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Propagation and impacts on roadway of mining-induced far-field strong tremors: insights from numerical simulations

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Mining-induced far-field mine tremors, which often cause strong ground tremors, are receiving more attention due to their increasing occurrence. Investigating the rock burst risk of roadway caused by those tremors is crucial to ensure production safety. In this study, Variational Mode Decomposition was used to investigate the wave characteristics of strong mine tremors. The propagation and attenuation of these tremors were explored using the dynamic analysis of Flac3d. The amplification factor was introduced to assess the impact of these tremors on roadways. Plastic zone volume increment and Brittle Shear Ratio (BSR) are used to assess the roadway failure and the rock burst potential caused by these tremors, respectively. The main findings are as follows: 1) Compared with the main frequency of near-field mine tremor waveform, the far-field mine tremors waveform are mainly low frequency below 5 Hz; 2) In the simulation, peak vibration velocities of P-wave and S-wave follow a power-law decay as the propagation distance increases, with P-wave attenuating faster than S-wave; 3) Under similar conditions, P-wave induce higher vibration velocities than S-wave, but S-wave generally exhibit a greater amplification factor than Pwave; 4) When the direction of dynamic load is consistent with the direction of maximum principal stress, the rock burst potential of roadway is higher, which explains the phenomenon that the rock burst potential of roadway under S-wave loading is higher than that under P-wave.

KEYWORDS

far-field strong mine tremors, waveform decomposition, dynamic analysis, rock burst potential, plastic failure

1 Introduction

Strong mine tremors reported worldwide are becoming an increasing threat to safe mining operations as mines extend to the deeper subsurface (Małkowski and Niedbalski, 2020). The essence of tremor is the fracture initiation, propagation, and reactivation of rock, usually triggered by the redistribution of regional stress due to mineral extraction in large scale. It has been well-documented that these rock failure processes are associated with the release of detectable seismic wave. Through the deep analysis of seismic wave, especially the method of focal mechanism inversion, the relevant

information on source rupture can be retrieved and the failure mode of the source can be revealed (Cesca and Grigoli, 2015). In recent years, a majority of coal mines have extended their operations to depths exceeding 600 m in the East, North, and Northeast of China (Qiao et al., 2014). Due to the high static stress of deep mining roadway, strong mine tremors could easily trigger roadway's dynamic failure. The occurrence of strong mine tremors and their induced dynamic disasters is a recurrent issue that significantly hazards the safety of underground miners, nearby residents, and the efficiency of mining operations. On 9 June 2019, a mine tremor with a magnitude of 2.3 struck the Longjiapu coal mine, resulting in 9 fatalities and 10 injuries. This tragic incident underscored the risks associated with strong mine tremors.

In underground mining, strong mine tremors can be divided into coal pillar failure (Mendecki et al., 2019), mining-induced structural and fault activation (Islam and Shinjo, 2009), and the brokening of the overlying hard roof (Jiang et al., 2019). Researchers have undertaken comprehensive investigations the mechanisms, focal mechanisms, predictive measures, and preventative strategies of these strong mine tremors (Sen et al., 2013; He et al., 2017; Wang, 2023). Nevertheless, in recent years in China, a type of strong mine tremors induced by coal mining has attracted strong attention: the mine tremors caused by the collapse of the hard, extremely thick roof in the high above of the coal seam. Because these strong mine tremors' vertical position is approximately 100 m or even farther from the coal seam, they are referred to as "far-field strong mine tremors". These tremors, given their proximity to the surface, frequently result in significant vibration on the ground, instigating widespread concern. Currently, coal mines with a high frequency of such far-field strong mine tremors include Hongqinghe (Zhang et al., 2021), Jinjitan, Shilawusu (Yang et al., 2023), Dongtan, Nantun (Guo et al., 2022; Liang et al., 2022), Baodian (Cao et al., 2015; He et al., 2022). Notably, Dongtan, Nantun, and Baodian all fall within the Yanzhou coalfield, positioned near one another, and share similar mining conditions characterized by the presence of substantial, maroon thick sandstone layers above the coal seam, often referred to as the "red layer". According to statistics, the cumulative count of strong mining tremors caused by coal mining under the "red layer" has reached more than 600 times. The mine tremors caused by coal mining under this special roof structure have attracted some Chinese scholars to study this phenomenon. Wang et al. (Wang et al., 2022) found that the mechanism of strong mine tremors in Dongtan Coal mine is mainly the structural instability of mine tremors which is dominated by tensile failure. However, Wu et al. (Wu et al., 2023a) conducted a comprehensive study of the focal mechanisms for 27 strong mine tremors in Dongtan Coal Mine, revealing that the predominant mechanism was attributed to pure shear failure. Zou et al. (Zou et al., 2022) contend that the position of the strong mine tremors in Dongtan Coal Mine is situated within the "red layer," and they propose that the severity of these mine tremors can be effectively mitigated through the implementation of deep hole blasting. Moreover, Wu et al. (Wu et al., 2023b) conducted simulations using LS-DYNA and UDEC 7.0 to model the pre-splitting of rock and the migration of overburden during deep-hole blasting at the working face. Their research reveals the mechanism of deep hole blasting in the face to achieve roof pre-cracking and pressure relief. Jiao et al. (Jiao et al., 2021) discovered that strong mine tremors result from the collaborative failure of multiple strata, as deduced from waveform analysis. They also propose using the rate of surface subsidence as a predictive indicator for the occurrence of strong mine tremors.

Current research predominantly centers on predicting strong mine tremors, assessing preventive and control strategies, and exploring focal mechanisms. The practice shows that this type of mine tremor rarely causes rock burst directly, and generally only has strong seisesthesia. Therefore, it is necessary to study the degree of roadway rock burst risk induced by this type of mine tremors. Dynamic analysis in Flac3d serves as a valuable tool for investigating dynamic stress induced by tremors in coal mining, and many studies have employed this approach to simulate mine tremors (Itasca Consulting Group, 2023). Liu et al. (Liu et al., 2019) discovered that the resonant frequency of the roadway in Zhuji Coal Mine is 35Hz, and the intensity of the mine tremors wave reaches its peak at this specific frequency. Kong et al. (Kong et al., 2019) conducted a study on the deformation and rock burst potential of roadways under various combinations of dynamic and static stress. Wang et al. (Wang et al., 2022) and Cao et al. (Cao et al., 2023) delved into the mechanisms of rock burst occurring in steeply inclined coal seams under the influence of dynamic and static stress combinations. Zhu et al. (Zhu et al., 2021) illustrated the evolution process in displacement, stress, and velocity along the fault plane induced by dynamic stress. This study takes the Dongtan Coal mine, where far-field strong mine tremors often occur, as the engineering background. Using Variational Mode Decomposition (VMD) decomposition and time-frequency analysis, the typical characteristics of far-field strong mine earthquakes are obtained. A numerical model based on Mohr-Coulomb and strain softening failure criterion is established, and the damping coefficient of the model is strictly calibrated. Meticulously analyzes the influence of different frequencies and dynamic load types (P-wave and Swave) on the plastic failure and rock burst potential of the roadway. The propagation law of dynamic load in complex multilayer media and the amplification effect of roadway surface are analyzed. The study's findings offer valuable insights into the attenuation of dynamic loads, enhancing our understanding of far-field strong mine tremors.

2 Study site

2.1 Engineering background

Dongtan Coal Mine is located in Yanzhou, Shandong, China, and attracted widespread attention due to frequent strong mine tremors (Figure 1A). In the course of mining operations in the Dongtan No. 6 mining area, a multitude of tremors were detectable by the earthquake monitoring network. These tremors resulted in the shaking of residential structures in proximity to the mine, thereby engendering significant social repercussions. The current main coal seam is 3^{upper} with a buried depth of about 680 m. The thickness of 3^{upper} coal seam is in the range of 4.12–6.79 m, with an average value of 5.41 m. The coal seam is nearly horizontal and the geological structure is relatively simple. The maximum horizontal principal stress is 24.96~27.12 MPa at the direction of SE 148.93°~150.00°. The maximum horizontal principal stress



is 2.15~3.29 times the minimum horizontal principal stress and 1.44~1.72 times the vertical stress.

Dongtan coal mine is divided into 7 mining areas, and the No.6 mining area is located in the south of the coal mine. Longwall (LW) fully mechanized mining technology is adopted. Four working faces have been exploited with the mining sequence of LW63^{upper}04 \Rightarrow LW63^{upper}05 \Rightarrow LW63^{upper}03 \Rightarrow LW63^{upper}06 (Figure 1B). The former three are finished and turned into goaves. As of October 2022, the LW63^{upper}06 has been advanced to 879 m (the panel length is 1,500 m). The lithology of the roof and floor of the 3^{upper} coal are mainly sandstone with high strength, and their main mechanical parameters have been measured in the laboratory (see Supplementary Table S1).

2.2 Characteristics of strong mine tremors

A 16-channel microseismic (MS) monitoring system, developed by the Institute of Innovative Technologies EMAG, Poland, has been installed to determine the location and energy of mine tremors. DLM microseismic sensors with an operating frequency of 0.1-200 Hz are included in this system. MS signals are transmitted via electric transmission lines to the surface. Eight sensors (4#, 5#, 6#, 14#, 15#, 17#, 28#, 29#) are installed in different positions of underground roadways in the No. 6 mining area and two sensors (16#, 29#) are installed on the ground to form a joint stereoscopic MS monitoring network (Figure 2A), which improves the positioning accuracy of the microseismic monitoring system in the vertical direction. The sampling frequency range of these sensors is 500 Hz. The No. 6 mining area of Dongtan Coal Mine experienced a substantial number of strong mine tremors during previous coal mining. These tremors prompted considerable social anxiety and drew significant government attention. Specifically, 15 mine tremors, with magnitudes exceeding 1.5, were recorded on LW63^{upper}03, while LW63^{upper}04 and LW63^{upper}05 reported 39 and 56 such tremors, respectively. Compared with other working faces, mine tremor on working face LW63^{upper}06 is more severe. Since the start of mining in February 2020, there have been more than 34 large energy mine tremors with a magnitude of more than 2.0 (Figure 2C). It can be seen from Figure 2B that strong mine tremors mainly occur in the roof rock layer, especially near the lower boundary of the "red layer".

In Dongtan Coal Mine, there have been numerous strong mine tremors, but they rarely lead to roadway rock bursts. This phenomenon might be closely linked to the vibration characteristics of these tremors and the distance between their source and the roadways. Figure 3 presents waveform diagrams of two distinct types of strong mine tremors. Wave 1, originating from the red layer located 180 m away from the coal seam, is classified as a far-field wave. Wave 2, originates from the central sandstone layer just 30 m from the coal seam. Both waves are recorded by the 14# microseismic sensor. Both mine tremors display approximately 2 s of intense fluctuations, followed by a phase of minor amplitude variations. This temporal pattern sets them apart from natural earthquakes, which typically last for over 10 s. Notably, Wave 2 has a higher dominant frequency than Wave 1. Additionally, the maximum amplitude of Wave 2 seems to exceed the detection range of the microseismic sensor. Zhou et al. (Zhou et al., 2016) used numerical simulations to show that medium properties cause frequency attenuation in blast waves over distance. Similarly, the stark contrast between the two waveforms might result from either frequency and amplitude attenuation during propagation or significant differences in tremor sources, making a clear explanation challenging. It is worth noting that what we analyze here is the frequency and vibration speed when the vibration propagates from the source to the roadway.

Ambient noise and instrument-related interference can complicate the recorded waveforms. Variational Mode Decomposition (VMD), a non-recursive signal processing method developed in 2014 was utilized to address this (Dragomiretskiy and Zosso, 2013; Liu et al., 2016; Liu et al., 2023). VMD is renowned for its effective noise reduction capabilities and its ability to mitigate the issue of frequency aliasing commonly seen in empirical mode decomposition (EMD). Figure 4 shows the result of two typical



Joint underground-surface MS monitoring network and Strong mine tremors during mining of LW63^{upper}06: (A) Joint underground-surface MS monitoring network in the study area, (B) mine tremors profile locations, (C) mine tremors plane locations.



waveform's VMD and Hilbert spectrum. The main frequency of wave 1 is in the range of 0–5 Hz, and the duration is longer than 2 s (Figure 4B). The components of wave 2 are quite complex, with frequencies ranging from 0 to 80 Hz, but the components in the range of 20–80 Hz last for about 1 s and disappear, and the components below 10 Hz last for more than 3 s (Figure 4D). The analysis above indicates that the high-frequency component of the waveform decays much faster than the low-frequency component.

3 Creation and calibration of numerical model

3.1 Global model and simulation approach

To analyze the effect of vibration on the roadway after it is generated from the tremor source, a numerical model of fine

calibration is established for analysis. According to the actual measured borehole histogram in study area, a global model was established after proper simplification of the rock strata (Figure 5A). The model size was 150 m ×30 m × 164 m (length × width × height). The number of zones in numerical modeling is 355,000, the size of the rock strata grid is less than 2 m, and the size of the coal seam grid is about 0.5 m. The four sides and the bottom of the model were constrained by velocity. The average burial depth of LW63^{upper}06 is 680 m, so vertical stress of 17.2 MPa was applied to the top of the model. Horizontal stress is applied by the real ground stress test results detailed in section 2.1. The Strain-softening model was used to simulate the coal seam, and the Mohr–Coulomb model was used to simulate other rock strata.

The simulation process, detailed in Figure 5B, began with the assignment of initial parameters and equilibrium calculations. Subsequently, the roadway was excavated, and the model was recalculated to reach equilibrium. Finally, the dynamic calculation



Spectrum characteristics of typical mine tremors. (A) VMD result of wave 1, (B) Hilbert spectrum of wave 1, (C) VMD result of wave 2, (D) Hilbert spectrum of wave 2.



model was opened, various dynamic loads were applied, and the study examined the propagation and attenuation patterns of these dynamic loads and their rockburst potential on the roadway.

3.2 Determination of physical and mechanical parameters

The physical and mechanical parameters of the rock mass were determined through laboratory tests conducted on test specimens, combined with a parameter conversion method previously proposed. To establish this relationship, data from 44 different sources were collected to compare experimental mechanical parameters with simulated ones, resulting in a linear conversion relationship as follows (Mohammad et al., 1997):

$$y_{\mu} = 1.0556 \times x_{\mu} \tag{1}$$

$$y_E = 0.4692 \times x_E \tag{2}$$

$$y_{UTS} = 0.4943 \times x_{UTS} \tag{3}$$

$$y_{UCS} = 0.2837 \times x_{UCS} \tag{4}$$

Where, y_{μ} and x_{μ} represent simulation and laboratory Poisson's ratios, respectively; y_E and x_E represent simulation and laboratory Young's modulus, respectively; y_{UTS} and x_{UTS} represent simulation and laboratory Uniaxial tensile strength, respectively; y_{UCS} and x_{UCS} represent simulation and laboratory Uniaxial compressive strength, respectively.

Dongtan Coal Mine extensively measured the mechanical parameters of coal and rock (see Supplementary Table S1). Utilizing the conversion relationships outlined in formulas (1), (2), and (3), the mechanical parameters for the model's rock strata were determined (shown in Supplementary Table S2). The coal seam employs a strain-softening constitutive model, with its mechanical parameters calibrated through numerous iterations.

3.3 Calibration of numerical simulation parameters

3.3.1 Calibration of strain softening parameters for coal

In Dongtan Coal Mine, the roof and floor primarily consist of high-rigidity sandstone, while the coal seam is notably weaker. Therefore, the impact of the coal body's strain-softening characteristics on roadway stability must be taken into account. A Fish program was used to calibrate the strain-softening parameters of coal (Supplementary Figure S1). The simulation model matches the dimensions of the laboratory coal sample, with a diameter and height of 50 mm and 100 mm, respectively (Figure 6A). During the simulation, parameters such as elastic modulus, cohesion, and softening coefficient are adjusted until the simulated stressstrain curve closely aligns with the experimental stress-strain curve (Figure 6B). The validated strain softening parameters for the coal are presented in Supplementary Table S3.

3.3.2 Calibration of Rayleigh damping coefficients

Attenuation during stress wave propagation can be simulated in Flac^{3d} numerical simulation by applying rayleigh damping, local damping, or hysteretic damping. In the dynamic calculation process of this paper, rayleigh damping is used. The Rayleigh damping equation can be expressed as a matrix:

$$[\complement] = \alpha[M] + \beta[K] \tag{5}$$

Where, [C] is the damping matrix, α is the mass-proportional damping constant, β is the stiffness-proportional damping constant, [*M*] is a mass matrix, and [*K*] is a stiffness matrix.

According to the FLAC^{3D} manual, the damping coefficient and center frequency must be determined for Rayleigh damping. For geological materials, the ξ_{min} is generally 2%–5%. The damping coefficient selected in this study was taken as 2%. The f_{min} can be determined from the velocity time history curve of the model during undamped free vibration (see Supplementary Figure S2). In the figure, one complete cycle of z-velocity takes 0.313 s, which converts to frequency is 3.2 Hz. This observation indicates that the natural vibration frequency of the model is 3.2 Hz.

4 Dynamic load propagation, attenuation, and its impact on roadway

4.1 Characteristics of wave propagation attenuation

As depicted in Figure 7, a total of 12 monitoring points were strategically positioned within the meticulously calibrated numerical model to meticulously capture and assess alterations in vibration velocity induced by dynamic loading. This arrangement comprises nine monitoring points distributed along the path from the source to the roadway's apex, with an additional three points situated at the roadway's base. In alignment with field conditions, the dynamic load source is applied to the primary key layer, known as the "red layer." The figure also delineates the P-wave's propagation pattern, encompassing a 10 m×10 m spatial extent, with its initial vibration directed forward along the *z*-axis.

4.1.1 Propagation and attenuation of P-Wave

When using Flac3d to simulate vibration waves, many researchers assume that the waveform is a sine wave, and this practice is also adopted in this paper (He et al., 2020; Mukhamedyarova et al., 2023). Sinusoidal wave at frequencies ranging from 2 Hz to 100 Hz, each with an amplitude of 3 m/s, were applied to the model. The application area of the P-wave is depicted in Figure 7. Figure 8 illustrates the propagation of the P-wave when excited at the source, forming an ellipsoidal pattern with the source as the epicenter. Notably, the figure reveals that the leading edge of the wavefront in the *Z*-axis direction precedes that in the *X*-axis direction. This observation indicates that the P-wave, traveling in the direction of excitation, exhibits a significantly higher velocity. At equivalent propagation distances, the P-wave velocity in the excitation direction far surpasses that in other orientations.



Calibration of mechanical parameters for coal samples: (A) Specimen loading method, (B) Simulated stress-strain curve and laboratory stress-strain curve.



Furthermore, it can be inferred that the S-wave generated due to P-wave excitation in the *Z*-direction experiences relatively slower propagation as they move along the *X*-direction.

Figure 9 displays a regression model we've developed to elucidate the attenuation patterns of peak vibration velocity at

various monitoring points under the influence of dynamic loads at different frequencies. The data clearly follows a power-law decay as vibration propagates over distance, which is consistent with the results of other scholars (Krylov, 2014). Upon analysis, we've observed that the attenuation coefficient decreases within the



2 Hz–30 Hz frequency range, but subsequently increases within the 30 Hz–90 Hz frequency range. Additionally, a notable phenomenon emerges as vibrations travel from P9 to P10: they experience varying levels of amplification due to the roadway's surface effect, followed by a significant drop in vibration velocity due to diffraction as they pass through the roadway.

The P9-to-P8 ratio was used as an amplification factor to quantify the amplifying effect of roadway surface on vibration velocity. Figure 10 displays peak vibration velocity and the amplification coefficient at P9 across different frequencies, revealing a strong correlation between these parameters. In the 2 Hz–20 Hz frequency range, both parameters gradually increase with frequency. Notably, between 30 Hz and 50 Hz, they experience rapid growth, but from 60 Hz to 100 Hz, this growth tapers off. At 70 Hz, P9 reaches its maximum vibration velocity at 0.436 m/s, and at 90 Hz, the amplification coefficient peaks at 2.34. These findings underscore the dynamic relationship between frequency and P9's response characteristics.

4.1.2 Propagation and attenuation of S-Wave

Similar to our research method with P-wave, the propagation of S-wave at various frequencies within coal and rock formations was systematically simulated. The key difference between S-wave and Pwave lies in their unique propagation characteristics. S-wave exhibit a perpendicular propagation direction relative to their vibrational motion, whereas P-wave propagate parallel to their vibration. We maintained consistent application positions and areas for S-wave as with P-wave, as outlined in Figure 11. Notably, in Figure 11A, the substantial variations in material properties of the propagation medium result in divergent propagation intensities for S-wave along the positive and negative *z*-axes when the source is at the interface. This phenomenon primarily arises from wave reflections induced by disparities in the physical properties of the multilayered rock formations. Similar to the phenomenon when P waves are applied, when S-wave are applied, vibrational velocity propagates outward in an elliptical pattern. In this scenario, the propagation velocity differs along the Z-axis compared to the X-axis. As depicted in Figure 11B, when S-wave are excited along the X-axis, they generate shear forces within the coal-rock mass along the z-axis, which is perpendicular to the direction of S-wave vibration. On either side of this pathway, vibrational velocity direction is opposite. Consequently, vibrational velocity in the Z-direction is notably smaller than in the X-direction. Hence, our primary focus in this study centers on vibrational velocity in the X-direction.

The power function is used to fit the attenuation rules of S waves of different frequencies when propagating in the numerical model (see Supplementary Figure S3). The plot reveals an increase in the attenuation coefficient as the frequency escalates from 2 Hz to 20 Hz, and a subsequent decrease as the frequency extends from 30 Hz to 100 Hz. This pattern is in contrast to the Pwave attenuation coefficient. Furthermore, when S-wave propagates towards P9, there is a noticeable increase in peak vibration velocity and a corresponding decrease when they reach P10. Similarly, we defined the amplification factor as the ratio of peak vibration velocity at P9 to P8. Figure 12 depicts the amplification coefficient and peak vibration velocity at P9 in response to S-wave of different frequencies. It is evident from the figure that, except for an abnormal peak at 30 Hz, both the amplification coefficient and P9's peak vibration velocity generally increase with higher frequencies. At a 100 Hz S-wave frequency, P9 registers a peak vibration velocity of 0.398 m/s, with an amplification coefficient as high as 4.69.



FIGURE 9

P-wave attenuation patterns at varying frequencies: (A) Attenuation law of P-wave with 2 Hz, (B) Attenuation law of P-wave with 5 Hz, (C) Attenuation law of P-wave with 90 Hz, (D) Attenuation law of P-wave with 100 Hz.



4.2 Dynamic load-induced plastic zone

In the Mohr-Coulomb constitutive, the plastic zone represents material failure, providing insight into the extent of material damage. This section delves into the influence of dynamic loads on the roadway by examining the plastic zone under different dynamic loading conditions.

4.2.1 P-wave induce alterations in the plastic zone

Figure 13 shows the plastic zone distribution around the source and the roadway after subjecting them to P-wave of different frequencies. The slices are taken along the middle of the model's *Y*axis. The figure demonstrates significant plastic failure at the source under dynamic loading, with an asymmetric distribution on either side of the dynamic source plane due to variations in rock material properties along the *z*-axis. The central source area experiences a combination of tensile and shear failures, while the surrounding



S-wave velocity propagation patterns in the X-Direction and Z-direction: (A) S-wave velocity propagation in the X-Direction; (B) S-wave velocity propagation in the Z-direction.



regions predominantly undergo shear failure. Around the roadway, tensile failure is localized to a specific area on the roadway surface, with the primary failure mode being shear. Interestingly, as P-wave frequency increases, the extent of failure at the source region tends to decrease. Different P-wave frequencies lead to distinct patterns of failure at the source. For P-wave frequencies of 2, 5, and 10 Hz, the failure region at the source appears continuous. However, for frequencies above 10 Hz, the failure region near the source exhibits a more fragmented distribution. This reflects the complex mechanism of dynamic loading on coal and rock materials, with varying P-wave frequencies having notably different effects on the same material.

To thoroughly examine the impact of P-wave dynamic loading at different frequencies on the plastic zones within coal and rock

materials, we've created a custom Fish program to quantify the volume and characteristics of these plastic zones, illustrated in Figure 14. It is important to note that the Fish program may not precisely identify this complex pattern when a mixed failure pattern involving both tensile and shear failure occurs. Additionally, please be aware that the plastic zone near the roadway is calculated as the increase in plastic zone size following dynamic loading, as some plastic zones emerge within the roadway after excavation. As shown in the figure, the V_{pb} initially decreases rapidly and then stabilizes as the frequency of P-wave dynamic loading increases. On the other hand, the $V_{\it pc}$ displays fluctuations, peaking at 100 Hz and reaching its minimum at 30 Hz. Notably, the V_{pb} significantly outweighs the V_{pc} , typically by a factor of 50–600. In the proximity of the roadway, Fish statistics consistently indicate shear failure. Specifically, at frequencies of 2, 5, 10, and 20 Hz, the V_{pa} remains relatively stable with minor fluctuations. However, when exposed to dynamic loading frequencies of 30 and 40 Hz, the V_{pa} rapidly increases and gradually rises with higher frequencies. The maximum V_{ba} occurs at 100 Hz. Interestingly, at a P-wave frequency of 60 Hz, the V_{pa} is smaller than it was before dynamic loading. This result may be attributed to the dynamic load causing a reduction in the difference between the maximum and minimum principal stresses around the roadway. As a consequence, the initial plastic zone may diminish or disappear.

4.2.2 S-wave induce alterations in the plastic zone

The plastic zone distribution in the vicinity of the source and around the roadway under the influence of S-wave at different frequencies was shown in Supplementary Figure S4. Shear failure remains the dominant mode of failure, with the volume of the plastic zone decreasing as the S-wave frequency increases. At 30 Hz, a



Plastic failure caused by P-wave (A) Plastic zone caused by P-wave with 2 Hz, (B) Plastic zone caused by P-wave with 20 Hz, (C) Plastic zone caused by P-wave with 50 Hz, (D) Plastic zone caused by P-wave with 100 Hz.

shear failure band extending along the *X*-axis is observed, covering the entire model. Similar to the effect of P-wave, when the S-wave frequency exceeds 10 Hz, the plastic zone at the source becomes discontinuous, indicating a reduced damage effect of high-frequency S-wave on the source.

Similarly, we utilized the Fish program to compute the volume of the plastic zone induced by the influence of S-wave at various frequencies, as depicted in Figure 15. The data highlights a gradual decrease in V_{sb} as S-wave frequency increases, aligning with what we observed with P-wave in Figure 14. However, at 30 Hz, there's an exception where the V_{sb} significantly increases, corresponding to the shear bands seen in Supplementary Figure S4. Importantly, the Fish program's calculations suggest minimal, if any, tensile failure at the source, indicating that shear failure predominantly characterizes the

influence of S-wave. The V_{sa} values in the figure reveal that the extent of roadway shear failure under dynamic loading initially decreases, then increases, and eventually shows slight fluctuations as the S-wave frequency rises. Interestingly, at 60 Hz, there's a notable dip in V_{sa} compared to other frequencies. The analysis above demonstrates the intricate and multifaceted impact of dynamic loading on the plastic zone of roadways.

4.3 Rock burst potential caused by dynamic load

Brittle shear ratio (BSR) is usually used to evaluate the rockburst potential in coal and rock mass, which is expressed as (Vennes and





Mitri, 2017):

$$BSR = \frac{\sigma_1 - \sigma_3}{UCS} \tag{6}$$

where σ_1 is maximum principle stress, σ_3 is minimum principle stress, and *UCS* is uniaxial compressive strength.

Based on the measured *UCS* of coal and rock (see Supplementary Table S1), combined with the σ_1 and σ_3 obtained through simulation, the BSR value can be determined. The BSR criteria for estimating the coal and rock mass damage and rockburst potential are shown in Table 1 (Castro et al., 2012). It can be estimated that the rock burst potential is major if the BSR value is higher than 0.7.

4.3.1 Rock burst potential caused by P-wave

Figure 16 displays the BSR cloud after roadway excavation and exposure to P-wave at different frequencies. When examining the distribution of the plastic zone and the BSR cloud in Figure 16A, we observe that the peak areas of the BSR cloud correspond to the pattern of plastic zone distribution, notably in the four corners of the roadway. However, the BSR values for the top and bottom plates are consistently below 0.4, indicating a minimal rockburst potential. The lower rockburst potential in the floor is due to its high uniaxial compressive strength, while the top coal experiences reduced risk due to vertical unloading after excavation, leading to a significant

decrease in maximum and minimum principal stress difference. The figure demonstrates that P-wave at varying frequencies have a limited impact on the BSR distribution, but different frequencies result in varying degrees of rockburst potential, as reflected in the maximum BSR values.

Figure 17 shows the BSR maxima at different P-wave frequencies. For P-wave frequencies of 5 Hz, 10 Hz, and 20 Hz, the BSR peaks are significantly higher compared to other frequencies. At a dynamic load frequency of 10Hz, the BSR peak reaches 2.61, reflecting a 0.09 increase from the roadway excavation condition. Based on the analysis above, it is clear that the rock burst potential resulting from P-wave in the frequency range of 5 Hz–20 Hz is significantly higher than that induced by P-wave of other frequencies.

4.3.2 Rock burst potential caused by S-wave

The BSR cloud is presented after the application of Swave at different frequencies (see Supplementary Figure S5). The observation suggests that the distribution of BSR following S-wave action closely resembles that after P-wave action. However, a notable distinction lies in the fact that the peak value of BSR is significantly higher following S-wave action compared to P-wave action.

Figure 18 presents the maximum BSR values resulting from the application of S-wave at various frequencies. The chart reveals that BSR peaks are notably higher at S-wave frequencies of 2 Hz, 5 Hz, and 10 Hz compared to other frequencies. At 60 Hz, the peak BSR value reaches 2.67, representing a 0.15 increase from the pre-dynamic load condition. The figure also shows that as Swave frequencies increase, the peak BSR value experiences an initial sharp decrease, followed by relatively stable minor fluctuations. This pronounced decline in peak values is observed at S-wave frequencies of 2 Hz, 5 Hz, and 10 Hz. Interestingly, the BSR peak reaches its highest value at 60 Hz, indicating a significant impact on roadway rockburst potential at this frequency.

5 Discussions

5.1 Propagation characteristics of P and S-wave

Figure 19 provides a comprehensive synthesis of previous section findings, allowing for a comparative analysis of P-wave and S-wave propagation characteristics. As shown in Figure 19A, the attenuation coefficient of P-wave initially decreases, with subsequent minor fluctuations, as the dynamic load frequency increases. The maximum attenuation coefficient values are observed at dynamic load frequencies of 2 Hz and 5 Hz. Conversely, S-wave attenuation coefficients show an initial increase followed by a decrease, stabilizing at higher frequencies, with peak values at 10 Hz and 20 Hz. Notably, S-wave attenuation coefficients are consistently lower than those of P-wave. This conclusion is consistent with that obtained by Torres using seismic observation data in the northern Gulf of California and Mexico (Bischoff et al., 2010).

As shown in Figure 19B, with the increase of frequency the peak vibration velocity induced by P-wave experiences initial minor fluctuations, followed by a rapid increase, and ultimately stabilizes at a high level. Conversely, the pattern for peak vibration velocity

BSR	Rock mass damage	Potential for rockburst
0.35	No to minor	No
0.35-0.45	Minor (e.g., surface spalling)	No
0.45-0.6	Moderate (e.g., breakout formation)	Minor
0.6–0.7	Moderate to major	Moderate
>0.7	Major	Major

TABLE 1 Criteria of BSR for rock mass damage and potential of rockburst evaluation.



BSR cloud: Roadway excavation vs Different P-wave loadings: (A) Plastic zone distribution and BSR cloud of the roadway before dynamic load, (B) BSR cloud of the roadway after P-wave with frequency 10 Hz, (D) BSR cloud of the roadway after P-wave with frequency 10 Hz.

caused by S-wave is simpler, showing a consistent increase with rising frequency. However, there is a distinctive spike in peak vibration velocity at 30 Hz. This may be due to the resonance of the roadway caused by the S-wave at 30 Hz, resulting in severe plastic failure of the roadway. This is consistent with the results of Liu et al., who discovered that the resonant frequency of the roadway in Zhuji Coal Mine is 35 Hz (Liu et al., 2019). The trend for the amplification

coefficient, both for P-wave and S-wave, closely follows that of the peak vibration velocity. Notably, beyond a frequency of 50 Hz, the amplification coefficient for S-wave significantly surpasses that of P-wave, highlighting a growing disparity between them with increasing frequency. This is consistent with the law of the plastic zone volume in Figure 21 below, where the total plastic zone volume caused by the S-wave is higher than that of the P-wave.





The 30 Hz S-wave exhibits distinct characteristics that merit further analysis, as illustrated in Figure 19B. Figure 20A clearly shows that the 30 Hz S-wave induces a higher vibration velocity at the P9 compared to other frequencies. Additionally, we have delineated the apparent fluctuation segment of each velocity curve, indicated by the light yellow area in Figure 20A. This region is determined based on vibration velocities higher than 10% of the peak vibration velocity. Notably, the extent of the pale yellow region reveals that the duration of the S-wave gradually decreases as the frequency increases, consistent with the general principle that higher frequencies decay more rapidly. However, the 30 Hz frequency exhibits an anomaly, with a significantly shorter duration compared to 40 Hz and 50 Hz. To gain further insights, Fast Fourier Transform (FFT) was conducted to analysis of the waveform within the yellow region, yielding Figure 20B. The dominant frequency of the vibration velocity at the P9 generally increases with the input frequency of the S-wave, except 30 Hz. At 30 Hz, there are four distinct peak frequencies, indicating a highly complex waveform composition compared to other frequencies. This complexity can be attributed to the fact that, as shown in Figure 15, the 30 Hz S-wave led to a substantial expansion of the roadway's plastic zone. In essence, when the 30 Hz S-wave propagates through the roadway, it induces the most severe plastic failure, resulting in a complex waveform composition.

5.2 Dynamic load-induced plastic zone changes

Figure 14 reveals an anomaly in V_{pa} when the roadway is exposed to P-wave activity at 60 Hz. To uncover the underlying cause of this peculiarity, we conducted a comprehensive analysis, which included detailed statistics of the plastic zones on both sides of the roadway, as well as those on the roof and floor. As depicted in Figure 21, we applied the Fish to compute the plastic zone volumes on both sides of the roadway, as well as those on the roof and floor. In Figure 21A, both \boldsymbol{V}_{pm} and \boldsymbol{V}_{pn} are negative at 60 Hz, and V_{pm} is negative at 2, 5, and 10 Hz. These negative values suggest a reduction in the plastic zone size after dynamic loading, which is quite perplexing. This phenomenon can be attributed to the stress adjustment in the zone due to dynamic loading, so that the stress conditions that would have allowed the zone to undergo plastic failure no longer exist. In addition to these anomalies, V pm gradually increases with the increase in P-wave load frequency, while the increase in V_{pn} is less pronounced. Figure 21B indicates that V_{sm} peaks after the impact of the 30 Hz S-wave. However, when the Swave frequency is 60 Hz, V_{sm} is significantly lower than at other frequencies, and V_{sn} is also smaller compared to other frequencies. The increase in the plastic zone induced by P-wave activity is notably less compared to that caused by S-wave activity. This discrepancy arises from the alignment of the X-axis direction of the maximum principal stress in the numerical model with the primary action direction of S-wave. In other words, S-wave enhances the maximum principal stress, while P-wave enhances the minimum principal stress. Consequently, the impact of these two waveforms on the plastic zone is distinct.

The abnormal plastic zone of the roadway induced by the 60 Hz P-wave prompted a spectrum analysis of the time-velocity curve at the P9 measurement point. This analysis is presented in Figure 22, following the same data selection principles as before. The analysis of Figure 22A reveals that the significant fluctuation time of the waveform lacks a clear pattern. The peak vibration speed at P9 tends to increase with the rising frequency, except at 70 Hz, where it decreases. In Figure 22B, the main frequency components of the waveforms at 60 Hz and 80 Hz display a distinct single peak, while those of other waveforms are more intricate, often featuring double or triple peaks. The complexity of the main frequency in the waveforms is attributed to the damage of the surrounding rock when the waveform reaches P9, leading to the generation of additional wave. Figure 14 shows that V_{pa} at 60 Hz and 80 Hz is notably lower compared to the surrounding frequencies, suggesting a potential correlation. It seems that when the plastic failure around the roadway becomes more severe, the detected waveform's main frequency also becomes more intricate.



Propagation characteristics of P and S-wave at varying frequencies: (A) Attenuation coefficients of P and S-wave; (B) roadway peak vibration velocity and amplification coefficient from P and S-wave.



P9 vibration velocity fluctuation curves and FFT analysis for S-wave of different frequencies: (A) Velocity time history curve of P9 measuring point under the action of S-waves of different frequencies, (B) FFT results of the velocity time history curve of each measuring point.

5.3 Change in rock burst potential of roadway after dynamic loading

Figure 23 illustrates the maximum BSR of the roadway when subjected to S-wave and P-wave of varying frequencies. The figure demonstrates a noticeable increase in roadway rockburst potential due to dynamic loading, compared to the roadway excavation. Except for the frequencies of 10 Hz and 20Hz, the BSR induced by S-wave at other frequencies surpasses that caused by P-wave of the same frequency. This discrepancy is attributed to the primary loading direction of S-wave aligning with the maximum principal stress direction. The increase in rockburst potential associated with P-wave is relatively minor, except for frequencies of 5 Hz, 10 Hz, and 20 Hz. Interestingly, the BSR due to S-wave generally decreases with increasing frequency, except for an anomaly at 60 Hz where the BSR reaches its peak, significantly exceeding that at other adjacent frequencies. Drawing insights from Figure 21A, it becomes evident that the plastic zones induced in both the roof and floor reach their minimum extent when subjected to P-wave frequencies of 5 Hz, 10 Hz, and 20 Hz. This observation underscores a noteworthy discrepancy between the findings derived from the analysis of plastic zones and those from the BSR index. This incongruity can be attributed to that when the surrounding rock undergoes plastic deformation, its stress levels tend to decrease. Similarly, as indicated by Figure 21B, when the frequency of S-wave is 60 Hz, the increment in the plastic zones observed in





the sidewalls and the top and bottom plates of the roadway is significantly less compared to the surrounding frequencies. This suggests that the 60 Hz S-wave dynamic load induces plastic deformation in a more confined region. It explains why this frequency yields the highest rockburst potential. This leads us to a plausible conclusion: the 60 Hz S-wave dynamic load elevates the maximum principal stress but does not quite reach the threshold for plastic failure in the material, resulting in a higher calculated BSR value.

6 Conclusions and prospects

6.1 Conclusions

Based on the engineering background of the Dongtan Coal mine, this study compared wave characteristics between nearfield and far-field strong mine tremors. Additionally, investigates the propagation and attenuation characteristics of P-wave and Swave at various frequencies, assessing their impact on the plastic



failure and the rock burst risk of roadways. The main conclusions are as follows:

- (1) Wave analysis reveals that compared with near-field strong mine tremors, far-field tremors are dominated by lowfrequency components below 5 Hz, which may caused by the rapid attenuation of high-frequency components along with propagation.
- (2) As the frequency increases, the P-wave amplification coefficient initially stabilizes but then sharply rises, with fluctuations at higher frequencies. S-wave generally amplifies with frequency, except for an unusual spike at 30 Hz, related to the serious plastic failure of the roadway caused by dynamic load resonating with the roadway.
- (3) After P-wave and S-wave loading, shear failure is dominant at the tremor source. The proportion of source shear failure after S-wave loading is higher than that after P-wave, and the roadway plastic zone caused by S-wave loading is larger than that caused by P-wave.
- (4) The rock burst potential increases significantly after P-wave and S-wave loading, and the rock burst potential after the S-wave loading is higher than that of P-wave.

6.2 Prospects

In recent years, the frequency of far-field strong mine earthquakes has been on the rise, making the study of their impact on roadways a crucial research area. Due to the challenges in conducting accurate field or theoretical research on this issue, numerical simulation methods have become indispensable. The conclusions drawn in this study are based on the modeling of the actual engineering background conditions of Dongtan Coal Mine. It can be predicted that the simulated dynamic load propagation law and its influence on the roadway may be quite different under different conditions due to the differences in the strata structure, the properties of coal and rock mass, and the actual stress state of the roadway. Therefore, it is an effective and reliable means to establish an accurate numerical model for specific engineering research. At present, sensors for vibration velocity monitoring in engineering are arranged in the form of points, so a small amount of data collected by sensors makes it difficult to range the real situation of the site. If large-scale vibration velocity monitoring can be realized in the future, field data-based research will flourish.

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 author.

Author contributions

ZH: Conceptualization, Investigation, Methodology, Writing-original draft. LD: Funding acquisition, Methodology, Writing-review and editing. SG: Funding acquisition, Writing-review and editing. JC: Formal Analysis, Writing-original draft. SC: Methodology, Writing-original draft. BH: Resources, Validation, Writing-review and editing. YL: Resources, Writing-review and editing. KZ: Resources, Writing-review and editing.

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Conflict of interest

Authors BH and YL were employed by Huaneng Coal Technology Research Co., Ltd.

The remaining authors declare that the research was conducted in the absence of any commercial or financial relationships that could be construed as a potential conflict of interest.

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Supplementary material

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