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Construction risk control technology of a large tunnel complex in urban area

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The stability of the surrounding rock analysis and evaluation during tunnel construction is the basis of tunnel construction risk control. In this paper, we focus on the stability of a large-scale transportation tunnel complex during its construction in a densely-populated urban area. The tunnel complex includes seven shallow-buried tunnels with large cross-sections. In order to gain insight into the excavation influence of the different tunnels, stability analyses were first carried out using FLAC^{3D} numerical simulation. Results showed that the tunnels were subjected to heave and crown settlement induced by adjacent excavation. Also, stress concentrated in the rock blocks connecting different tunnels. Subsequently, a bench-scale model test was performed to understand the failure of the rock blocks and to examine the accuracy of the numerical simulation. The test results agreed well with the numerical simulation. Based on the numerical and test results, the mechanism of the rock blocks failure was explained and construction risk control technology to stabilize the rock blocks was proposed. The construction risk analyses revealed: 1) tunnels are subjected to significant heave due to the excavation of tunnels located above; 2) the stability of the rock blocks is the paramount determinant for stabilizing the whole tunnel complex; 3) ensuring rock blocks to be in a state of triaxial stress is conducive to its stability and hence the stability of multiple tunnels.

KEYWORDS

construction risk, tunnel complex, stability, rock blocks, laboratory model test, numerical simulation

1 Introduction

The analysis and evaluation of tunnel stability is the basis of tunnel construction risk control (Liu et al., 2018; Xu et al., 2021). The stability of the rock mass during underground construction is the highest priority for engineers and scholars in the field of underground engineering because it is closely related to people's lives and property safety. Previous experience has shown that rock mass failure in underground

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works induces heavy casualties and property loss, especially in densely-populated urban areas. For example, the collapse of a large-scale metro tunnel support system in Singapore claimed four victims and caused subsidence of the nearby Nicoll Highway (COI, 2005; Whittle, 2006). Another serious accident during the metro construction in Sao Paulo (Barton, 2008), seven lives were lost. The risk of tunnel construction is primarily assessed by the geological environment, surrounding rock stability, liner deformation, surrounding rock stress, and so on (Yu et al., 2017; Li et al., 2019).

The construction stability of large tunnel complex is related to the degree of fracture and shear strength of the rock. The correct selection of rock strength is the premise of tunnel stability calculation. Many scholars have made a lot of research on the accurate measurement of rock strength. To better understand the shear behavior of infilled rock joints with standard joint roughness coefficient (JRC) profiles, Zhao (Zhao et al., 2020) used the direct shear method to experiment on sand-filled joints by replicating standard JRC profiles on rock-like materials and placing fill material within the joints. Due to the discrete nature of rock strength, a new natural rock formation strength model was developed based on the shear-related roughness classification of fuzzy integrated evaluation (Zhao et al., 2021a). And other rock strength measurement methods are also available (Zhao et al., 2019; Cao et al., 2016). In fact, the importance of rock mass stability has long been recognized and a great deal of research in recent years has focused on this issue for the underground engineering safety (Liu et al., 2011; Feng et al., 2012; Qian and Lin, 2016; Liang and Liu, 2022). Unfortunately, due to the complexity of geotechnical engineering, stability analysis of rock mass still heavily relies on empiricism even though a few failure criteria (e.g., the Mohr-Coulomb criterion) has been proposed (John, et al., 2007). Zheng et al. (2006, 2008) formulated a strength reduction finite element method (SRFEM) to deal with this problem. Athough still in research phase, the SRFEM has gradually been recognized in engineering practice in China (Zhang, et al., 2007; Pan et al., 2021; Wang et al., 2022).

On the other hand, limited space on the ground has stimulated the development of underground space. For example, in recent years there has been significant construction of underground infrastructure in urban areas with the emergence of adjacent transport tunnel projects (Kim, 1996; Liu et al., 2009; Kong et al., 2016; Hu et al., 2021; Zhao et al., 2021b), even the projects of tunnel complex (multiple tunnels) (Zheng, et al., 2009; Duan and Li, 2012; Zhang et al., 2018; Zheng et al., 2021; Zheng et al., 2022). It is clear that the trend of urban underground excavations is becoming increasingly complex, which makes the complicated stability analysis of rock mass an even more critical issue. And the risks of construction are significantly increased. For tunnel



construction risk control measures, different construction situations require targeted analysis. Safety risk control mainly includes pre-construction risk identification and risk warning in construction (Li et al., 2018). During the risk identification phase, a number of challenges need to be overcome. First, traditional methods require a lot of human resources, materials and time. The second problem is the shortage of experienced risk identification experts. Limited by time and space, timely sharing of knowledge becomes unrealistic. Finally, the tunnel construction team will be disbanded at the end of the construction. Therefore, the accumulation of security risk knowledge is completely blank. On the other hand, the quality of risk warning depends on the monitoring data during construction. However, the formulation of control standards based on monitoring data is not reasonable (Zhou et al., 2015).

This paper, based on an unprecedented tunnel complex that contains seven parallel or intersecting tunnels with shallow depths and large cross sections, investigates the stability of the rock mass using FLAC^{3D} numerical models. Stress distribution, crown displacement and the development of plastic zones induced by excavations of multiple tunnels were analyzed. Results showed that the stability of the rock blocks connecting different tunnels was the primary determinant of the overall stability of the multiple tunnels. Furthermore, in order to examine the accuracy of the above analyses, a scaled model test was performed in the laboratory. The test results confirmed the accuracy of the numerical simulation in terms of the potential failure zones and the stress distribution in the rock mass. Subsequently, the mechanism of the stability of the rock blocks was discussed on the basis of the Mohr-Coulomb criterion. According to the analysis, a practical reinforcement scheme strengthening the rock blocks was proposed to reduce the construction risk of the entire tunnel complex project.

Depth (m)		Description	General properties		
A-A	B-B				
0-16.8	0-21.4	Gray and brown miscellaneous fill	Heterogeneity and low strength		
16.8-28.8	21.4-34.8	Fuchsia and brown argillaceous sandstone	Moderately weathered, high strength and water-sensitive		
28.8-34.8	34.8-47.6	Gray sandstone	Slightly weathered and high strength		
34.8~	47.6~	Brown argillaceous sandstone	Slightly weathered, high strength and water-sensitive		

TABLE 1 Geological profile of the ground.

2 Engineering background

The large-scale tunnel complex, consisting of seven shallowburied tunnels with large cross-sections, is located in the Hongyan Village in Chongqing, China. Specifically, the seven tunnels are as follows (Figure 1): a subway station (SS); a ventilation duct (VD) intersecting with the subway station; two parallel double-lane highway tunnels (HT-L and HT-R); two ramp tunnels (RT-A and RT-B) that are in close proximity to the highway tunnels; an existing railway tunnel (ERT) which was built in the 1970s and now is out of service. Additionally, it is worth mentioning that the minimum distance between the SS and the ERT is only 1.2 m and the minimum distance between the SS and the HT-L is only 2.0 m. The south and west sides of the project are adjacent to populated residential areas with densely built buildings.

A detailed field study revealed that the vertical geological profile of the tunnel complex: the sandstone and argillaceous sandstone strata are overlaid by miscellaneous fill material with a thickness of approximately 20 m, as can be seen in Table 1.

3 Numerical simulation

Clearly, due to the complexity of the multiple tunnels, a twodimensional (2D) model based on the plane strain assumption is incapable of meeting the accuracy requirements of stability analysis. Consequently, in accordance with the actual terrain and engineering design parameters, 3D numerical simulations were performed utilizing finite difference analysis software FLAC^{3D} (Itasca, 2012), which is widely applied in the field of geotechnical engineering.

3.1 Three-dimensional numerical model

The numerical simulation makes the following basic assumptions.

1) The tunnel is located in a semi-infinite horizontal laminated medium, with each layer made up of thin layers that extend infinitely in the horizontal direction.

- 2) Each layer is an isotropic homogeneous medium, and the conditions for coordination of displacements between layers and between subsurface structures and rock layers are satisfied, with no relative slippage or detachment between them.
- The influence of pore water pressure and sand liquefaction is ignored, and only ground stress under gravity's action is considered.

In order to minimize the boundary effect, the model dimensions were selected to be 60 m (length)*160 m (width) *50 m (depth). In terms of boundary conditions, the displacements were set to be zero in all three directions with no horizontal and vertical movements allowed at the bottom of the model. The movements in two horizontal directions were restrained, and only vertical movement was permitted on the four side boundaries.

Shotcrete and lining were modeled using shell elements and elastic solid elements respectively. In addition, the rock mass was modeled utilizing elasticperfectly plastic constitutive elements (based on the Mohr-Coulomb failure criterion) without joint consideration.

The total number of hexahedral elements used for the rock mass was 13, 1642 eight-noded, while 2, 1001 structural elements were used for shotcrete and cladding (Figure 2).

3.2 Excavation simulation

The *Z*-axis is the tunneling direction, with excavation starting from -Z toward +Z direction. The simulation was carried out in the following excavation sequence:

- 1) Firstly, the ERT was excavated using bench cut method with its proper lining.
- 2) Secondly, the SS was excavated using double-wall poilt method followed by the excavation of the VD using bench cut method, and the lining was constructed as soon as the excavations were completed.



TABLE 2 Calculation parameters of rock mass.

Name	γ (N/m ³)	E (MPa)	μ	c (kPa)	φ (°)	S _t (kPa)
Miscellaneous fill	21000	30	0.45	50	8.9	_
Argillaceous sandstone	25900	1420	0.35	100	26.0	115
Sandstone	25200	6892	0.30	700	42.0	302

Notes: γ is the unit weight; E is Young's modulus; μ is Poisson's ratio; c and ϕ are the cohesion and the friction angle; S_t is the tensile strength.

Tunnel name	Shotcrete				Lining			
	t (cm)	γ (N/m ³)	E (GPa)	μ	t (cm)	γ (N/m³)	E (GPa)	μ
ERT ^a	_	_	_	_	30	23000	20	0.3
SS	33	25000	23	0.25	100	25000	35	0.25
VD	27	25000	23	0.25	70	25000	35	0.25
HT-L	27	25000	23	0.25	60	25000	35	0.25
HT-R	27	25000	23	0.25	70	25000	35	0.25
RT-A and RT-B	24	25000	23	0.25	50	25000	35	0.25

TABLE 3 Calculation parameters of shotcrete and lining.

Note: t is the thickness of structures; γ is the unit weight; E is Young's modulus; μ is Poisson's ratio.

^aThe lining of ERT, is built by masonry (without metal bars inside) material.

- 3) Thirdly, after the lining strength of the SS and the VD reached the design strength, the HT-L and the HT-R were excavated respectively using bench cut method followed by lining construction.
- 4) Finally, the RT-A and the RT-B were excavated simultaneously utilizing bench cut method.

For more details regarding these excavation methods can be found in Wang et al. (2010).

3.3 Parameters of numerical model

To determine the geological conditions, a comprehensive site investigation was carried out by Survey and Design Institute of Chongqing, the major contractor of the project. According to the Code for Design on Highway Tunnel (JTG D70-2004, 2004), the rock mass was classified as Grade IV. The survey also found that the tunnel complex was built at an elevation of 65 m above the water table and that pore water and joint water were both in short



supply. Therefore, the influence of water on the rock mass is insignificant. Based on the geotechnical drilling, *in situ* data and laboratory testing, together with previous experience (Qiu et al., 2010) in the Chongqing region, the relevant calculation parameters of the rock mass, shotcrete and liner design parameters used in this study are given in



section. **(B)** Distribution diagram of the minimum principal stress of A-A cross-section. **(C)** Distribution diagram of the deviatoric stress of A-A cross-section.

Table 2. The shotcrete and liner design parameters are listed in Table 3.

4 Results analyses

4.1 Stress analyses of rock mass

1) Stress regime before the excavation of the tunnel complex.

Figure 3A,B shows the distribution of maximum and minimum principal stress while Figure 3C provides plots of the deviatoric stress before the tunnel complex is built.



Negatives indicate compressive stress whereas positives indicate tensile stress Due to the disturbance induced by excavation of ERT. The influence range of surrounding rock is from ERT vault to 6 m above the vault.

Cross-section A-A

Figure 4A and Figure 4B illustrate that the rock mass mainly bears compressive stress and Figure 4C shows the plots of deviatoric stress. Tensile stress zones mainly localize the periphery of individual tunnels. The maximum tensile stress is 0.73 Mpa, occurring near the crown of the SS. This tensile stress far exceeds the tensile strength of rock mass (see Table 2), which will bring to cracking and even small-scale collapse of the rock. Figure 4A clearly shows that stress concentrates in two zones after the



excavation of multiple tunnels: 1) the rock blocks connecting the SS and the HT-L; 2) the rock blocks connecting the SS and the HT-R. The maximum value in the stress concentration zone is 2.1 MPa.

Cross-section B-B.

As shown in Figure 5A,B, the rock mass primarily bears compressive stress and the deviatoric stress plots are shown in Figure 5C. Tensile stress zones only localize the periphery of



The distribution diagram of the plastic zones of the tunnels. (A) The 3D distribution diagram of the plastic zones (tunnel support structures are not shown). (B) The distribution diagram of the plastic zones of the A-A cross-section. (C) The distribution diagram of the plastic zones of the B-B cross-section.

individual tunnels. The maximum tensile stress is 0.97 Mpa, occurring at the toe of the SS's left sidewall. This tensile stress far exceeds the tensile strength of rock mass (see Table 2), which will bring to cracking and even small-scale collapse of the rock. Obviously, stress concentrates in the rock blocks connecting the SS and the HT-L after the excavation of multiple tunnels. The maximum value in the stress concentration zone is 2.4 MPa.

In summary, during the construction of the multiple tunnels, stress concentrates in the rock blocks. This is likely to decrease the strength of the rock mass due to the propagation of discontinuities (such as joints) and ultimately threaten the stability of the tunnel complex. It also should be noted that although the tensile stress exceeds the tensile strength in limited areas and therefore brings cracking of the rock, the area of tensile stress zones is so limited that this would not cause failure of multiple tunnels.

4.2 Displacement analyses of linings

- 1) As can be seen in Figure 6A, the excavation of the SS and other multiple tunnels located above the ERT results in a significant heave displacement of the ERT. In particular, the scope of -20-20 m is subjected to strong influence and the maximum heave is 11.2 mm (curve ②). This heave is triggered by the load release effect induced by the excavation. In addition, it is worth noting that the heave caused by the excavation of the SS is over 10 mm (curve ①), accounting for 89.3% of the final heave displacement (curve ②). Because the lining of the ERT was constructed using masonry (without metal bars inside) whose tensile strength is low, the heave poses a great threat to the stability of the ERT.
- 2) According to Figure 6B, the weight of the overlying rock mass causes the crown settlement of the SS after its excavation with the maximum value of 13.3 mm occurring at the coordinate of 28 m along its axial direction (curve ①). Despite the ERT is located directly below the 0–20 m scope of the SS, curve 1 shows that the crown settlement of the SS in the 20–60 m scope is slightly larger than that of the 0–20 m scope (the difference is only 0.7 mm), which indicates that the weight of the overlying rock mass exerts a stronger influence on the crown settlement of the SS compared with the presence of the ERT.

Curve ② proves that the load release effect induced by the excavation of the multiple tunnels located over the SS leads to a small uplift of the SS (the maximum value is only 0.6 mm). This small heave probably results from the fact that the multiple tunnels are not excavated directly above the SS. Also, the overlying rock mass of the SS is sandstone with high strength, which may contribute to the small heave of the SS.

3) According to Figure 6C, for both the HT-L and the HT-R, the overall crown settlement gradually increases with increasing buried depth along the tunnel axial direction. The maximum settlement values of the HT-L and the HT-R are 13.0mm and 13.8 mm respectively.

Since the VD is excavated directly below the HT-R whereas no excavation is done directly below the HT-L, more remarkable settlements occur in the middle section along the axial direction of the HT-R.

4.3 Plastic zone analyses

As shown in Figure 7A–C, plastic zones of the rock mass mainly localize in three areas: 1) the rock blocks connecting the SS and the ERT; 2) the rock blocks connecting the SS and the HT-L; 3) the rock blocks connecting the SS and the HT-R. Also, plastic zones near the crown of the HT-L have been developed on the ground surface.



Due to the residual strength, being in plastic states does not necessarily mean strength failure of the rock blocks. Nevertheless, numerical simulation clearly proves that the stability of the rock blocks is the paramount factor for stabilizing the entire tunnel complex.

5 Discussions

5.1 Stability of the ERT

The three-dimensional numerical simulation shows that the ERT is subjected to significant heave induced by the excavation of multiple tunnels located above it. On the other hand, because the ERT was constructed in the 1970s (now it is out of service) and its lining was built of masonry (without metal bars inside) material with low tensile strength, the heave may cause instability of the ERT. Clearly, this would threaten the stability of the whole tunnel complex. Therefore, given the fact that the ERT is no longer in operation, it is recommended that the affected section of the ERT (-40m-40 m along its axial coordinate) should be sealed up by concrete filling before the excavation of multiple tunnels for the sake of the overall stability. The timing of the seal operation is crucially important. It must allow enough time for the hardening process of concrete.

5.2 Stability test of rock blocks

It has been widely accepted that it is important to make good use of the bearing capacity of the rock mass during tunnel construction. However, in terms of the construction of the tunnel complex, the bearing capacity of the rock mass is severely undermined by multiple openings in the rock mass and hence causing stress concentration in the rock blocks and the shotcrete and lining. As a result, the stability of the rock blocks emerges as the most critical factor for stabilizing of the entire tunnel complex.

In order to gain insight into the failure process of the rock mass and verify the accuracy of the numerical simulation, a scaled model test was conducted in the laboratory.

5.2.1 Overview of the test

The test model was made according to the A-A cross-section. For the purpose of simplifying the problem, the test was on the basis of the plane strain assumption and the actual terrain was not considered. Moreover, the influence of the ERT was negligible due to the proposal mentioned earlier (the ERT should be sealed up). Figure 8 shows the test model and cross-sectional dimensions.

The test apparatuses consist of a scaled model (including lining models and the simulated rock mass), a set of loading equipment and micro geotechnical pressure cells.

During the test, the ground load level was gradually increased by a step of 20 kN until the failure of the model. Simultaneously, the failure modes and behavior of the rock mass were carefully observed. Micro geotechnical pressure cells were used to measure the stress on the liner.

5.2.2 Test materials and similarity rules

The lining models were made of low-pressure polyethylene. A homogeneous material was used to simulate the actual rock mass (argillaceous sandstone), and the stratum characteristic of the actual rock mass was not considered because of the complexity of the simulation test. The simulated rock mass was made of cement, fly ash, gypsum, river sand and water (25.0%, 13.2%, 6.5%, 5.3%, and 50%, by mass percentage).

Material parameters of the prototype were adopted from actual engineering data and their model counterparts are listed in Table 4. During the laboratory test, five groups of parameters were simulated and the similarity ratios (Physical variables with subscript p refer to the prototype, and physical variables with subscript m refer to the model) are: $C_l = l_p/l_m = 100$; $C_E = E_p/E_m = 100$; $C_p = \rho_p/\rho_m = 1.2$; $C_c = C_p/C_m = 1.7$; $C_{\phi} = \phi_p/\phi_m = 1.2$. Under the premise of ensuring the reliability of the test results, the similarity ratio is determined mainly based on the size of the test setup, the tunnel prototype and the loading capacity of the test chamber.

5.2.3 Test results and analyses

1) Failure modes

The failure process of the rock mass is shown in Figure 9: initially, the destruction occurred at the spring line of the SS 1)

Name	c (kPa)	φ (°)	E (MPa)	ρ (g/cm ³)	μ	t (cm)
Rock mass prototype	252	33.5	4000	24.5	0.35	_
Simulated rock mass	352	30.0	41.5	20.0	0.30	_
Lining prototype	_	_	20	2.55	0.2	30
Lining model	_	_	0.15	0.94	0.2	0.3

TABLE 4 Rock mass and lining parameters of the prototype and model.

Notes: c and ϕ are the cohesion and the friction angle; E is Young's modulus; ρ is the density; μ is Poisson's ratio, t is thickness.



FIGURE 9

Detailed failure modes in the test. (A) Failure modes around the SS. (B) Failure modes around the HT-L. (C) Failure modes around the HT-R.

and then the toe of sidewall of the HT-L 2): next, the rock blocks connecting the SS and the HT-L broke; subsequently, the rock blocks connecting the SS and the HT-R damaged 3); eventually, cracks occurred in the area of d and e and they spread into the ground surface. Model failure occurred when the load level reached 200 kN (Figure 11). A detailed description of the failure modes is shown in Figure 10.

2) Stress distribution on the lining

The failure zones observed in the test were in fairly good agreement with the theoretical failure zones calculated by the strength reduction finite element method (SRFEM), as can be seen from Figure 10A and Figure 10B. The numerical model used in SRFEM is a 2D model and its dimensions and boundary conditions are the same as the 3D model. The parameters of the rock mass are listed in Table 4. Details about the SRFEM can be found in Zheng et al. (2006, 2007).

In terms of the stability of the rock blocks, only the rock blocks connecting the SS and the HT-L (the left rock blocks) damaged in the SRFEM numerical simulation (2D). However, apart from that, the rock blocks connecting the SS and the HT-R (the right rock blocks) also damaged during the test. This discrepancy may be due to the fact that the actual terrain was considered in the SRFEM numerical simulation whereas it was not considered in the model test: the slope of the terrain produced asymmetric rock mass pressure to tunnels whereas the load in the test was nearly symmetric.

Nevertheless, both the test and the SRFEM numerical results showed that the damage of the left rock blocks was more severe compared to the right rock blocks. This is mainly due to the greater thickness of the right rock blocks and hence the greater bearing capacity compared with the left rock blocks.

The pressure exerted on the linings when the rock mass failed can be seen in Figure 10C. Data from the pressure cells SS-2 and SS-4 are 788.74 kPa and 783.33 kPa respectively, which were much greater compared with other pressure cells. This indicates that stress concentrates in the left rock blocks and the right rock



blocks. This test result agrees well with the stress distribution in the 3D numerical simulation.

The failure zones of rock blocks damage in the test is basically consistent with the zones of surrounding rock stress concentration in the numerical simulation, and the lining damage pattern is also basically consistent with the lining stress distribution in the numerical simulation, which indicates that the test and numerical simulation results have certain correlation and can provide support for the conclusion.



5.3 Mechanism of the rock blocks failure

As an example, consider the left-hand rock blocks (Figure 11A) to explain the failure mechanism of the rock blocks. During the whole process, the Mohr circle at the rock blocks underwent five changes: the first time, after the excavation of the SS, the tangential stress at the rock blocks was released significantly due to stress relief, and the Mohr circle was cut with the strength envelope; the second time, after the completion of the SS construction, the Mohr circle diameter decreased, and it gradually moved away from the strength envelope; the third time, a large concentration of tangential stresses occurred during the excavation of the HT-L, and the diameter of the molar circle increased until it was tangential to the strength envelope; the fourth time, after the support was set in HT-L, the radial stress increased and the diameter of the molar circle became smaller and gradually moved away from the strength envelope; the fifth time, after the excavation of the tunnel complex was completed, the molar circle again gradually moved away from the strength envelope.

Clearly, the tangential stress (σ_{τ}) concentration leads to the radial expansion of the rock blocks due to the Poisson

effect. Meanwhile, the tunnel lining provides radial stress (σ_r), turning the rock blocks into a state of biaxial stress. As the degree of stress concentration increases, the tangential stress increases to a certain level when the lining becomes incapable to provide sufficient radial stress (lining is buckled) to sustain the biaxial stress state. Subsequently, the radial expansion increases sharply as a consequence of the end of biaxial stress state. The state of uniaxial stress produces shear failure of the rock blocks, which ultimately leads to the failure of the multiple tunnels.

5.4 Reinforcement for rock blocks

The above analyses suggest that providing sufficient radial stress is vitally important for stabilizing rock blocks. This can be demonstrated by plotting the Mohr's circle diagrams of the principal stresses of the rock blocks.

Based on the plane strain assumption together with ignoring the effect of the ERT and the VD, two cases were calculated using FLAC^{3D}: 1) tunnels without support structures; 2) tunnels with support structures. The principal stress analysis of the left rock blocks is shown in Figure 11B.

Before the excavation of the SS, the Mohr's circle is below the failure envelope and the rock blocks are stable. At this stage the rock mass has not been disturbed by excavation.

Under the condition of tunnels without lining, the Mohr's circle is above the failure envelope after the excavation of the SS, which indicates the beginning of rock blocks failure. When the excavation of the HT-L is completed, the radius of the Mohr's circle increases and hence the Mohr's circle is above the failure envelope due to the tangential stress concentration. At this stage the rock blocks fail.

As for the tunnels with lining, the Mohr's circle is on the failure envelope after the excavation of the SS, indicating the state of equilibrium. When the excavation of the HT-L is completed, the radius of the Mohr's circle increases but the Mohr's circle moves to the right due to the radial stress provided by lining. Therefore, it is still below the failure envelope: the rock blocks remain stable.

In engineering practice, auxiliary stabilization measures that can provide radial stress (e.g., rock bolts and prestressed anchor cables) can improve the stress state of the rock blocks, ensuring the rock blocks are in a state of triaxial stress. It should be noted, however, the number of these measures must be limited in order to avoid excessively weakening the integrity of rock blocks.

6 Conclusion

This paper presents stability analyses of a large-scale tunnel complex based on a case study in Chongqing, China. The

analyses were performed using FLAC3D and compared with a scaled model test in the laboratory. The following conclusions can be drawn from the study:

- The stability of the surrounding rock analysis and evaluation during tunnel construction is the basis of tunnel construction risk control. For tunnel construction risk control measures, different construction situations require targeted analysis.
- 2) During the excavation of the multiple tunnels, stress concentrates in the rock blocks and the plastic zones localize in the rock blocks. Because the rock blocks function as the main bearing body and connect the lining of different tunnels to form a bearing system, the stability of the rock blocks is the paramount determinant for stabilizing the entire tunnel complex.
- 3) The ERT and the SS are subjected to heave displacement due to the excavation of the tunnels located above. Particularly, the ERT is strongly influenced. In order to stabilize the multiple tunnels, it is proposed to seal up the ERT by concrete filling given it is no longer in operation.
- 4) The stability of rock blocks is the key to control construction risk. Ensuring the rock blocks to be in a state of triaxial stress is conducive to its stability. This can be achieved by applying rock bolts and pre-stressed anchor cables in engineering practice. Radial stress provided by these measures can turn the rock blocks into a state of triaxial stress, thereby enhancing its bearing capacity.
- 5) Multiple tunnel failure modes were obtained in the scaled model experiments. Despite some limitations in the study, the conclusions still provide important technical guidance to the construction of this tunnel complex. And it can reduce construction risk.

Data availability statement

The original contributions presented in the study are included in the article/Supplementary Material, further inquiries can be directed to the corresponding authors.

Author contributions

XG: contributed to conceptualization, methodology, analysis, and writing; CK, DW, and FL: contributed to validation, resources, and funding acquisition; CK and HW: contributed to experiment design, and data collection; FL, ML, and SR: contributed to investigation, supervision, and review editing. All authors have read and agreed to the published version of the manuscript.

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Conflict of interest

Authors DW and ML were employed by the company of Chongqing urban constrution investment (group) Co., Ltd.

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