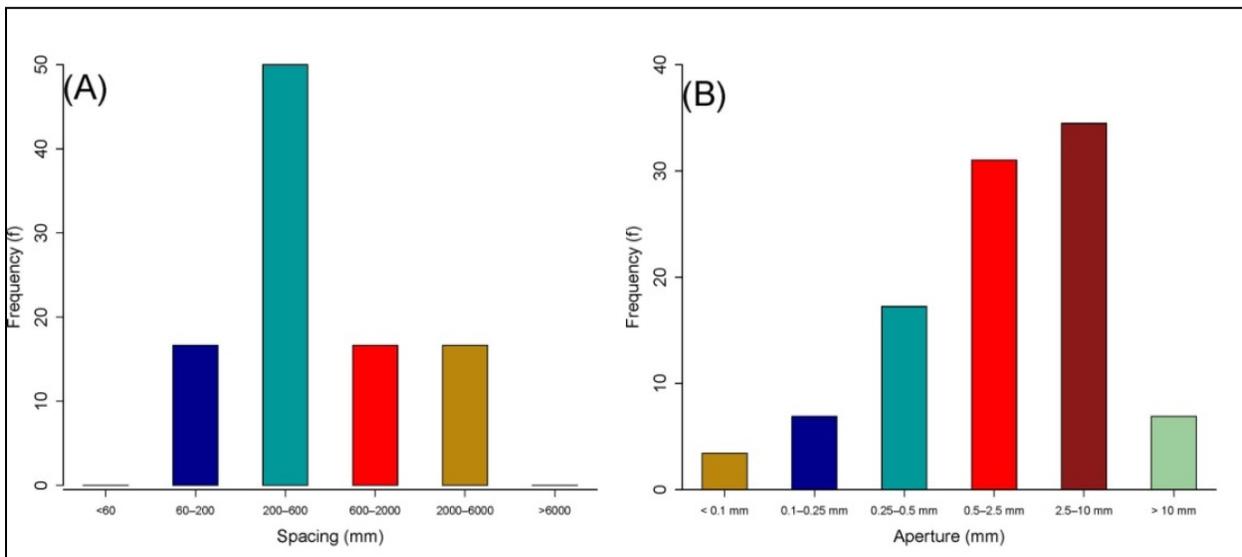
2). Most of the rock masses of the cut slopes analyzed are intersected with three joint sets (J1, J2, and J3) (Figure 2).



**Fig. 2.** Determining prevailing joint sets, through rose (left) and contour diagrams (right) using Dips analyst software.

Then, the geometric properties (spacing, persistence, aperture, and roughness) of the discontinuities are examined in the field survey as well and statistically represented with the chart in Figure 3. The plotted discontinuity spacing versus cumulative

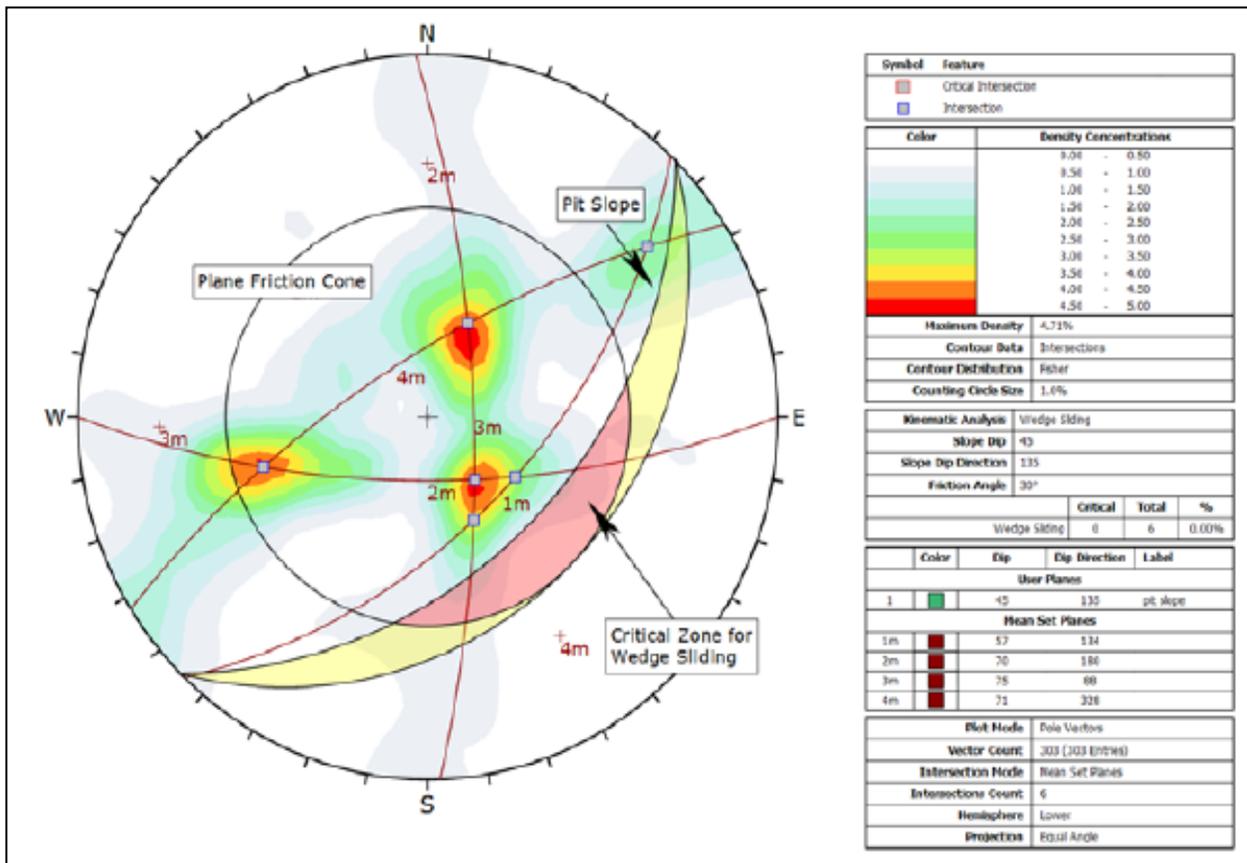
frequency histogram generally very closed spacing fall between 0.6 m and 2 m (Figure 3a). Meanwhile, the average discontinuities aperture range is between 0.5 mm and 10 mm that is partly open to the wide near-surface (Figure 3b).



**Fig. 3.** The statistical distribution of discontinuity spacing (a), and aperture (b) were measured in this study.

The persistence of discontinuities ranges between 3 m and 10 m, it may be less at depth, thus, it also classifies as medium-high persistence. In addition, discontinuity asperities were already evaluated as undulating rough to smooth with a roughness

coefficient (JRC) ranging between 8 and 17 (Table 2). Finally, the slope stability is supplemented accordingly as given in Figure 4.



**Fig. 4.** Kinematic slope stability analysis for present study using Dips analyst.

The kinematic analysis illustrated in Figure 4 is used to determine the probability of each failure type and its proportion of occurrence

for each site, which is then summarized in Table 4.

**Table 4.** Summary of kinematic analysis results for this study, according to Goodman's (1980) guidelines.

Sites	Failure risk probability (%)		
	Planar sliding	Wedge sliding	Flexure toppling
S- 1	*All 21.21 Set (1) 100	*I <sub>1</sub> 290/47 (27.82)	All 6.25
S- 2	0.0	I <sub>2,3</sub> , 320/46 (25.88)	All 11.43 Set (1) 100
S- 3	0.0	I <sub>1</sub> , I <sub>2</sub> (5.71)	All 17.14
S- 4	All 11.3 Set (1) 100	I <sub>1-2</sub> , 021/54 (15.29)	All 2.86
S- 5	All 5.71 Set (1) 33.33	I <sub>1-2</sub> , 021/49 (20.71) I <sub>2,3</sub> 033/51 (20.71)	All 2.86
S- 6	All 2.86	I <sub>2</sub> -slope 303/42 (3.19)	All 14.29 Set (1) 25.0
S- 7	All 2.70	I <sub>2</sub> -slope, 139/7 (9.02)	All 16.27 Set (1) 100

\* All: implies that both geometrical and weather influences together affect the slope instability, I<sub>2-3</sub>, 320/46: refers to the orientations of the line resulting from the intersection of two joints.

The kinematic analysis approach results inserted in Table 5 were analyzed and discussed in detail in the discussion section.

**Table 5.** Summary of failure probable mode inferred with kinematic analysis.

Sites	Failure mode		
	Planar	Wedge	Toppling
S- 1	*Yes	Yes	No
S- 2	No	Yes	Yes
S- 3	No	No	Yes
S- 4	Yes+	Yes	No
S- 5	Yes	Yes	No
S- 6	No	No	Yes
S- 7	No	Yes+	Yes

\* Yes: Failure occurs; Yes<sup>+</sup>: failure probability >10 %; and No: for failure probability < 5%.

In addition to kinematic analysis, site measurements relating to the RQD, Uniaxial Compressive Strength (UCS), joint characteristics, and groundwater conditions are utilized to estimate the perusal RMR<sub>b</sub> and

SMR rating values in this study (Table 6). The SMR values are also quantified based on the RMR<sub>b</sub> perusal rating, the joint and slope adjustment factors (F1, F2, F3, and F4) for each site slope. Based on the all obtained

results, the stability description for each slope is provided in Table 7.

**Table 6.** RMR<sub>b</sub> and SMR quantification rating outcomes for all the slopes.

Sites	RMR <sub>b</sub>	SMR** rating values calculated for each discontinuity set (J1, J2, and J3)						SMR** slope's average
		J1		J2		J3		
		Ky	Dv	Ky	Dv	Ky	Dv	
S-1	54.4	28.4	64.4	56.75	64.4	64.4	53.5	55.30
S-2	52.5	54.5	50.75	54.5	50.75	55.4	50.75	52.67
S-3	49.2	56.3	53.45	57.2	53.45	57.2	53.45	55.17
S-4	51.5	58.6	58.6	59.5	34.5	59.5	55.57	55.16
S-5	51.5	42.6	39.75	42.6	39.75	43.5	33.5	40.28
S-6	50.5	42.4	42.4	41.5	42.4	41.5	42.4	39.10
S-7	57	51.0	31.5	41.5	45.25	51.0	39	44.65

\* Ky: Planar, \*\*Dv: Toppling, and \*\*\*kky: Wedge failure; **SMR\*\*** = RMR<sub>basic</sub>+ (F1.F2.F3) + F4

**Table 7.** Stability definition based on SMR values and according to Romana (1985).

Sites No.	SMR	Class No.	Slope description	Stability	Inferred failure from SMR rating	Possibility of slope failure stability %
S -1	55.30	III	Normal	* PS	P/W	40
S -2	52.67	III	Normal	PS	P/W	40
S -3	55.17	III	Normal	PS	P/W	60
S -4	55.16	III	Normal	PS	P/W	40
S -5	40.28	IV	Bad	**US	Big planar or toppling	0
S -6	39.10	IV	Bad	** THE	Planar	20
S -7	44.65	III	Normal	PS	P/W	40

\* PS: Partially stable; \*\*US: unstable; P/W: Planar along some joints or wedge failure.

## 5. Discussion and assessment

In the context of this study, in addition to using kinematic and SMR ratings, other impacts contributing to slope instability have

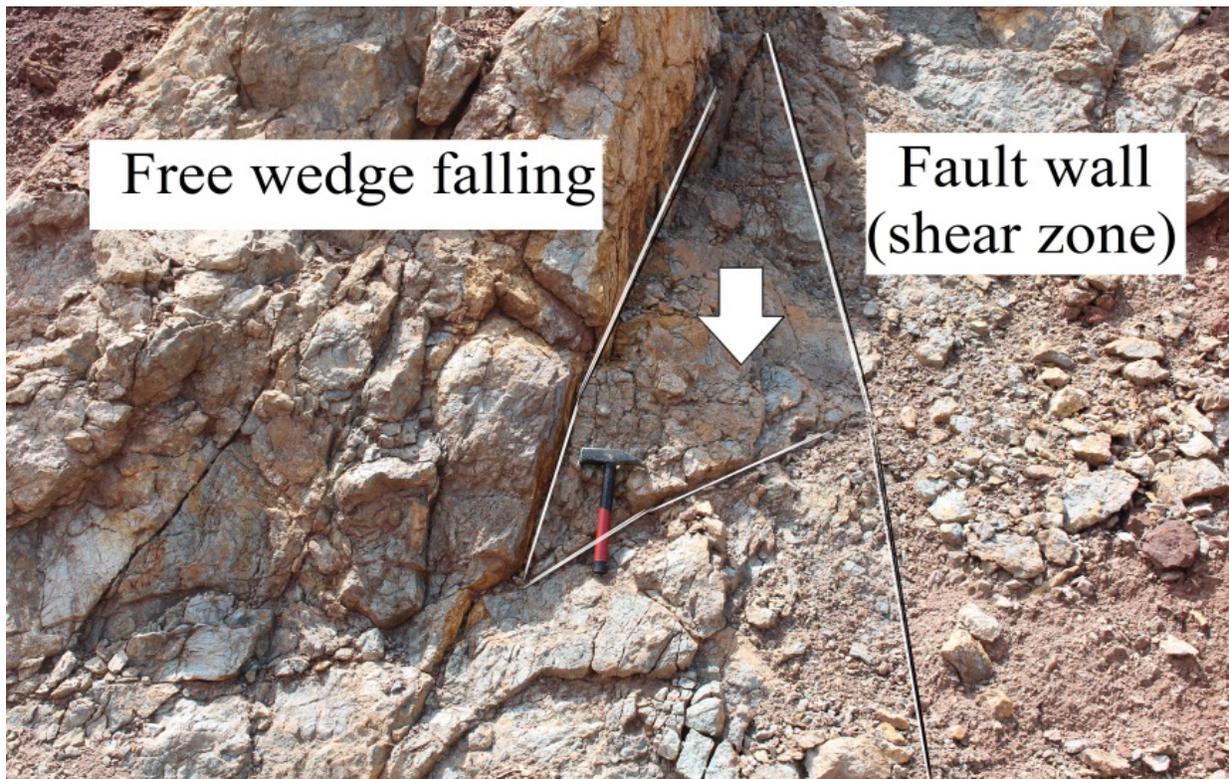
also been discussed and attempted to be included in the final sustainability assessment.

## 5.1. Kinematic analyses

Based on the kinematic analysis results, it can be argued that three site slopes (S-1, S-4, and S-5) satisfy the required conditions for the planar failure occurrence. The dip orientation of joints and slope face, beddings plane is almost in parallel ( $\pm 20^\circ$ ), and the daylight conditions resulting from the J1 critical joint, thus, a planar failure occurrence is predicted in these three cut slopes, which may be due to sliding of the blocks along any single joint line. Unlike, no planar failure is expected for S-2 and S-3 cut slopes, this might be due to the angular geometric ( $>20^\circ$ ) relationship between the face of slope and prevalent discontinuity sets. In comparison, it should be noticed that the wedge failure probability

percentage is substantially higher than that of planar and toppling failures.

The main reason for wedge failure formation can be due to the three dominant sets of joints (J1, J2, and J3), or to the appearance of a shear zone within the rock masses. This implies that the potential mode of failure is mainly depending on the nature of discontinuities and shear strength along with them. For example, a wedge failure is likely to occur on certain site slopes although it has one prevailing joint set, this may be due to shear zones' presence or stress cracks within the rock mass itself (Figure 5).



**Fig. 5.** Structurally- controlled (wedge) failure formed by shear zones in Bozkır units in the present study.

On contrary, no circular failure is expected in all site slopes, this may be due to sliding surface along discontinuities (Wyllie, 2017), or because of the steep dipping of discontinuities towards slope face, as well as a critical high of slopes and irregular slope face that can interfere with other triggering factors, to failure in slope. These conditions are therefore sought to be researched in detail than summarized with the term "All" in the legend of Table 4.

## 5.2. SMR rating

Table 6 shows that despite their high RMRb values, slopes belonging to Bozkr units with good rock quality remain unstable and vulnerable to failure. S-6, for example, although having good RMRb values (50.4), sounds like a very problematic slope site.

Table 7, shows that 5 of the slopes are partly stable or under the risk of failure. However, practically this means that most of the slopes under investigation are partially stable at the present, but could lose their strength over time under various external effects. Thus, several further failures are likely to occur unless remedial action is taken.

Other hands, The failure modes indicated using SMR rating are varied slightly from those inferred from kinematic analysis, which could be due to a variety of factors including lithology, structure, and excavation method effects on those slopes.

## 6. Conclusion

This research project has helped in identifying rock slopes instability with its associated problems in the KAR. thus, will help in mitigating the slope failure reasons and provide security for road users. This study concludes that most of the seven investigated slopes are partially stable. Thus, remedial action is very urgently needed to prevent any potential degradation of the remaining major blocks. Moreover, our results confirm that the findings obtained from the kinematic analysis fit well with the current site conditions. Hence, kinematic stability research is crucial to study the sustainability of essential infrastructures like roads. This instability may be due to reduced normal stress and increased shear stress along discontinuity planes.

We remarked that the SMR does not consider the slope height. It is only taking into account the orientations of discontinuity in the failed mass, which is the main drawback in slope stability assessment using this rating. Finally, we are convinced that a thorough analysis of the investigated slopes by the equilibrium limit technique would be useful for a better understanding of the possible failure mode, failure surface conditions, and impact of joints leading to unstable blocks.

## Author's contributions

Ali Ferat Bayram delineated the research and supported the evaluation and debate. Ahmed Ibraheem Mohamed has carried out fieldwork, data collection, and laboratory testing. Arsalan Ahmed Othman assisted with the manuscript writing, preparation, and discussion. The manuscript was reviewed and revised by all the authors.

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