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Measurements of seismic waves induced by highvelocity impacts: Implications for seismic shaking surrounding impact craters on asteroids

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Measurements of seismic waves induced by high-velocity impacts: Implications for 1 2 seismic shaking surrounding impact craters on asteroids 3 Kazuma Matsue¹, Minami Yasui¹, Masahiko Arakawa¹, and Sunao Hasegawa² 4 5 6 1. Graduate School of Science, Kobe University 7 1-1, Rokkkodai-cho, Nada-ku, Kobe, Hyogo 657-8501, Japan 8 2. Japan Aerospace Exploration Agency, Institute of Space and Astronautical Science 9 3-1-1, Yoshinodai-cho, Chuo-ku, Sagamihara, Kanagawa 252-5210, Japan 10 11 Tel: +81-78-803-5741 12 Fax: +81-78-803-5741 13 E-mail: minami.yasui@pearl.kobe-u.ac.jp 14 15 Manuscript pages: 53 16 Number of figures: 15

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Abstract

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On asteroids, impact-induced seismic shaking is an important geological process related to the modifications of the surface features that are due to the movement of regolith grains on the asteroid surface. Mass movements caused by seismic shaking (e.g., landslides and crater erasures) were recently observed by several spacecrafts. To elucidate the seismic shaking areas induced by impacts on asteroids, it is necessary to determine the excitation and the decay processes of impact-induced seismic waves. Here we conducted impact cratering experiments using 500-um quartz sand to simulate an asteroid surface at the Institute of Space and Astronautical Science (ISAS), Japan Aerospace Exploration Agency (JAXA). We used 4.75-mm- and 2-mm-dia. projectiles with density from 14.9 to 1.2 g cm⁻³; the impact velocity ranged from 7 to 0.2 km s⁻¹. We measured the acceleration waveform by using three to five accelerometers setting on the target surface, and we examined three parameters characterizing the impact-induced seismic wave with varying distances from the impact point: the propagation velocity (V_{prop}) , the maximum acceleration (g_{max}) , and the half period of the first upward peak (T_{half}) . The V_{prop} was obtained to be 52.4 ± 7.2 m s⁻¹, regardless of the impact velocity and the projectile properties, and the $T_{\rm half}$ slightly depended on the impact velocity as $v_{\rm i}^{0.14}$ but was independent of the projectile properties. The g_{max} had a close relationship with the distance from the impact point normalized by the crater rim radius, x/R_{rim} , regardless of the impact velocity and the projectile properties. The obtained empirical equation was

- shown as $g_{\text{max}} = 10^{2.21} (x/R_{\text{rim}})^{-3.18}$. By using the empirical equations we obtained
- 49 herein, we observed that the seismic energy normalized by the kinetic energy of a
- projectile, E_s/E_k , decreased with the increase of the normalized distance, x/R_{rim} . We
- also estimated the area of impact-induced seismic shaking on an idealized small (500 m
- 52 in diameter) body at the impact velocity of 5 km s^{-1} .
- **Keywords:** Asteroids; Asteroids, surfaces; Experimental techniques; Cratering; Impact
- 54 processes

1. Introduction

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56 1.1. Seismic shaking on an asteroid surface

Seismic shaking induced by a high-velocity impact is one of the most important 57 58 surface geologic processes that occurs on an asteroid surface. Such shaking frequently 59 modifies the asteroid surface topography. Several planetary explorations recently revealed the detailed surface geology of asteroids, and evidence of regolith migration was 60 identified on the asteroid surfaces. For example, NEAR-Shoemaker and Hayabusa 61 62 spacecrafts obtained numerous close-up images of the asteroids 433 Eros and 25143 63 Itokawa. The images of the asteroid surfaces showed unconsolidated gravel and pebbles which were typically stacked upon each other [Miyamoto et al., 2007]. Regolith migration 64 65 was then identified on the asteroid surface as alignments of boulders, landslides, and 66 crater erasures [Veverka et al., 2001; Cheng et al., 2002; Thomas et al., 2002; Robinson 67 et al., 2002; Miyamoto et al., 2007]. 68 These features could be explained by global-scale and/or local-scale seismic shaking caused by high-velocity impacts on the regolith surface. Impact-induced seismic shaking 69 70 on a global scale could trigger regolith migration under a vacuum and in the 71 microgravitational environment that is typically present on small asteroids [Miyamoto et 72 al., 2007]. These phenomena could preferentially remove small undulations such as small 73 craters.

Asteroid 433 Eros, one of the most intensively studied small extraterrestrial bodies,

was revealed to have surface areas at which the crater number density differed considerably. It also appeared that the number density of small impact craters (<~100 m in diameter) was much smaller than that extrapolated from the >100-m impact craters, based on the cumulative size distribution of impact craters detected on the asteroid [Robinson et al., 2002]. To study this deficiency in the number of small craters, Richardson et al. [2005] developed a theoretical model for the crater erasure process caused by global seismic shaking. They stated that their calculations indicated that the vibration of the regolith layer covering the entire asteroid surface was responsible for erasing the craters. A deficiency in the number of craters with diameters <500 m was observed around Shoemaker crater (within 9 km from the center of the crater), and this unique feature of low crater density suggested that the crater degradation could have been caused by local seismic shaking induced by the formation of Shoemaker crater [Robinson et al., 2002]. Several research groups have carried out numerical simulations to investigate the effect of impact-induced seismic shaking on the motion of particles in the regolith layer of an asteroid surface. Asphaug and Melosh [1993] performed a numerical simulation using a hydrodynamic code for a high-velocity impact, and they examined the seismic resurfacing of the asteroid surface. Greenberg et al. [1994, 1996] proposed a 'global jolt' resulting from large impacts on the surfaces of an asteroid such as 243 Ida and 951 Gaspra, and they discussed the effect of the jolt on the crater size distributions. The results of both

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the Asphaug and Melosh and Greenberg et al. studies suggested that there is a key parameter that induces the resurfacing that is related to impact craters on asteroids, that is, an impact-induced seismic efficiency factor that could be defined by the ratio of the seismic energy to the kinetic energy of the impactor.

Such an impact-induced seismic efficiency factor is poorly understood, and only a few impact experiments have been conducted to evaluate impact-induced seismic waves in a laboratory setting. McGarr et al. [1969] conducted high-velocity impact experiments using loose sand and epoxy-bonded sand as the targets and a Lexan projectile at impact velocities of 7–0.8 km s⁻¹. They observed the impact-induced seismic waves by setting accelerometers on the target surface. Their results demonstrated an impact-induced seismic efficiency factor of 6.0×10^{-6} for the loose sand and 7.6×10^{-5} for the bonded sand target.

Yasui et al. [2015] conducted impact experiments on targets made of 200- μ m-dia. glass beads impacted at the impact velocities of <150 m s⁻¹ to study the attenuation rate of the impact-induced seismic wave. They observed that the maximum acceleration, i.e., g_{max} , of the seismic wave decayed with the distance from the impact point normalized by the crater radius, x/R, according to the following power law equation: $g_{\text{max}} = g_0(x/R)^{-2.2}$ within x/R = 5. Their calculations revealed that the average impact-induced seismic efficiency factor was 5.7×10^{-5} . However, this result was obtained at impact velocities <150 m s⁻¹, which is much slower than the relative collisional velocities

among asteroids. Moreover, spherical glass beads were used as the target, and these beads might be an idealized granular material simulating asteroid regolith. The seismic efficiency factor had a wide range $(10^{-2}-10^{-6})$ depending on the impactor and target properties [Gault and Heitowit, 1963; McGarr et al., 1969; Collins et al., 2005] and the impact energies [Shishkin, 2007].

More realistic numerical simulations were recently conducted by using iSALE to reproduce an impact crater formed on cohesive materials such as porous basalt, quartzite, and sandstone. The estimated seismic efficiency factor ranged from 10⁻³ to 10⁻⁵ for the elastic waves that were induced by the impact at a site far from the impact point [Güldemeister and Wünnemann, 2017; Wójcicka et al., 2019].

We conducted the present study to determine the attenuation rate of a seismic wave propagating through a target simulating an asteroid regolith, and we investigated the impact-induced seismic efficiency factor at high impact velocities (i.e., those comparable to the relative collisional velocities among asteroids). We then carried out impact cratering experiments and measured the impact-induced seismic waves by using accelerometers set on a granular target, following the methods described by McGarr et al. [1996] and Yasui et al. [2015]. For our investigation of the effects of the impact velocity and the projectile properties on the impact-induced seismic waves, the target material (quartz sand) was fixed. We also estimated the impact-induced seismic energy by using a revised sinusoidal wave model delineated by Yasui et al. [2015], and we discuss the

impact-induced seismic efficiency factor. We also assessed the seismic shaking area on an idealized small body by using the obtained empirical equations obtained herein and the crater scaling laws.

1.2. π -scaling theory

Crater scaling theories are essential to evaluations of planetary-scale impact phenomena under various impact conditions. These theories have been tested in laboratory experiments. In the present study, we applied the π -scaling theory proposed by Holsapple [1993] to our experimental results. This conventional theory for cratering processes can be used to extrapolate our laboratory-scale results to planetary-scale impacts. We used the following non-dimensional scaling parameters, i.e., π_R , which is related to the crater radius (R_c) , and π_2 and π_4 , which are related to various impact conditions:

$$\pi_{\rm R} = R_{\rm c} \left(\frac{\rho_{\rm t}}{m_{\rm p}}\right)^{1/3},\tag{1}$$

$$\pi_2 = \frac{gr_p}{v_i^2},\tag{2}$$

$$\pi_4 = \frac{\rho_t}{\rho_p},\tag{3}$$

where $\rho_{\rm t}$ is the bulk density of target, $m_{\rm p}$ is the projectile mass, $\rho_{\rm p}$ is the projectile density, $r_{\rm p}$ is the projectile radius (they are written as $m_{\rm p}=4\pi\rho_{\rm p}r_{\rm p}^3/3$ for a spherical

projectile), g is the gravitational acceleration, and v_i is the impact velocity.

In a gravity regime in which the crater formation process is controlled by the gravity of the body, the following relationship is proposed on the basis of a point source approximation and a coupling parameter, $C = r_p v_i^{\mu} \rho_p^{\nu}$:

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$$\pi_{R} = H_{1} \pi_{2}^{-\frac{\mu}{2+\mu}} \pi_{4}^{\frac{2+\mu-6\nu}{3(2+\mu)}}, \tag{4}$$

where H_1 , μ , and ν are the constants that depend on the target properties, We were able to determine these values in laboratory experiments.

2. Experimental methods

162 2.1. Impact experiments

We used quartz sand with a mean grain diameter of 500 μ m as a target simulating the regolith layer on an asteroid surface. The grains of the quartz sand have irregular shapes but rounded corners. The measured angle of repose was $33 \pm 2^{\circ}$, which is much higher than that of the 200- μ m spherical glass beads ($23 \pm 2^{\circ}$) used by Yasui et al. [2015]. The grain density of the quartz sand was 2.7 g cm⁻³ and the calculated bulk density of the quartz-sand target was 1.46 g cm⁻³; thus, the estimated bulk porosity was approximately 42%. The quartz sand was poured into two stainless-steel containers: the larger 60-cm-diameter, 20-cm-deep container was used for the impact velocities >1.1 km s⁻¹, and the smaller 30-cm-diameter, 10-cm-deep container was used for the impact velocities <230

m s⁻¹. Before every shots, the surface of the target was flattened with the use of a hard metal ruler.

We conducted the impact experiments by using a two-stage vertical gas gun located at the Institute of Space and Astronautical Science (ISAS), Japan Aerospace Exploration Agency (JAXA). Fig. 1a is a schematic of the experimental setup. The 2-m-high target chamber had a 1.5-m diameter and was set below the gun barrel: the chamber was evacuated below 10 Pa before each shot. The impact velocity, v_i , ranged from 6.9 to 1.1 km s⁻¹, and we used eight types of projectile with densities ranging from 14.9 to 1.2 g cm⁻³ to study the effects of the projectile density on the crater scaling law and the impact-induced seismic waves.

Each of the projectiles was a sphere, made of the following: polycarbonate (PC, density of 1.2 g cm⁻³), soda-lime glass (Gl, 2.5 g cm⁻³), aluminum (Al, 2.7 g cm⁻³), titanium (Ti, 4.6 g cm⁻³), zirconia (ZrO₂, 5.7 g cm⁻³), stainless steel (SUS, 7.9 g cm⁻³), copper (Cu, 9.0 g cm⁻³), and tungsten carbide (WC, 14.9 g cm⁻³). The diameter of the PC projectile was 4.75 mm, and the diameter of the other seven projectiles was 2.0 mm. Each projectile was launched in the direction parallel to the gravity acceleration of the Earth, so that it impacted at a direction perpendicular to the target surface.

To examine the effects of the grain shape on the cratering process, we also conducted low-velocity cratering experiments using the quartz sand at the impact velocity of 170–227 m s⁻¹ in order to compare our results with those of the Yasui et al. [2015] study. It is

Fig. 1

difficult to achieve an impact velocity below 1.0 km s⁻¹ by using the two-stage gas gun at ISAS, and thus our low-velocity cratering experiments were carried out using the same experimental set up as that used by Yasui et al. [2015].

2.2. Observations of impact phenomena

We used an uniaxial piezoelectric accelerometer (#SV1113, Nippon Avionics, Tokyo) with a charge amplifier (#AG2101, Nippon Avionics) to measure the acceleration activated on the target surface affected by an impact-induced seismic wave. The charge sensibility was 5.47 pC s² m⁻¹, and the response frequency of the acceleration was 0.5 Hz–10 kHz. For our investigation of the decay process of impact-induced seismic waves, we set three, four, or five accelerometers (Fig. 1b) on the target surface at 35.3 to 5.1 cm from the impact point, outside of the impact crater rim. The measurement direction of the accelerometers were perpendicular to the target surface; the accelerometers could therefore detect the normal components of the acceleration activated on the surface. The accelerometers were buried at the depth of 2.5 cm (Fig. 1c).

Each seismic wave measured by the accelerometers was recorded by a data logger with an A/D conversion rate of 100 kHz (midi LOGGER GL900, Graphtec, Yokohama, Japan). We repeated the cratering experiments using the same impact conditions several times to acquire the seismic wave data at different propagation distances, ensuring that we would have a sufficient amount of data to determine the attenuation rate of the seismic

waves.

It is necessary to remove any artificial noise added to the actual signals, because such noises are naturally induced by the instantaneous force generated by the projectile launching. This force could activate the vibration of the target container set in the target chamber. The bottom of the target container was thus insulated by a rubber sheet to remove the activated vibration (Fig. 1a) so that the acceleration of each impact-induced seismic wave was measured precisely.

We observed the impact phenomena by using a high-speed digital camera (Hyper Vision HPV-X, Shimadzu, Kyoto, Japan) to determine the impact time of each projectile on the target surface. The frame rate was set at 10⁵ frames s⁻¹. The impact time could be determined within 10 μs. To synchronize the trigger of the high-speed digital camera with the trigger of the data logger, we generated a trigger signal by using a function generator to send the signal to both devices at the moment when the velocity measurement system detected a signal showing the projectile passing through.

After the shot, a photograph was taken just above the target surface for the analysis of the precise distance of each accelerometer from the impact point. We used a digital caliper to measure the rim-to-rim distances in two directions where they were crossed at right angles. The average distance was defined as the rim diameter, D_{rim} . In several shots, we measured a cross-section of the impact crater by using a laser profiler (#LJ–V7300, Keyence, Osaka, Japan) with 10- μ m resolution to determine the ratio of the depth to the

diameter of the impact crater.

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3. Results

3.1. Morphology of impact craters

The experimental conditions and the results on the morphology of the impact craters are summarized in Table 1. Fig. 2 shows the crater profiles obtained under various impact conditions. As shown in the figure, all of the impact craters formed in these experiments were categorized as a simple crater with a bowl-shaped depression surrounded by a distinct rim. A bright area was spread over the center of the crater floor in each case; this area might have been comminuted quartz sand grains.

The impact crater formed by the PC projectile is shown in Fig. 2a, b. The crater size at $v_i = 6.9 \, \mathrm{km \, s^{-1}}$ was about twice that obtained at 1.1 km s⁻¹. When an impact velocity was high (Fig. 2b), several small pieces of black debris were observed on the bright area of the crater floor; these might have been charred broken projectile fragments. Impact craters formed by an Al projectile and a Cu projectile are shown in Fig. 2c, d, and the impact velocity was almost the same: approximately 4 km s⁻¹. The crater size increased with the increase of the projectile density: the crater made by the Cu projectile was 1.3 times larger than that by the Al projectile at the density ratio of Cu to Al of approximately

Fig. 3a is crater profiles obtained with the PC projectiles at different impact velocities.

Fig. 3

Table 1

Fig. 2

Both the diameter and the depth of the craters clearly became larger as the impact velocity increased. To study the similarity of the crater shapes formed at different impact velocities, we normalized both the vertical and horizontal axes in Fig. 3a by each crater rim radius, $R_{\rm rim}$, where $R_{\rm rim} = D_{\rm rim}/2$, and then we plotted the normalized crater profile at each impact velocity (Fig. 3b). The normalized crater profiles were very similar, regardless of the impact velocity. Thus, the similarity of the crater shapes was completely confirmed. Fig. 3c shows the crater profiles obtained at the impact velocity of $v_i = 4.0 \text{ km s}^{-1}$ for the projectiles with densities ranging from 14.9 to 1.2 g cm⁻³. Both the diameter and the depth of the craters became larger as the projectile density increased — except for the PC projectile with its different diameter. Fig. 3c has been re-scaled by using the $R_{\rm rim}$ values in order to show the normalized crater profiles in Fig. 3d. The normalized crater profiles were very similar around only the crater rim, regardless of the projectile density, but the normalized crater depth was deeper as the projectile density increased. We analyzed the normalized crater profiles to determine the following parameters (Fig. 1c): the crater rim diameter (D_{rim}) , the crater diameter measured on the pre-shot surface (D_c) , the crater depth from the rim peak to the crater bottom (d_{rim}) , and the crater depth from the pre-shot surface to the crater bottom (d_c) . All of these parameters are explained in Fig. 1c. The calculated ratio of the crater rim diameter to the crater diameter, $D_{\text{rim}}/D_{\text{c}}$, was calculated to be 1.26 ± 0.04 , regardless of the impact conditions. The ratio of the

crater depth to the crater diameter, d_c/D_c , is shown in Fig. 4. As mentioned above, the

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 $d_{\rm c}/D_{\rm c}$ increased with the increase of the projectile density, but it did not systematically depend on the impact velocity for each projectile. This $d_{\rm c}/D_{\rm c}$ increase with the projectile density could have been caused by the penetration of the projectile into the target, because a high-density projectile penetrates deeper due to its greater momentum.

We were able to determine the precise crater diameter, $D_{\rm c}$, by using the crater profiles obtained by the laser profiler, but the crater profile was not measured in all shots. The crater rim diameter, $D_{\rm rim}$, was measured in all shots instead of the $D_{\rm c}$. We thus used the following equation, $D_{\rm c} = D_{\rm rim}/1.26$, to construct the crater size scaling law. The experimental conditions and the crater size results are summarized in Table 2.

Table 2

Fig. 5

To study the relationship between π_4 and π_R in Eq. (4), we plotted the data for the same projectile diameter at a constant impact velocity (i.e., a constant π_2) in Fig. 5a. This figure includes the data for seven projectiles (the exception is the PC projectile) with different densities at the impact velocity of 2 and 4 km s⁻¹. The π_R values increased slightly with the increase of π_4 at each impact velocity. This relationship can be described as follows:

$$\pi_{R} = 10^{a} \cdot \pi_{4}^{b}. \tag{5}$$

The coefficients of a and b were $a = 1.24 \pm 0.01$ and $b = 0.047 \pm 0.019$ for $v_i = 2$ km s⁻¹, and $a = 1.34 \pm 0.01$ and $b = 0.042 \pm 0.017$ for $v_i = 4$ km s⁻¹. The coefficient b was independent of the impact velocity, and its calculated average was 0.044. Therefore, the

power law index of π_2 in Eq. (4) can be determined from the relationship between π_R .

292 π_4^{-b} and π_2 in Fig. 5b as follows:

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$$\pi_{R} \cdot \pi_{4}^{-b} = 10^{-0.265 \pm 0.014} \pi_{2}^{-0.176 \pm 0.002}$$
. (6)

The power law indices of the coupling parameters, μ and ν , in Eq. (4) were derived from b and the power law index of π_2 , and their values were μ =0.43 and ν =0.35. The obtained μ was almost the same as that summarized by Housen and Holsapple [2011] for sand targets based on the studies of Stöffler et al. [1975] and Cintala et al. [1999]. Housen and Holsapple [2003] suggested that the typical value of ν could be assumed to be 1/3. Our present finding is quite similar to this value.

3.2. Impact-induced seismic waves

We classified the impact-induced seismic waves observed in this study into two categories depending on the distance from the impact point: (1) a single pulse-like wave (Fig. 6a, b), and (2) a damped oscillation wave (Fig. 6c, d). The single pulse-like wave usually appeared near the crater rim. Fig. 6a, b provides an example of a single pulse-like wave obtained at the impact velocity of 5.2 km s⁻¹ (run #140530-3). The acceleration history in the range of time up to 100 ms is illustrated in Fig. 6a, and Fig. 6b shows the history up to 15 ms. The time of '0 ms' corresponds to the time at which the projectile impacted the target surface, which we determined from the high-speed digital camera

Fig. 6

observation.

A complex wave with a small amplitude and a short wavelength was typically observed after 50 ms from the impact (Fig. 6a). Our high-speed digital camera observations revealed that many ejected grains from the crater dropped on the target surface at ~ 50 ms after the impact. This complex wave could therefore also be generated by the impact of grains dropped on the accelerometer. We then limited the seismic waves to those that occurred during the first 30 ms for the analysis of these waveforms. The single pulse-like wave was approximated by a damped oscillation wave with a large damping rate, and the wave can thus be described as one cycle of a damped sinusoidal curve with a frequency of several hundred Hz. The amplitude of the acceleration attenuated with the propagation distance.

These features were very similar to those reported for an unconsolidated loose sand target [McGarr et al., 1969] and those observed with a target of glass beads [Yasui et al., 2015]. The damped oscillation wave usually appeared far from the crater rim. Fig. 6c, d provide an example of the damped oscillation obtained at the impact velocity of 4.1 km s⁻¹ (run #150309–1); Fig. 6c shows the acceleration history in the range of time up to 200 ms, and Fig. 6d shows that up to 20 ms. A complex wave with a small amplitude and a short wavelength was also observed after 100 ms of the impact (Fig. 6d). This complex wave could also be generated by the impact of grains that fell on the accelerometer. The distance between the accelerometer and the crater rim shown in Fig. 6c, d was greater

than the x = 15 cm of the accelerometer (Fig. 6a, b). It thus appears that the complex wave was generated by grains that fell at a later time. The seismic wave observed far from the crater rim is also approximated by a damped oscillation wave with a smaller damping rate as shown in Fig. 6c, d; the wave typically oscillated several times.

We analyzed these observed seismic waves to obtain three parameters characterizing them as expressed in Fig. 6e: the traveling time ($t_{\rm max}$), the maximum acceleration ($g_{\rm max}$), and the duration of the first upward acceleration ($T_{\rm half}$). We used these parameters to determine the decay rate of the acceleration during the wave propagation. The experimental conditions and results including the data of three parameters are summarized in Tables 3–5. Table 3 concerns the PC projectile at the impact velocity of $\sim 200~{\rm m~s^{-1}}$, Table 4 is about the PC projectile at impact velocities $> 1~{\rm km~s^{-1}}$, and Table 5

Table 3

Table 4

Table 5

is the results for 2-mm projectiles with the projectile density ranging from 14.9 to 2.5 g

 cm^{-3} at impact velocities >1 km s⁻¹.

3.3. Propagation velocity of impact-induced seismic waves

The traveling time of the impact-induced seismic wave $(t_{\rm max})$ is defined as the duration between the impact time and the time point at which the maximum acceleration of the first upward peak is observed. We used $t_{\rm max}$ to determine the propagation velocity of each impact-induced seismic wave on the target. Fig. 7a illustrates the relationship

Fig. 7

between the traveling time t_{max} and the propagation distance, x, for a PC projectile at

the x from 5 cm to 30 cm at different impact velocities. The average crater rim radius at each impact velocity is also shown on this figure as a horizontal line and in Table 6.

350

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Table 6

The t_{max} value increases linearly with the increase of the x outside the crater rim 352 at each impact velocity, and thus the propagation velocity (V_{prop}) of the impact-induced 353 seismic waves could be constant outside the crater rim. We therefore calculated the V_{prop} 354 from the slope of this linear relationship, and each dataset corresponding to each impact 355 356 velocity was fitted by the linear equation. The V_{prop} obtained at each impact velocity is 357 shown in Table 6. Regardless of the impact velocity in the range from 7 to 0.2 km s⁻¹, the V_{prop} values were almost constant within the error. The average V_{prop} was 51.1 ± 8.9 m 358 s^{-1} . The traveling time, t_{max} , should be zero at the impact point of x = 0, but we 359 observed that all of the fitting lines had offsets at the intersection on the vertical axis at 360 361 t = 0 (Fig. 7a). The offset was larger as the impact velocity was higher. This offset was 362 also reported by Yasui et al. [2015], who suggested that the offset might be explained by 363 a high propagation velocity of an impact-induced seismic wave inside the crater rim. At high-velocity impacts, the target around the impacted region was highly 364 365 compressed by a shock wave to achieve high pressure [Melosh, 1989], and the shock 366 wave propagated inside the crater. A shock wave velocity is experimentally known to increase with the shock pressure, and therefore the traveling time inside the crater could 367 368 be dominated by the shock wave velocity, and then the large offset might be explained by the short traveling time of the shock wave inside the crater. The $t_{
m max}$ at the crater rim 369

for each impact velocity can be obtained by extrapolating each fitting line.

velocities.

By using the $t_{\rm max}$ at the crater rim and the impact time ($t_{\rm max}=0$ at x=0), we can speculate that the propagation velocity of an impact-induced seismic wave inside the crater rim is about three times higher than that outside the crater rim, ~175 m s⁻¹. This high velocity could be caused in part by the shock wave propagation inside the crater rim. Fig. 7b delineates the relationship between the $t_{\rm max}$ and the x for 2-mm projectiles with different densities at the impact velocity of 4 km s⁻¹. We suspected that this relationship might move upward as the average crater rim radius becomes larger, and this behavior was recognized at the other impact velocities. However, the slope of the linear fitting line was almost the same, regardless of the projectile density. The $V_{\rm prop}$ obtained for each projectile at each impact velocity is listed in Table 7. The average $V_{\rm prop}$ was 53.0 ± 6.4 m s⁻¹, which is almost the same as that for the PC projectile at different impact

Table 7

Yasui et al. [2015] measured the propagation velocity of impact-induced seismic waves in glass beads with the mean diameter of 200 μ m: 108.9 ± 16.2 m s⁻¹, which is twice the velocity for the quartz sand obtained in the present experiments. They discussed the effect of the frequency of the seismic wave on the propagation velocity, and they concluded that the propagation velocity decreased with the decrease of the frequency. In the present investigation, the measured frequency of the first peak was 700–100 Hz, which is approximately one-half of the values reported by Yasui et al. [2015]. Thus, this

might be one of the reasons that the propagation velocity of quartz sand is smaller than that of glass beads. In addition, the grains of the quartz sand used herein have irregular shapes, differing sizes, and various angles of repose; these differences might also contribute to the low propagation velocity.

3.4. Maximum acceleration of impact-induced seismic waves

To examine the attenuation process of impact-induced seismic waves, we measured the maximum acceleration of the impact-induced seismic wave (g_{max}) in our experiments. Fig. 8a clarifies the relationship between the g_{max} and the x for a PC projectile at the impact velocities from 7 to 0.2 km s⁻¹; the decay rate of the acceleration with the distance should be derived from this relationship. It is quite clear in the figure that the g_{max} attenuates with the x at each impact velocity, and the measured g_{max} was from 100 to 1 m s⁻². Thus, the g_{max} values can be fitted by the following power law equation:

$$g_{\text{max}} = 10^{a_{\text{r}}} \cdot x^{-b_{\text{r}}},\tag{7}$$

where $a_{\rm r}$ and $b_{\rm r}$ are constants, and $b_{\rm r}$ represents the attenuation rate of an impact-induced seismic wave. These values are provided in Table. 6. The slope $b_{\rm r}$ at the impact velocities from 7 to 4 km s⁻¹ shows similar values from 3.1 to 2.5, whereas the $b_{\rm r}$ at the impact velocities from 3 to 0.2 km s⁻¹ changed with the impact velocity and/or the distance from the impact point, from 3.8 to 1.1. In contrast, the $a_{\rm r}$, which is related to the

Fig. 8

magnitude of the absolute acceleration, depended to a slight extent on the impact velocity as shown in Table. 6.

Fig. 8b shows the relationship between the $g_{\rm max}$ and the x obtained for different projectiles at the constant impact velocity of 4 km s⁻¹. The measured $g_{\rm max}$ ranged from 40 to 1 m s⁻². Each data set for the different projectiles was fitted by Eq. (7). The obtained $a_{\rm r}$ and $b_{\rm r}$ are shown in Table 7. The $a_{\rm r}$ value increased with the increase of the projectile density, whereas the $b_{\rm r}$ value systematically changed with the projectile density and was between 1.9 and 3.1. The obtained $a_{\rm r}$ and $b_{\rm r}$ values at different impact velocities showed a similar relationship with the projectile density (Table 7). For each projectile, the obtained $a_{\rm r}$ increased with the increase of the impact velocity, similar to that observed for the PC projectile.

Yasui et al. [2015] clarified that the g_{max} was scaled well by using the distance from the impact point normalized by the crater rim radius (x/R_{rim}) because the R_{rim} includes all of the effects of the impact conditions according to the crater size scaling law described in Eq. (6). This relationship between g_{max} and x/R_{rim} can be written as follows:

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$$g_{\text{max}} = 10^{a_{\text{n}}} (x/R_{\text{rim}})^{-b_{\text{n}}}.$$
 (8)

Fig. 8c-e provides the g_{max} results expressed by the normalized distance, x/R_{rim} . Panels (c) and (d) of Fig. 8 illustrate the data for a PC projectile and an Al projectile at different impact velocities, respectively, and Fig. 8e is for all of the projectiles with different projectile densities at different impact velocities. As seen in Fig. 8c, the data for the 4.75-mm-dia. PC projectile are slightly scattered, and the data at 1 km s⁻¹ are somewhat smaller values than the others, but all of the $g_{\rm max}$ values obtained at different impact velocities were well merged and could be scaled by the normalized distance. Similarly, Fig. 8d (for a 2-mm-dia. Al projectile) demonstrates that all of the data were

merged well, and the relationship between $g_{\rm max}$ and the normalized distance was independent of the impact velocity. The data for the other 2-mm-dia. projectiles showed a similar tendency. Fig. 8e indicates that all of the data for the different projectiles merged well, and this relationship was independent of both the impact velocity and the projectile density. The data could be fitted by using Eq. (8), and the $a_{\rm n}$ and $b_{\rm n}$ were 2.21 ± 0.04 and 3.18 ± 0.10 , respectively. From the empirical equation, the $g_{\rm max}$ at the crater rim $(x/R_{\rm rim}=1)$ was estimated as $160~{\rm m~s^{-2}}$; this value is approximately 16 times larger than the Earth's gravity. At $g_{\rm max}=1~{\rm G}$, the normalized distance was estimated as 2.4. The quartz sand in this region might have been fluidized by the impact-induced seismic wave.

Yasui et al. [2015] measured the $g_{\rm max}$ of 200-µm glass beads for PC, alumina, and SUS projectiles at impact velocities <150 m s⁻¹. The empirical equation fitted by Eq. (8) is also shown in Fig. 8e. Comparing our results with those reported by Yasui et al., the decay rate $(b_{\rm n})$ of the quartz sand was approximately 1.5 times larger than that of the glass beads. Surprisingly, the $g_{\rm max}$ of the quartz sand at the crater rim $(x/R_{\rm rim}=1)$

was almost the same as that of the glass beads, i.e., $\sim 150 \text{ m s}^{-2}$. We thus speculated that the minimum acceleration necessary for crater excavation at the crater rim might be approximately 15 G in the gravity regime for a crater formation process at the crater diameters from 20 to 6 cm in this experiment. An excess acceleration beyond 1 G corresponding to the fluidization of granular materials might be necessary to cause excavation flow inside the crater rim.

3.5. Duration of first upward acceleration

The duration of the first upward acceleration could be explained by the generation mechanism of impact-induced seismic waves. Yasui et al. [2015] defined the duration of the first upward acceleration as a half-period of the first wave (T_{half}), which indicates the length of time between the time point at which the acceleration rose from the background level and the time point at which it fell back down to the background level.

Fig. 9a, b illustrates the relationship between the $T_{\rm half}$ and the distance from the impact point: Fig. 9a is for a PC projectile at different impact velocities, and Fig. 9b is for 2-mm-dia. projectiles with different densities at the impact velocity of 4 km s⁻¹. In Fig. 9a, the data of the $T_{\rm half}$ are somewhat scattered but they are independent of the distance from the impact point. The $T_{\rm half}$ results were somewhat dependent on the impact velocity from 7 to 1 km s⁻¹, and the $T_{\rm half}$ at 200 m s⁻¹ was clearly smaller than the values at high impact velocities. Yasui et al. [2015] measured the $T_{\rm half}$ of 200- μ m

glass beads at impact velocities <150 m s⁻¹. They reported a T_{half} of ~0.7 ms, which is slightly smaller than that at 200 m s⁻¹ and much smaller than the values obtained at high impact velocities in our present experiments.

As shown in Fig. 9b, the $T_{\rm half}$ was independent of both the distance from the impact point and the projectile density. The average $T_{\rm half}$ for the 2-mm-dia. projectiles at 4 km s⁻¹ was 2.01 ± 0.38 ms, which is approximately three times greater than that of glass beads. In light of the data in these figures, we can say that the $T_{\rm half}$ increases with the increase of the impact velocity, and it is independent of the projectile density. Our examination of this tendency revealed the relationship between the $T_{\rm half}$ and the impact velocity for each projectile (Fig. 9c). The data are slightly scattered and the error was not very small for each result, but the $T_{\rm half}$ increased exponentially with the increase of the impact velocity. Moreover, all of the data could be fitted by one empirical power law equation as follows:

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$$T_{\text{half}} [\text{ms}] = (1.65 \pm 0.11) \cdot (v_i [\text{km s}^{-1}])^{0.14 \pm 0.05}.$$
 (9)

Yasui et al. [2015] suggested that the $T_{\rm half}$ could be consistent with the duration of a projectile penetration into the target. They calculated the penetration duration of the projectile $(t_{\rm p})$ by using the deceleration model proposed by Niimi et al. [2011]. They chose the resistance law described by the dynamic pressure for a deceleration mechanism in a hydrodynamic regime: $t_{\rm p} \propto v_{\rm i}^{-1}$. However, our experimental results cannot be

explained by this resistance law, because $T_{\rm half}$ is proportional to $v_{\rm i}^{0.14}$. We thus chose another resistance law described by viscous drag force, as follows:

$$m_{\rm p}\frac{dv}{dt} = -A_0\eta r_{\rm p}v,\tag{10}$$

where v is the projectile penetration velocity, t is the time, η is the viscosity of target material, and A_0 is the constant related to the geometrical factor determined by an experiment. The penetration duration of the projectile (t_p) could be obtained by the following equations from the solution of Eq. (10) at $v = v_0$ (constant):

$$t_{\rm p} = -\tau \ln \left(\frac{v_0}{v_{\rm i}} \right) \tag{11}$$

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$$\tau = \frac{4\pi \rho_{\rm p} r_{\rm p}^2}{3A_0 \eta}.$$
 (12)

The fitting line obtained by using Eq. (11) is also shown in Fig. 9c, and it can be seen that Eq. (11) agrees with all of the data, as does Eq. (9). We can obtain the penetration velocity just before the projectile stops (v_0) and the effective viscosity ($A_0\eta$) from Eqs. (11) and (12): the v_0 was 1.8 m s⁻¹, regardless of the projectile properties, and the $A_0\eta$ increased with the increase of the projectile density for the same projectile radius and the value was 40–240 Pas.

4. Discussions

504 4.1. Impact-induced seismic energy propagating with distance

The impact-induced seismic energy (E_s) is an important factor in evaluations of the

seismic shaking induced by the impact of small bodies on solid planets and asteroids. In this section, we estimate the E_s by using the empirical equations we obtained herein that describes the three parameters V_{prop} , g_{max} , and T_{half} . In the model proposed by Yasui et al. [2015], it is assumed that the E_s is the kinetic energy of the target material vibrating at the shell region, with a width corresponding to one cycle of the impact-induced seismic wave. The impact-induced seismic energy E_s is determined as follows:

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$$E_{\rm s} = \int_{x_1}^{x_2} \frac{1}{2} \cdot 2\pi x^2 \rho_{\rm t} V(x, t)^2 dx, \tag{13}$$

where ρ_t is the target density, x_1 and x_2 are the inside and outside distances of the shell region from the impact point, t is the time, and V(x,t) is the vibration velocity of the target particle induced by the impact. For the estimation of E_s , it is difficult to use the original waveform of the acceleration obtained in the present study because the waveform had a complex shape, as shown in Fig. 6. Therefore, in order to calculate the V(x,t), we simulated the impact-induced seismic wave (Fig. 10a).

Fig. 10

The seismic wave has a positive phase and a negative phase (see Fig. 6), and a positive acceleration phase has a larger amplitude compared to a negative acceleration phase. We therefore divided the seismic wave into two regions: one region showing a half cycle of a sinusoidal wave with $T_{\rm half}$ and $g_{\rm max}$ in the positive amplitude region, and the other region with $T_{\rm 2half}$ and $g_{\rm min}$ in the negative amplitude region. The acceleration in each region is expressed as follows:

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$$a_{\rm up}(x,t) = g_{\rm max} \sin \left[\frac{\pi}{T_{\rm half}} \left(t - \frac{x}{V_{\rm prop}} \right) \right], \tag{14}$$

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$$a_{\text{down}}(x,t) = g_{\text{min}} \sin \left[\frac{\pi}{T_{\text{2half}}} \left\{ t - \left(T_{\text{half}} + \frac{x}{V_{\text{prop}}} \right) \right\} \right], \tag{15}$$

where $a_{\rm up}(x,t)$ and $a_{\rm down}(x,t)$ are the acceleration in the positive and negative amplitude regions, respectively. The average $V_{\rm prop}$ obtained was 52.4 ± 7.2 m s⁻¹, regardless of the impact velocity and the projectile property. The $g_{\rm max}$ depended on the distance from the impact point normalized by the crater rim radius, $x/R_{\rm rim}$, and it is expressed as Eq. (8). The $g_{\rm min}$ showing the maximum acceleration in the negative amplitude region could be obtained from our experimental results, and it is illustrated in Fig. 10b; the relationship between the absolute value of $g_{\rm max}/g_{\rm min}$ and the $x/R_{\rm rim}$ is shown. The $|g_{\rm max}/g_{\rm min}|$ had a good correlation with the $x/R_{\rm rim}$, regardless of the projectile density and the impact velocity, and it can be expressed as the following power law equation:

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$$|g_{\text{max}}/g_{\text{min}}| = 10^{1.09 \pm 0.03} (x/R_{\text{rim}})^{-2.20 \pm 0.07}$$
. (16)

The T_{half} was slightly dependent on the impact velocity (v_i) ; it is expressed as Eq. (9).

The vibration velocity of the target particle can be obtained by calculating the integral of

Eqs. (14) and (15) for t, and it can be expressed as follows:

$$V_{\rm up}(x,t) = \frac{T_{\rm half}}{\pi} g_{\rm max} \left[1 - \cos \left\{ \frac{\pi}{T_{\rm half}} \left(t - \frac{x}{V_{\rm prop}} \right) \right\} \right], \tag{17}$$

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$$V_{\text{down}}(x,t) = \frac{2T_{\text{half}}}{\pi} g_{\text{max}}$$

$$+ \frac{T_{\text{2half}}}{\pi} g_{\text{min}} \left[1 - \cos \left\{ \frac{\pi}{T_{\text{2half}}} \left(t - \left\{ T_{\text{half}} + \frac{x}{V_{\text{prop}}} \right\} \right) \right\} \right]. \tag{18}$$

Here, the $T_{2\text{half}}$ was determined by the following boundary condition: $V_{\text{down}}(x, x)$

545 $x/V_{\text{prop}} + T_{\text{half}} + T_{\text{2half}} = 0$. From this boundary condition, the T_{2half} can be

546 expressed as:

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$$T_{2\text{half}} = -T_{\text{half}} \left(\frac{g_{\text{max}}}{g_{\text{min}}} \right), \tag{19}$$

and Eq. (18) can be rewritten as follows:

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$$V_{\text{down}}(x,t) = \frac{T_{\text{half}}}{\pi} g_{\text{max}} \left[1 + \cos \left\{ -\frac{\pi}{T_{\text{half}}} \left(\frac{g_{\text{max}}}{g_{\text{min}}} \right) \left(t - \left\{ T_{\text{half}} + \frac{x}{V_{\text{prop}}} \right\} \right) \right\} \right]. \tag{20}$$

Fig. 11a–c presents an example of the original waveform obtained in this experiment

Fig. 11

and the simulated waveforms calculated by using the Eqs. (14), (15), (17), and (20) for the 2-mm-dia. All projectile at the impact velocity of 4.14 km s⁻¹ (run #151105–3). Fig. 11d clarifies the relationship between the vibration velocity, V(x,t), and the distance from the impact point normalized by the crater rim radius $x/R_{\rm rim}$ at different times from 18.0 to 7.5 ms. The maximum V(x,t) decreased with the increase of the time. In addition, the normalized distance at the maximum V(x,t), i.e., $x_{\rm peak}/R_{\rm rim}$, increased with the increase of the time.

The x_1 and x_2 values of Eq. (13) can be obtained from this relationship: they can be calculated at the V(x,t) = 0 for each time. The impact-induced seismic energy (E_s) can be calculated by substituting Eqs. (17) and (20) into Eq. (13) with x_1 and x_2 . Fig.

12a shows the relationship between the impact-induced seismic energy normalized by the projectile kinetic energy, E_s/E_k , and the normalized distance at the maximum V(x,t), i.e., $x_{\text{peak}}/R_{\text{rim}}$, in the range of $x_{\text{peak}}/R_{\text{rim}}$ from 1 to 10 at various impact conditions. The E_s/E_k depends on the impact velocity and the projectile size and density. For example, the E_s/E_k for a PC projectile at 200 m s⁻¹ was more than 10 times larger than that at 7 km s⁻¹, and the E_s/E_k for an Al projectile (2.7 g cm⁻³) was twice as large as that for a WC projectile (14.9 g cm⁻³) at 4 km s⁻¹. The E_s/E_k continued to decrease with the increase of the normalized distance $x_{\rm peak}/R_{\rm rim}$, following the power law relationship of $E_{\rm s}/E_{\rm k} \propto \left(x_{\rm peak}/R_{\rm rim}\right)^{-4.5}$. This

 $x_{\rm peak}/R_{\rm rim}$, following the power law relationship of $E_{\rm s}/E_{\rm k} \propto \left(x_{\rm peak}/R_{\rm rim}\right)^{-4.5}$. This continuous decrease of the $E_{\rm s}/E_{\rm k}$ could be caused by the energy dissipation in the plastic wave. The impact generates a detached stress wave, initially as a strong shock wave, that propagates away from the impact point, decays in magnitude, and eventually becomes an elastic seismic wave. When the stress wave is in the shock and plastic regimes it loses the energy rapidly with the normalized distance. On the other hand, once the stress wave becomes an elastic wave, the energy losses should be small.

Thus, the elastic seismic wave did not appear in these experimental conditions, and the impact-induced seismic efficiency factor could not be determined because the impact-induced seismic efficiency factor is the $E_{\rm s}/E_{\rm k}$ that ends up in the seismic/elastic wave that can propagate large distances from the impact point. We expect that the $E_{\rm s}/E_{\rm k}$ continues to decrease until the plastic wave changes to the elastic wave, and then the

 $E_{\rm s}/E_{\rm k}$ becomes constant. Therefore, our obtained $E_{\rm s}/E_{\rm k}$ could be the upper limit of the impact-induced seismic efficiency factor.

We compared the E_s/E_k values obtained in this study with the reported impact-induced seismic efficiency factor from previous experiments. For example, McGarr et al. [1969] examined the seismic impulse normalized by the kinetic energy of a projectile, I/E_k (which is similar to our E_s/E_k parameter) for a loose sand target at impact velocities from 7 to 2 km s⁻¹ in laboratory experiments. They obtained the average I/E_k value of $(0.6 \pm 0.4) \times 10^{-5}$. We replotted their I/E_k values against the normalized distance shown in Fig. 12a: we calculated the crater rim radius for each experiment in the paper of McGarr et al. [1969] by using our crater size scaling law expressed by Eq. (6). The McGarr et al. data then seemed to increase with the increase of the normalized distance; this trend is the opposite of our results for each impact condition. As we described above, the E_s/E_k depends on the impact velocity, and it could increase with the decrease of the impact velocity.

We thus presume that the increase of the I/E_k with the increase of the normalized distance is not a true dependency on the distance; rather, it could be caused by the impact velocity change from 7 to 2 km s⁻¹. If so, it might be possible that the I/E_k also decreases with the increase of the normalized distance, because the maximum acceleration g_{max} was observed to decrease with the normalized distance, x/R_{rim} , for the loose sand used by McGarr et al. [1969] as shown in Fig. 12b. That is, their obtained seismic wave did

Fig. 12

not also appear to be the elastic wave. Their obtained $I/E_{\rm k}$ might therefore correspond to the upper limit of the impact-induced seismic efficiency factor, as is the case in our present investigation. Their $I/E_{\rm k}$ value was larger than the values we obtained herein, even though the impact velocity range was almost the same. The $g_{\rm max}$ for loose sand is also greater than that for our quartz sand (Fig. 12b). This might be caused by differences in mechanical properties of the target, such as the bulk sound velocity and the target density.

Moore et al. [1970] carried out large-scale impact experiments using a missile at the estimated impact velocity of 2.5 km s⁻¹ at the White Sands Missile Range (New Mexico, U.S.). The impact-induced seismic efficiency factor of 10^{-5} at the quite large kinetic energy of the projectile, 4.5×10^{20} J, was obtained. In addition, Latham et al. [1970] estimated the impact-induced seismic efficiency factor as 10^{-5} to 10^{-6} by using the lunar module impact provided by the Apollo; the impact velocity of the lunar module was estimated to be 1.7 km s⁻¹ and the kinetic energy was 3.6×10^{11} J [Toksöz et al., 1974]. Their values are larger than our calculated $E_{\rm s}/E_{\rm k}$ at the normalized distances far from 6; however, our values could be the upper limit of the impact-induced seismic efficiency factor.

Herein, we measured the acceleration along the direction perpendicular to the target surface. However, the particle motion induced by the seismic wave has not only a vertical component (Z); the motion also has horizontal (X) and radial (Y) components in the target

Fig. 13

621 (Fig. 13a). We thus measured the acceleration excited along these three axes by using a

triaxial accelerometer to confirm the amplitude of the acceleration along the X- and Y-

axes. Fig. 13b shows the acceleration measured at the X-, Y-, and Z-axes for a PC

projectile at 2.4 km s⁻¹. The triaxial accelerometer was set at 11 cm from the impact point

 $(x/R_{\rm rim} = 1.5)$. We observed that acceleration at the Y-axis was not detected.

Based on the crater formation model proposed by Maxwell [1977], each flow line drawn from an impact point never crosses another flow line, which means that the particle motion does not have a radial component. This is the reason why the acceleration at the Y-axis was not detected. In addition, we firstly observed that the maximum acceleration at the X-axis was about five times larger than that at the Z-axis, and thus it is clear that the kinetic energy of the particle is dominated by the horizontal motion. When we calculate the particle velocity considering the acceleration along the X-axis, the $E_{\rm s}/E_{\rm k}$ could be more than one order of magnitude larger than that calculated in Fig. 12a. This is why our calculated E_s/E_k was smaller than the impact-induced seismic efficiency factor estimated from the missile experiments and the lunar module impact [Moore et al., 1970; Latham et al., 1970]. In the future, in order to determine the impactinduced seismic efficiency factor, it is necessary to measure the acceleration by using a triaxial accelerometer at a distant region where the seismic wave changes to the elastic wave.

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4.2. Possibility of detecting an impact-induced seismic shaking on asteroids

To estimate the effects of impact-induced seismic shaking on asteroids, we used the empirical equations that we obtained in this study. These equations were obtained under Earth's gravity, 1 G, so we should consider the effect of microgravity when applying our established model to the resurfacing processes that are due to impact-induced seismic shaking on asteroid surfaces. As described next, we constructed an impact-induced seismic shaking model including the effect of gravity on asteroids by comparing our experimental results with the scaling law for ejecta velocity distribution, and we discuss the possibility of the impact-induced seismic shaking on asteroids.

We first consider the particle velocity ejected at the position of the crater radius in accord with the scaling law for ejecta velocity distribution that was proposed by Housen and Holsapple [2011]. The ejection velocity at the position of the crater radius, $v_{\rm e}$, is written as follows:

$$v_{\rm e} = c_0 \sqrt{gR_{\rm c}},$$
(21)

where g is the gravitational acceleration in m s⁻² and c_0 is the constant determined by experiments for each target material. Next, we assume that the ejection velocity of a particle at the position of the crater radius $(x/R_c=1)$ can be simply estimated by using the product of the maximum acceleration, g_{max} , in Eq. (8), and the duration of the first upward acceleration, T_{half} . The maximum acceleration at $x/R_c=1$ is rewritten as

 $g_{\text{max}} = (1.26)^{b_{\text{n}}} g_0$ under a certain gravity, where $R_{\text{c}} = R_{\text{rim}}/1.26$ and g_0 is the maximum acceleration at the crater rim. Thus, the ejection velocity can be written as follows:

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$$v_{\text{e}\perp}[\text{m s}^{-1}] = g_{\text{max}} \cdot T_{\text{half}} = (1.26)^{b_{\text{n}}} g_0 T_{\text{half}}, \tag{22}$$

where $v_{\rm e\perp}$ is the ejection velocity of the particles in the normal direction. It is assumed that the target particles are ejected from the target surface with the ejection angle of 45°, a typical ejection angle of granular materials [e.g., Tsujido et al., 2015], and it is assumed that the ejection velocity estimated from the scaling law of Eq. (21) is equal to be that obtained in this experiment as Eq. (22). Thus, the ejection velocity of the particles at $x/R_{\rm c}=1$ can be rewritten as $v_{\rm e}=\sqrt{2}v_{\rm e\perp}$, and the following equation is constructed by using Eqs. (21) and (22):

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$$c_0 \sqrt{gR_c} = \sqrt{2} \times (1.26)^{b_n} g_0 T_{\text{half}}, \tag{23}$$

672 that is,

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$$g_0 = \frac{c_0 (1.26)^{-b_n}}{\sqrt{2}} \times \frac{\sqrt{gR_c}}{T_{\text{half}}}.$$
 (24)

The R_c is calculated from the crater size scaling law of Eq. (6) and is shown as follows:

675
$$R_{\rm c} = (8.75 \times 10^{-1}) g^{-0.176} v_{\rm i}^{0.352} r_{\rm p}^{0.824} \left(\frac{\rho_{\rm t}}{\rho_{\rm p}}\right)^{-0.289}, \quad (25)$$

where the projectile mass, $m_{\rm p}$, can be written as $4\pi\rho_{\rm p}r_{\rm p}^3/3$ by assuming a spherical

shape. Therefore, the g_0 can be rewritten as follows:

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$$g_0 = \frac{(3.17 \times 10^{-1})c_0 (gr_p)^{0.412} v_i^{0.176} (\rho_t/\rho_p)^{-0.145}}{T_{\text{half}}}, \quad (26)$$

where $b_n = 3.18$ in Eq. (8). To compare our semi-theoretical equation for g_0 with the obtained g_0 , which is 10^{a_n} ($a_n = 2.21 \pm 0.04$) in Eq. (8), in our experiments, we used the T_{half} of the empirical equation, Eq. (9). Substituting Eq. (9) into Eq. (26), the g_0 can be rewritten as follows:

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$$g_0 = (3.97 \times 10^2) (gr_p)^{0.412} v_i^{0.036} \left(\frac{\rho_t}{\rho_p}\right)^{-0.145}, \quad (27)$$

where $c_0 = 0.76$ m s⁻¹ as obtained by Tsujido et al. [2015]. Fig. 14 shows the calculated results of Eq. (27) at the impact velocity (v_1) of 0.2–7 km s⁻¹ for various projectiles. The gray zone shown in Fig. 14 represents the 10^{a_n} in Eq. (8), and all of the calculated g_0 values were a little smaller than the observed g_0 (10^{a_n} in Eq. (8)). The difference between the calculated g_0 values and the observed g_0 might be caused by the ejection angle and/or the duration, T_{half} . The g_0 was measured below the target surface (2.5 cm from the target surface) in our experiments while the calculated g_0 was estimated on the target surface. Therefore, the g_0 might depend on the depth from the target surface. However, their differences were within a factor of 2 and they were consistent within error bars. Our semi-theoretical equation of Eq. (26) could be suitable for the estimation of g_0 at various impact conditions.

Fig. 14

Although the g_{max} at the crater rim, that is, the g_0 was very similar in all of our experiments, the semi-theoretical equation of Eq. (26) suggests that the g_0 is sensitive to the product of surface gravity and the projectile radius. In our experiments, the projectile radius was changed by a factor of 2, and thus an effect of the projectile radius on the g_0 was not recognized. This result should be verified by future experiments and/or numerical simulations.

Lastly, we considered the impact-induced seismic shaking on an idealized small (500 m in diameter) body with a bulk density of 1.46 g cm⁻³, which is the same as our target.

In this calculation, we should include the effect of the impactor size on the T_{half} of Eq. (26), and we therefore used the theoretical equations of Eqs. (11) and (12). Substituting Eqs. (11) and (12) into Eq. (26), the g_0 can be rewritten as follows:

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$$g_0 = -(5.79 \times 10^{-2})(A_0 \eta) (gr_p)^{0.412} v_i^{0.176} r_p^{-2} \rho_t^{-0.145} \rho_p^{-0.856} \left[\ln \left(\frac{v_0}{v_i} \right) \right]^{-1}, \quad (28)$$

where $c_0 = 0.76$ m s⁻¹ obtained by Tsujido et al. [2015], and $A_0\eta$ and v_0 were determined by our experiments using PC projectiles. The $A_0\eta$ and v_0 values were 114 Pas and 1.3 m s⁻¹, respectively. In this calculation, the impactors with the diameter of 0.01, 0.1, 0.5, and 1 m and the density of 1.4 g cm⁻³ (which is the average density of C-type asteroids as shown by Britt et al. [2002]) were assumed to collide at the impact velocity of 5 km s⁻¹, a typical collisional velocity in the region of main belt. We then calculated the maximum acceleration attenuated with the distance calculated for the

different sizes of the impactor by using Eq. (8), i.e., $g_{\text{max}} = g_0(x/R_{\text{rim}})^{-3.18}$.

Fig. 15 illustrates the relationship between the distance from the impact point Fig. 15

normalized by the crater rim radius, x/R_{rim} , and the maximum acceleration, g_{max} . The

region at which the impact-induced seismic shaking occurred was assumed to be within

the distance from the impact point corresponding to the g_{max} beyond the surface gravity

of the idealized small body. The g_{max} was observed to increase with the decrease of the

impactor size at the same normalized distance. The g_{max} with the impactor size of 0.01

m at $x/R_{\text{rim}} = 1$ was more than three orders of magnitude larger than that obtained with

722 a 1-m impactor.

The area at which the impact-induced seismic shaking occurred changed with the impactor size. The normalized distance at which the impact-induced seismic shaking occurred changed from 9 to 0.9 when the impactor size increased from 0.01 m to 1 m. When an impact crater with a rim diameter of 2.8 m is formed by a 0.01-m impactor on this idealized small body, impact-induced seismic shaking could be induced within the distance of 12 m from the impact point. However, when a crater with a diameter of 123 m is formed by a 1-m impactor, the region at which the impact-induced seismic shaking occurs could be limited to within the crater rim, and thus the seismic shaking will not

5. Summary

occur far from the crater.

We carried out impact cratering experiments using eight types of spherical projectiles with density ranging from 14.9 to 1.2 g cm⁻³ on 500-µm quartz sand at the impact velocity from 7 to 0.2 km s⁻¹. We examined the waveform of the acceleration generated just after the impact, and we assessed the characteristics of the impact-induced seismic wave by using three to five accelerometers set at 5.1 to 30.1 cm from the impact point in order to consider the possibility of impact-induced seismic shaking on asteroids. Our results are summarized as follows:

- 1. We applied our results obtained under various impact conditions (e.g., the projectile size, projectile density, and impact velocity) to the crater size scaling law proposed by Housen and Holsapple [2011], and we determined the following empirical equation, $\pi_R \cdot \pi_4^{-0.044} = 10^{-0.265} \pi_2^{-0.176}$, where π_R , π_2 , and π_4 are the non-dimensional scaling parameters related to the crater rim radius, gravity, impact velocity, projectile density, and target density. We obtained the power law indices of the coupling parameters, μ and ν , as 0.43 and 0.35, respectively.
 - 2. The propagation velocity of an impact-induced seismic wave, V_{prop} , was determined by using the duration between the impact time and the time showing the maximum acceleration of the first upward peak. The obtained V_{prop} was $52.4 \pm 7.2 \text{ m s}^{-1}$ outside the crater, regardless of the impact velocity and the projectile properties. The V_{prop} inside the crater rim was speculated to be ~175 m s⁻¹, which is approximately three times larger than that outside the crater rim.

- 754 3. The maximum acceleration of the impact-induced seismic wave, g_{max} , could be scaled as the distance from the impact point normalized by the crater rim radius, x/R_{rim} , regardless of the impact velocity and the projectile properties, and the obtained empirical equation was shown as $g_{\text{max}} = 10^{2.21} \cdot (x/R_{\text{rim}})^{-3.18}$. The g_{max} of quartz sand at the $x/R_{\text{rim}} = 1$ obtained in this study matched well with that of the 200- μ m glass beads reported by Yasui et al. [2015].
- The duration of the first upward acceleration was defined as a half period of the first upward wave, $T_{\rm half}$, and we observed that $T_{\rm half}$ was slightly dependent on the impact velocity but independent of the projectile properties. The obtained empirical equation was $T_{\rm half}$ [ms] = 1.65($v_{\rm i}$ [km s⁻¹])^{0.14}. Our obtained $T_{\rm half}$ could be explained by the resistance law described by viscous drag force.
- 765 5. We calculated the seismic energy normalized by the kinetic energy of the projectile, 766 E_s/E_k , by using our obtained empirical equations at the normalized distance from 767 $x/R_{rim} = 1$ to 10 under various impact conditions. The E_s/E_k depended on the 768 impact velocity and projectile properties, but the decay constant was approximately 769 4.5, regardless of the impact condition.
- 6. We estimated the distance from the crater rim at which the impact-induced seismic shaking occurred on an idealized small (500 m in diameter) body by using our semi-theoretical equation. When we assumed the impact velocity of the impactor to be 5 km s⁻¹, the impact-induced seismic shaking area could be calculated for various

impactor sizes.

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853	Figure Captions:
854	
855	Figure 1
856	Schematic illustration of the experimental setup. (a) Two-stage vertical gas gun. (b) Five
857	accelerometers set on the target surface. (c) Cross-section of the accelerometers set on the
858	target surface, and an illustration of the crater dimensions.
859	
860	Figure 2
861	Impact craters on the quartz sand target. (a) The PC projectile at 1.13 km s ⁻¹ , run
862	#151104-3. (b) The PC projectile at 6.89 km s ⁻¹ , run $#141003-2$, which was the largest
863	impact crater in this study. (c) The Al projectile at 4.12 km s ⁻¹ , run #150309–1. (d) The
864	Cu projectile at 3.63 km s ⁻¹ , run #150820–5. Each scale bar indicates 5.0 cm. The thick
865	black lines around the crater rim are the accelerometers' cords.
866	
867	Figure 3
868	Crater profiles obtained at different impact conditions. (a) The results obtained with the
869	PC projectile at the impact velocities of 1.1, 2.2, 4.2, and 6.2 km s^{-1} . (b) The crater
870	profiles expressed by the relationship between the values of $x/R_{\rm rim}$ and the $y/R_{\rm rim}$ for
871	panel (a). (c) The results of the projectiles with different densities at the impact velocity
872	of ~ 4.0 km s ⁻¹ . (d) The crater profile expressed by the relationship between the x/R_{rim}

873	and	the y/R_{rim} for panel (c).
874		
875	Fig	ure 4
876	The	relationship between the ratio of the crater depth to the crater diameter, d_c/D_c , and
877	the	projectile density. Each symbol indicates an impact velocity range.
878		
879	Fig	ure 5
880	(a)	The relationship between the non-dimensional scaling parameter π_4 and the π_R
881		defined in Eqs. (1) and (2). The results were obtained by the experiments using seven
882		projectiles with different densities at a constant impact velocity of 2.0 km s ⁻¹
883		(circles) and 4.0 km s ⁻¹ (squares). The solid and dashed lines represent the fitting
884		lines determined by using the power law equation of Eq. (5).
885	(b)	The relationship between the non-dimensional scaling parameter π_2 and the π_R
886		π_4^{-b} (in this study, we obtained $b = 0.044$). The solid line represents the fitting line
887		determined by using the power law equation of Eq. (6).
888		
889	Fig	ure 6
890	(a)	A typical sample of the impact-induced seismic wave showing a single pulse-like
891		wave observed in this study. A PC projectile was impacted at the velocity of 5.16 km

 $\ensuremath{\text{s}^{\text{-1}}}$ (run #140530–3). The three different curves represent the data recorded by three

- accelerometers set at different distances from the impact point, x. Time 0 is the impact time of the projectile.
- 895 (b) Enlargement of the horizontal axis of panel (a) ranging from -2 to 15 ms.
- 896 (c) An example of the impact-induced seismic wave showing a damped vibration wave
 897 observed far from the crater rim. An Al projectile was impacted at the velocity of
 898 4.12 km s⁻¹ (run #150309–1).
- 899 (d) Enlargement of the horizontal axis of panel (c) ranging from -2 to 20 ms.
- 900 (e) The physical properties of an impact-induced seismic wave: the traveling time (t_{max}), 901 the maximum acceleration (g_{max}), and the duration of the first upward wave of the 902 acceleration (T_{half}).

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Figure 7

The relationship between the traveling time ($t_{\rm max}$) and the distance from the impact point, x (a) for PC projectile at different impact velocities, and (b) for 2-mm projectiles with different densities at the impact velocity of 4 km s⁻¹. Horizontal lines represent the average crater rim radius at each impact velocity. The fitting lines were obtained by using a linear function.

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Figure 8

(a, b) The relationship between the maximum acceleration (g_{max}) and the distance from

the impact point, x (a) for PC projectile at impact velocities ranging from 7 to 0.2
${\rm km\ s^{-1}}$ and (b) for 2-mm projectiles with the projectile density ranging from 14.9 to
2.5 g cm ⁻³ at the impact velocity of 4 km s ⁻¹ . Each line represents the fitting line
obtained by using Eq. (7).

(c–e) The relationship between the maximum acceleration (g_{max}) and the distance from the impact point normalized by the crater rim radius, x/R_{rim} (c) for PC projectile at impact velocities from 7 to 0.2 km s⁻¹, (d) for Al projectile at impact velocities from 5 to 2 km s⁻¹, and (e) for all data. The solid and dashed lines in panel (e) represent the fitting lines obtained by using Eq. (8) and the empirical equation used for 200- μ m glass beads obtained by Yasui et al. [2015], respectively.

Figure 9

- (a, b) The relationship between the duration of the first upward acceleration (T_{half}) and the distance from the impact point, x (a) for PC projectile at impact velocities from 7 to 0.2 km s⁻¹ and (b) for 2-mm projectiles with the projectile density from 14.9 to
 2.5 g cm⁻³ at the impact velocity of 4.0 km s⁻¹. The solid line in panel (b), and the dotted line and gray area in panels (a) and (b) represent the average values for all of the data obtained in this study, that for the 200-μm glass beads used by Yasui et al.
 [2015], and the standard deviation for each data set, respectively.
 - (c) The average T_{half} at each impact velocity for all projectiles. The solid and dashed

933 lines represent the fitting lines obtained by using Eqs. (9) and (11), respectively. 934 935 Figure 10 936 (a) Schematic of a simulated impact-induced seismic wave for Eqs. (14) and (15). (b) The relationship between the absolute ratio of the maximum acceleration in the 937 938 positive amplitude region to that in the negative amplitude region, i.e., $|g_{\text{max}}/g_{\text{min}}|$, 939 and the normalized distance, x/R_{rim} for various projectiles. The solid line represents 940 the fitting line obtained with Eq. (16). 941 942 Figure 11 943 (a) Waveforms of acceleration measured at x = 10.7, 13.0, 15.4, 18.1,and 22.4 cm from 944 the impact point, (run #151105-3). 945 (b) The relationship between the acceleration, a, and the time in ms, calculated by using 946 Eqs. (14) and (15) under the same impact conditions as those in panel (a). The 947 numbers in the legend represent the x/R_{rim} . 948 (c) The relationship between the vibration velocity, V, and the time in ms, calculated 949 with Eqs. (17) and (20) under the same impact conditions as those in panel (a). The 950 numbers in the legend represent the x/R_{rim} . 951 (d) The relationship between the vibration velocity, V, and the distance from the impact

point normalized by the crater rim radius, x/R_{rim} , calculated with Eqs. (17) and (20)

953 under the same impact conditions as those in panel (a). The numbers in the legend 954 represent the elapsed times.

Figure 12

- (a) The relationship between the seismic energy normalized by the kinetic energy of the projectile, $E_{\rm s}/E_{\rm k}$, and the distance from the impact point normalized by the crater rim radius at the maximum vibration velocity, $x_{\rm peak}/R_{\rm rim}$, on Fig. 11d under various impact conditions. The open symbols represent the data of loose sand obtained by McGarr et al. [1969].
- (b) The relationship between the maximum acceleration, g_{max} , and the distance from the impact point normalized by the crater rim radius, x/R_{rim} . This figure is same as Fig. 8e. The solid circles represent the data of loose sand from McGarr et al. [1969].

Figure 13

- (a) Cross-section of the triaxial accelerometer setting on the target surface. The X-, Y-, and Z-vectors indicate the directions of the measured acceleration.
- (b) Impact-induced seismic waves at the three orthogonal directions measured by the triaxial accelerometer set at 11 cm from the impact point for a PC projectile at the impact velocity of 2.4 km s⁻¹: the crater rim radius is 7.5 cm.

Figure 14

Comparison of the g_0 values calculated using the semi-theoretical equation of Eq. (27) for various impact conditions (symbols) with those obtained in our experiments (gray zone).

Figure 15

The relationship between the maximum acceleration, $g_{\rm max}$, and the distance from the impact point normalized by the crater rim radius, $x/R_{\rm rim}$, for an impactor with a density of 1.4 g cm⁻³ at the impact velocity of 5 km s⁻¹ on an idealized small (500 m in diameter) body and a 1.46 g cm⁻³ density. The thick lines represent the results calculated with Eqs. (8) and (28) with impactor diameters from 0.01 to 1 m. The horizontal dotted line shows the surface gravity of the idealized small body. The number in each parenthesis represents the crater rim radius.

Run number	Projectile	$v_{\rm i}$, m s ⁻¹	D_{rim} , cm	$D_{\rm c}$, cm	$d_{ m rim}$, cm	$d_{ m c}$, cm	$D_{\rm rim}/D_{\rm c}$	$d_{ m c}/D_{ m c}$
150819-3	PC	4230	17.7	13.1	3.8	3.0	1.36	0.23
151104-1	PC	2208	13.7	10.7	2.6	2.1	1.29	0.20
151104-2	PC	6150	20.2	16.0	4.1	3.5	1.26	0.22
151104-3	PC	1125	11.2	8.6	2.3	2.0	1.30	0.23
151105-2	PC	1206	11.7	9.2	2.0	1.6	1.27	0.17
150610-1	Al	1994	8.2	6.3	1.4	1.1	1.30	0.17
150611-3	Al	5015	12.1	9.5	2.8	2.4	1.28	0.26
150612-1	Al	1815	8.2	6.3	1.3	0.9	1.31	0.15
150612-2	Al	2062	8.6	7.0	1.2	0.9	1.24	0.13
150820-1	Al	2920	9.5	7.2	1.5	1.2	1.32	0.16
150820-7	Al	3953	10.8	8.4	1.5	1.2	1.29	0.14
151105-3	Al	4143	10.6	8.2	1.8	1.3	1.29	0.16
150611-5	Gl	5318	11.5	9.3	1.8	1.4	1.24	0.15
150611-6	Gl	2281	8.6	6.9	1.4	1.1	1.25	0.16
150612-5	Gl	2023	8.4	7.0	1.3	1.1	1.19	0.16
150820-2	Gl	4023	10.5	8.4	1.9	1.6	1.24	0.19
150821-1	Gl	3427	10.2	7.9	1.7	1.3	1.28	0.16
150821-2	Gl	4013	10.5	8.6	1.5	1.2	1.23	0.14
150820-3	Ti	3834	11.9	9.8	2.1	1.7	1.21	0.17
150821-3	Ti	2044	10.3	8.2	1.9	1.6	1.25	0.20

150821-5	Ti	1972	10.0	7.7	1.9	1.6	1.29	0.20
160720-1	Ti	5181	13.4	11.0	2.1	1.8	1.22	0.16
150820-4	ZrO ₂	3934	13.3	11.4	2.8	2.5	1.17	0.22
160719-3	ZrO ₂	2153	10.8	8.5	2.0	1.6	1.27	0.19
160720-2	ZrO ₂	5727	14.7	11.3	2.9	2.4	1.28	0.21
150610-2	SUS	1841	11.3	8.9	2.6	2.2	1.26	0.25
150611-4	SUS	4936	15.8	12.7	4.1	3.7	1.24	0.29
150612-3	SUS	1848	11.0	8.7	2.7	2.3	1.27	0.27
150612-4	SUS	1824	10.9	8.6	2.7	2.3	1.26	0.27
150820-6	SUS	3805	14.2	11.9	3.1	2.7	1.20	0.23
150820-5	Cu	3626	14.4	11.7	3.7	3.2	1.23	0.27
151105-1	Cu	2114	11.9	9.8	3.1	2.8	1.22	0.29
160719-1	Cu	2238	12.4	9.8	2.9	2.4	1.27	0.25
160720-4	Cu	5348	15.8	12.3	2.7	2.2	1.29	0.18
160721-3	Cu	2306	12.5	10.3	2.6	2.2	1.22	0.22
150819-4	WC	3422	16.6	13.6	4.1	3.6	1.22	0.26

 D_{rim} : Crater rim diameter

 D_c : Crater diameter measured on the pre-shot surface

 d_{rim} : Crater depth from the rim peak to the crater bottom

 d_c : Crater depth from the pre-shot surface to the crater bottom

 $D_{\text{rim}}/D_{\text{c}}$: Ratio of crater rim diameter to crater diameter

 d_c/D_c : Ratio of crater depth to crater diameter

Run number	Projectile	$m_{ m p}$, mg	$v_{\rm i},{ m m~s}^{-1}$	$R_{\rm rim}$, cm	π_2	$\pi_{ m R}$
140528-1	PC	68.0	2038	7.1	5.60×10 ⁻⁹	15.65
140528-2	PC	68.0	2289	7.6	4.44×10 ⁻⁹	16.86
140528-3	PC	68.0	2500	7.5	3.72×10 ⁻⁹	16.63
140528-4	PC	68.0	3852	8.3	1.57×10 ⁻⁹	18.20
140529-1	PC	68.0	1424	6.5	1.15×10 ⁻⁸	14.37
140529-2	PC	68.0	1552	6.7	9.66×10 ⁻⁹	14.70
140529-3	PC	68.0	1684	6.6	8.21×10 ⁻⁹	14.65
140529-4	PC	68.0	1579	6.4	9.34×10 ⁻⁹	14.16
140530-1	PC	68.0	2279	7.2	4.48×10 ⁻⁹	15.88
140530-2	PC	68.0	3748	8.3	1.66×10 ⁻⁹	18.19
140530-3	PC	68.0	5155	9.6	8.76×10 ⁻¹⁰	21.08
140530-4	PC	68.0	3381	8.5	2.04×10 ⁻⁹	18.82
140710-2	PC	68.0	221	3.0	4.76×10 ⁻⁷	6.54
140710-6	PC	68.0	170	2.7	8.04×10 ⁻⁷	5.97
140715-2	PC	68.0	222	3.2	4.71×10 ⁻⁷	6.99
140715-3	PC	68.0	217	3.2	4.92×10 ⁻⁷	7.00
140718-1	PC	68.0	227	3.2	4.50×10 ⁻⁷	7.15
140718-2	PC	68.0	208	3.1	5.36×10 ⁻⁷	6.89
140718-3	PC	68.0	204	3.2	5.59×10 ⁻⁷	7.08
140718-4	PC	68.0	217	3.3	4.92×10 ⁻⁷	7.31

140720-1 PC 68.0 173 2.9 7.78×10 ⁻⁷ 6.45 140720-2 PC 68.0 211 3.2 5.23×10 ⁻⁷ 7.07 140720-3 PC 68.0 217 3.3 4.97×10 ⁻⁷ 7.24 140720-4 PC 68.0 189 3.0 6.49×10 ⁻⁷ 6.58 140720-5 PC 68.0 205 3.1 5.52×10 ⁻⁷ 6.75 141001-1 PC 68.0 1739 6.6 7.70×10 ⁻⁹ 14.65 141002-1 PC 68.0 1506 6.5 1.03×10 ⁻⁸ 14.32 141002-2 PC 68.0 2495 7.6 3.74×10 ⁻⁹ 16.80 141002-3 PC 68.0 3268 8.7 2.18×10 ⁻⁹ 19.08 141002-4 PC 68.0 4417 9.3 1.19×10 ⁻⁹ 20.51 141002-5 PC 68.0 5754 10.3 7.03×10 ⁻¹⁰ 22.76 141003-1 PC 68.0 6887 11.4 4.91×10 ⁻¹⁰ 25.19 141003-2 PC 68.0 6887 11.4 4.91×10 ⁻¹⁰ 25.19 141003-3 PC 68.0 4003 9.5 1.45×10 ⁻⁹ 21.00 141003-4 PC 68.0 1689 6.8 8.16×10 ⁻⁹ 14.97 141003-5 PC 68.0 3016 8.4 2.56×10 ⁻⁹ 18.62 141202-1 PC 68.0 6158 10.8 6.14×10 ⁻¹⁰ 23.75 141202-2 PC 68.0 6402 10.9 5.68×10 ⁻¹⁰ 24.08 141203-1 PC 68.0 5330 9.9 8.19×10 ⁻¹⁰ 21.85 141203-3 PC 68.0 5330 9.9 8.19×10 ⁻¹⁰ 21.85 141203-3 PC 68.0 2385 7.5 4.09×10 ⁻⁹ 16.62 141203-4 PC 68.0 2385 7.5 4.09×10 ⁻⁹ 16.62							
140720-3 PC 68.0 217 3.3 4.97×10 ⁻⁷ 7.24 140720-4 PC 68.0 189 3.0 6.49×10 ⁻⁷ 6.58 140720-5 PC 68.0 205 3.1 5.52×10 ⁻⁷ 6.75 141001-1 PC 68.0 1739 6.6 7.70×10 ⁻⁹ 14.65 141002-1 PC 68.0 1506 6.5 1.03×10 ⁻⁸ 14.32 141002-2 PC 68.0 2495 7.6 3.74×10 ⁻⁹ 16.80 141002-3 PC 68.0 3268 8.7 2.18×10 ⁻⁹ 19.08 141002-4 PC 68.0 4417 9.3 1.19×10 ⁻⁹ 20.51 141002-5 PC 68.0 5754 10.3 7.03×10 ⁻¹⁰ 22.76 141003-1 PC 68.0 6711 11.1 5.17×10 ⁻¹⁰ 24.52 141003-2 PC 68.0 6887 11.4 4.91×10 ⁻¹⁰ 25.19 141003-3	140720-1	PC	68.0	173	2.9	7.78×10 ⁻⁷	6.45
140720-4 PC 68.0 189 3.0 6.49×10 ⁻⁷ 6.58 140720-5 PC 68.0 205 3.1 5.52×10 ⁻⁷ 6.75 141001-1 PC 68.0 1739 6.6 7.70×10 ⁻⁹ 14.65 141002-1 PC 68.0 1506 6.5 1.03×10 ⁻⁸ 14.32 141002-2 PC 68.0 2495 7.6 3.74×10 ⁻⁹ 16.80 141002-3 PC 68.0 3268 8.7 2.18×10 ⁻⁹ 19.08 141002-4 PC 68.0 4417 9.3 1.19×10 ⁻⁹ 20.51 141002-5 PC 68.0 5754 10.3 7.03×10 ⁻¹⁰ 22.76 141003-1 PC 68.0 6887 11.4 4.91×10 ⁻¹⁰ 25.19 141003-2 PC 68.0 6887 11.4 4.91×10 ⁻¹⁰ 25.19 141003-3 PC 68.0 4003 9.5 1.45×10 ⁻⁹ 21.00 141003-4	140720-2	PC	68.0	211	3.2	5.23×10 ⁻⁷	7.07
140720-5 PC 68.0 205 3.1 5.52×10 ⁻⁷ 6.75 141001-1 PC 68.0 1739 6.6 7.70×10 ⁻⁹ 14.65 141002-1 PC 68.0 1506 6.5 1.03×10 ⁻⁸ 14.32 141002-2 PC 68.0 2495 7.6 3.74×10 ⁻⁹ 16.80 141002-3 PC 68.0 3268 8.7 2.18×10 ⁻⁹ 19.08 141002-4 PC 68.0 4417 9.3 1.19×10 ⁻⁹ 20.51 141002-5 PC 68.0 5754 10.3 7.03×10 ⁻¹⁰ 22.76 141003-1 PC 68.0 6711 11.1 5.17×10 ⁻¹⁰ 24.52 141003-2 PC 68.0 6887 11.4 4.91×10 ⁻¹⁰ 25.19 141003-3 PC 68.0 4003 9.5 1.45×10 ⁻⁹ 21.00 141003-4 PC 68.0 1689 6.8 8.16×10 ⁻⁹ 14.97 141003-5	140720-3	PC	68.0	217	3.3	4.97×10 ⁻⁷	7.24
141001-1 PC 68.0 1739 6.6 7.70×10 ⁻⁹ 14.65 141002-1 PC 68.0 1506 6.5 1.03×10 ⁻⁸ 14.32 141002-2 PC 68.0 2495 7.6 3.74×10 ⁻⁹ 16.80 141002-3 PC 68.0 3268 8.7 2.18×10 ⁻⁹ 19.08 141002-4 PC 68.0 4417 9.3 1.19×10 ⁻⁹ 20.51 141002-5 PC 68.0 5754 10.3 7.03×10 ⁻¹⁰ 22.76 141003-1 PC 68.0 6711 11.1 5.17×10 ⁻¹⁰ 24.52 141003-2 PC 68.0 6887 11.4 4.91×10 ⁻¹⁰ 25.19 141003-3 PC 68.0 4003 9.5 1.45×10 ⁻⁹ 21.00 141003-4 PC 68.0 1689 6.8 8.16×10 ⁻⁹ 14.97 141003-5 PC 68.0 3016 8.4 2.56×10 ⁻⁹ 18.62 141202-1	140720-4	PC	68.0	189	3.0	6.49×10 ⁻⁷	6.58
141002-1 PC 68.0 1506 6.5 1.03×10 ⁻⁸ 14.32 141002-2 PC 68.0 2495 7.6 3.74×10 ⁻⁹ 16.80 141002-3 PC 68.0 3268 8.7 2.18×10 ⁻⁹ 19.08 141002-4 PC 68.0 4417 9.3 1.19×10 ⁻⁹ 20.51 141002-5 PC 68.0 5754 10.3 7.03×10 ⁻¹⁰ 22.76 141003-1 PC 68.0 6711 11.1 5.17×10 ⁻¹⁰ 24.52 141003-2 PC 68.0 6887 11.4 4.91×10 ⁻¹⁰ 25.19 141003-3 PC 68.0 4003 9.5 1.45×10 ⁻⁹ 21.00 141003-4 PC 68.0 1689 6.8 8.16×10 ⁻⁹ 14.97 141003-5 PC 68.0 3016 8.4 2.56×10 ⁻⁹ 18.62 141202-1 PC 68.0 6158 10.8 6.14×10 ⁻¹⁰ 23.73 141202-2	140720-5	PC	68.0	205	3.1	5.52×10 ⁻⁷	6.75
141002-2 PC 68.0 2495 7.6 3.74×10 ⁻⁹ 16.80 141002-3 PC 68.0 3268 8.7 2.18×10 ⁻⁹ 19.08 141002-4 PC 68.0 4417 9.3 1.19×10 ⁻⁹ 20.51 141002-5 PC 68.0 5754 10.3 7.03×10 ⁻¹⁰ 22.76 141003-1 PC 68.0 6711 11.1 5.17×10 ⁻¹⁰ 24.52 141003-2 PC 68.0 6887 11.4 4.91×10 ⁻¹⁰ 25.19 141003-3 PC 68.0 4003 9.5 1.45×10 ⁻⁹ 21.00 141003-4 PC 68.0 1689 6.8 8.16×10 ⁻⁹ 14.97 141003-5 PC 68.0 3016 8.4 2.56×10 ⁻⁹ 18.62 141202-1 PC 68.0 6158 10.8 6.14×10 ⁻¹⁰ 23.75 141202-2 PC 68.0 6596 10.8 5.35×10 ⁻¹⁰ 23.73 141203-1 <td>141001-1</td> <td>PC</td> <td>68.0</td> <td>1739</td> <td>6.6</td> <td>7.70×10⁻⁹</td> <td>14.65</td>	141001-1	PC	68.0	1739	6.6	7.70×10 ⁻⁹	14.65
141002-3 PC 68.0 3268 8.7 2.18×10 ⁻⁹ 19.08 141002-4 PC 68.0 4417 9.3 1.19×10 ⁻⁹ 20.51 141002-5 PC 68.0 5754 10.3 7.03×10 ⁻¹⁰ 22.76 141003-1 PC 68.0 6711 11.1 5.17×10 ⁻¹⁰ 24.52 141003-2 PC 68.0 6887 11.4 4.91×10 ⁻¹⁰ 25.19 141003-3 PC 68.0 4003 9.5 1.45×10 ⁻⁹ 21.00 141003-4 PC 68.0 1689 6.8 8.16×10 ⁻⁹ 14.97 141003-5 PC 68.0 3016 8.4 2.56×10 ⁻⁹ 18.62 141202-1 PC 68.0 6158 10.8 6.14×10 ⁻¹⁰ 23.75 141202-2 PC 68.0 6596 10.8 5.35×10 ⁻¹⁰ 23.73 141203-1 PC 68.0 5192 10.3 8.63×10 ⁻¹⁰ 22.60 141203-2<	141002-1	PC	68.0	1506	6.5	1.03×10 ⁻⁸	14.32
141002-4 PC 68.0 4417 9.3 1.19×10 ⁻⁹ 20.51 141002-5 PC 68.0 5754 10.3 7.03×10 ⁻¹⁰ 22.76 141003-1 PC 68.0 6711 11.1 5.17×10 ⁻¹⁰ 24.52 141003-2 PC 68.0 6887 11.4 4.91×10 ⁻¹⁰ 25.19 141003-3 PC 68.0 4003 9.5 1.45×10 ⁻⁹ 21.00 141003-4 PC 68.0 1689 6.8 8.16×10 ⁻⁹ 14.97 141003-5 PC 68.0 3016 8.4 2.56×10 ⁻⁹ 18.62 141202-1 PC 68.0 6158 10.8 6.14×10 ⁻¹⁰ 23.75 141202-2 PC 68.0 6596 10.8 5.35×10 ⁻¹⁰ 23.73 141203-3 PC 68.0 5192 10.3 8.63×10 ⁻¹⁰ 24.08 141203-2 PC 68.0 5330 9.9 8.19×10 ⁻¹⁰ 21.85 141203-3	141002-2	PC	68.0	2495	7.6	3.74×10 ⁻⁹	16.80
141002-5 PC 68.0 5754 10.3 7.03×10 ⁻¹⁰ 22.76 141003-1 PC 68.0 6711 11.1 5.17×10 ⁻¹⁰ 24.52 141003-2 PC 68.0 6887 11.4 4.91×10 ⁻¹⁰ 25.19 141003-3 PC 68.0 4003 9.5 1.45×10 ⁻⁹ 21.00 141003-4 PC 68.0 1689 6.8 8.16×10 ⁻⁹ 14.97 141003-5 PC 68.0 3016 8.4 2.56×10 ⁻⁹ 18.62 141202-1 PC 68.0 6158 10.8 6.14×10 ⁻¹⁰ 23.75 141202-2 PC 68.0 6596 10.8 5.35×10 ⁻¹⁰ 23.73 141203-3 PC 68.0 6402 10.9 5.68×10 ⁻¹⁰ 24.08 141203-2 PC 68.0 5330 9.9 8.19×10 ⁻¹⁰ 21.85 141203-3 PC 68.0 2385 7.5 4.09×10 ⁻⁹ 16.62	141002-3	PC	68.0	3268	8.7	2.18×10 ⁻⁹	19.08
141003-1 PC 68.0 6711 11.1 5.17×10^{-10} 24.52 141003-2 PC 68.0 6887 11.4 4.91×10^{-10} 25.19 141003-3 PC 68.0 4003 9.5 1.45×10^{-9} 21.00 141003-4 PC 68.0 1689 6.8 8.16×10^{-9} 14.97 141003-5 PC 68.0 3016 8.4 2.56×10^{-9} 18.62 141202-1 PC 68.0 6158 10.8 6.14×10^{-10} 23.75 141202-2 PC 68.0 6596 10.8 5.35×10^{-10} 23.73 141202-3 PC 68.0 6402 10.9 5.68×10^{-10} 24.08 141203-1 PC 68.0 5192 10.3 8.63×10^{-10} 22.60 141203-2 PC 68.0 5330 9.9 8.19×10^{-10} 21.85 141203-3 PC 68.0 2385 7.5 4.09×10^{-9}	141002-4	PC	68.0	4417	9.3	1.19×10 ⁻⁹	20.51
141003-2 PC 68.0 6887 11.4 4.91×10 ⁻¹⁰ 25.19 141003-3 PC 68.0 4003 9.5 1.45×10 ⁻⁹ 21.00 141003-4 PC 68.0 1689 6.8 8.16×10 ⁻⁹ 14.97 141003-5 PC 68.0 3016 8.4 2.56×10 ⁻⁹ 18.62 141202-1 PC 68.0 6158 10.8 6.14×10 ⁻¹⁰ 23.75 141202-2 PC 68.0 6596 10.8 5.35×10 ⁻¹⁰ 23.73 141202-3 PC 68.0 6402 10.9 5.68×10 ⁻¹⁰ 24.08 141203-1 PC 68.0 5192 10.3 8.63×10 ⁻¹⁰ 22.60 141203-2 PC 68.0 5330 9.9 8.19×10 ⁻¹⁰ 21.85 141203-3 PC 68.0 2385 7.5 4.09×10 ⁻⁹ 16.62	141002-5	PC	68.0	5754	10.3	7.03×10 ⁻¹⁰	22.76
141003-3 PC 68.0 4003 9.5 1.45×10 ⁻⁹ 21.00 141003-4 PC 68.0 1689 6.8 8.16×10 ⁻⁹ 14.97 141003-5 PC 68.0 3016 8.4 2.56×10 ⁻⁹ 18.62 141202-1 PC 68.0 6158 10.8 6.14×10 ⁻¹⁰ 23.75 141202-2 PC 68.0 6596 10.8 5.35×10 ⁻¹⁰ 23.73 141202-3 PC 68.0 6402 10.9 5.68×10 ⁻¹⁰ 24.08 141203-1 PC 68.0 5192 10.3 8.63×10 ⁻¹⁰ 22.60 141203-2 PC 68.0 5330 9.9 8.19×10 ⁻¹⁰ 21.85 141203-3 PC 68.0 2385 7.5 4.09×10 ⁻⁹ 16.62	141003-1	PC	68.0	6711	11.1	5.17×10 ⁻¹⁰	24.52
141003-4 PC 68.0 1689 6.8 8.16×10 ⁻⁹ 14.97 141003-5 PC 68.0 3016 8.4 2.56×10 ⁻⁹ 18.62 141202-1 PC 68.0 6158 10.8 6.14×10 ⁻¹⁰ 23.75 141202-2 PC 68.0 6596 10.8 5.35×10 ⁻¹⁰ 23.73 141202-3 PC 68.0 6402 10.9 5.68×10 ⁻¹⁰ 24.08 141203-1 PC 68.0 5192 10.3 8.63×10 ⁻¹⁰ 22.60 141203-2 PC 68.0 5330 9.9 8.19×10 ⁻¹⁰ 21.85 141203-3 PC 68.0 2385 7.5 4.09×10 ⁻⁹ 16.62	141003-2	PC	68.0	6887	11.4	4.91×10 ⁻¹⁰	25.19
141003-5 PC 68.0 3016 8.4 2.56×10 ⁻⁹ 18.62 141202-1 PC 68.0 6158 10.8 6.14×10 ⁻¹⁰ 23.75 141202-2 PC 68.0 6596 10.8 5.35×10 ⁻¹⁰ 23.73 141202-3 PC 68.0 6402 10.9 5.68×10 ⁻¹⁰ 24.08 141203-1 PC 68.0 5192 10.3 8.63×10 ⁻¹⁰ 22.60 141203-2 PC 68.0 5330 9.9 8.19×10 ⁻¹⁰ 21.85 141203-3 PC 68.0 2385 7.5 4.09×10 ⁻⁹ 16.62	141003-3	PC	68.0	4003	9.5	1.45×10 ⁻⁹	21.00
141202-1 PC 68.0 6158 10.8 6.14×10^{-10} 23.75 141202-2 PC 68.0 6596 10.8 5.35×10^{-10} 23.73 141202-3 PC 68.0 6402 10.9 5.68×10^{-10} 24.08 141203-1 PC 68.0 5192 10.3 8.63×10^{-10} 22.60 141203-2 PC 68.0 5330 9.9 8.19×10^{-10} 21.85 141203-3 PC 68.0 2385 7.5 4.09×10^{-9} 16.62	141003-4	PC	68.0	1689	6.8	8.16×10 ⁻⁹	14.97
141202-2 PC 68.0 6596 10.8 5.35×10^{-10} 23.73 141202-3 PC 68.0 6402 10.9 5.68×10^{-10} 24.08 141203-1 PC 68.0 5192 10.3 8.63×10^{-10} 22.60 141203-2 PC 68.0 5330 9.9 8.19×10^{-10} 21.85 141203-3 PC 68.0 2385 7.5 4.09×10^{-9} 16.62	141003-5	PC	68.0	3016	8.4	2.56×10 ⁻⁹	18.62
141202-3 PC 68.0 6402 10.9 5.68×10 ⁻¹⁰ 24.08 141203-1 PC 68.0 5192 10.3 8.63×10 ⁻¹⁰ 22.60 141203-2 PC 68.0 5330 9.9 8.19×10 ⁻¹⁰ 21.85 141203-3 PC 68.0 2385 7.5 4.09×10 ⁻⁹ 16.62	141202-1	PC	68.0	6158	10.8	6.14×10 ⁻¹⁰	23.75
141203-1 PC 68.0 5192 10.3 8.63×10 ⁻¹⁰ 22.60 141203-2 PC 68.0 5330 9.9 8.19×10 ⁻¹⁰ 21.85 141203-3 PC 68.0 2385 7.5 4.09×10 ⁻⁹ 16.62	141202-2	PC	68.0	6596	10.8	5.35×10 ⁻¹⁰	23.73
141203-2 PC 68.0 5330 9.9 8.19×10 ⁻¹⁰ 21.85 141203-3 PC 68.0 2385 7.5 4.09×10 ⁻⁹ 16.62	141202-3	PC	68.0	6402	10.9	5.68×10 ⁻¹⁰	24.08
141203-3 PC 68.0 2385 7.5 4.09×10 ⁻⁹ 16.62	141203-1	PC	68.0	5192	10.3	8.63×10 ⁻¹⁰	22.60
	141203-2	PC	68.0	5330	9.9	8.19×10 ⁻¹⁰	21.85
141203-4 PC 68.0 2254 7.6 4.58×10 ⁻⁹ 16.66	141203-3	PC	68.0	2385	7.5	4.09×10 ⁻⁹	16.62
171205-7 10 00.0 2254 7.0 4.50^10 10.00	141203-4	PC	68.0	2254	7.6	4.58×10 ⁻⁹	16.66

141204-1	PC	68.0	2317	7.3	4.34×10 ⁻⁹	16.16
150819-1	PC	68.0	4322	9.7	1.25×10 ⁻⁹	21.33
*150819-3	PC	68.0	4230	8.9	1.30×10 ⁻⁹	19.52
*151104-1	PC	68.0	2208	6.9	4.77×10 ⁻⁹	15.17
*151104-2	PC	68.0	6150	10.1	6.15×10 ⁻¹⁰	22.29
*151104-3	PC	68.0	1125	5.6	1.84×10 ⁻⁸	12.32
151105-2	PC	68.0	1206	5.8	1.60×10 ⁻⁸	12.87
#150309-1	Al	11.6	4124	5.6	5.76×10 ⁻¹⁰	22.14
#150310-1	Al	11.6	2077	4.4	2.27×10 ⁻⁹	17.62
#150310-2	Al	11.6	2059	4.5	2.31×10 ⁻⁹	17.67
#150610-1	Al	11.6	1994	4.2	2.46×10 ⁻⁹	16.52
150611-2	Al	11.7	4936	6.1	4.02×10 ⁻¹⁰	24.14
150611-3	Al	11.7	5015	6.3	3.90×10 ⁻¹⁰	24.88
#150612-1	Al	11.9	1815	4.0	2.97×10 ⁻⁹	15.70
#150612-2	Al	11.5	2062	4.4	2.30×10 ⁻⁹	17.60
150820-1	Al	11.7	2920	4.7	1.15×10 ⁻⁹	18.77
#150820-7	Al	11.8	3953	5.4	6.27×10 ⁻¹⁰	21.43
*,#151105-3	Al	11.7	4143	5.3	5.71×10 ⁻¹⁰	20.94
160721-2	Al	11.9	5622	6.1	3.10×10 ⁻¹⁰	24.11
150611-5	Gl	10.6	5318	5.8	3.47×10 ⁻¹⁰	23.93
150611-6	Gl	10.6	2281	4.4	1.88×10 ⁻⁹	18.19
#150612-5	Gl	11.3	2023	4.2	2.39×10 ⁻⁹	16.86
*,#150820-2	Gl	10.9	4023	5.2	6.06×10 ⁻¹⁰	21.24

150821-1	Gl	10.7	3427	5.1	8.34×10 ⁻¹⁰	20.77
#150821-2	Gl	10.6	4013	5.3	6.09×10 ⁻¹⁰	21.58
#150311-1	Ti	18.9	4097	6.5	5.84×10 ⁻¹⁰	22.00
*,#150820-3	Ti	18.9	3834	5.9	6.67×10 ⁻¹⁰	20.06
#150821-3	Ti	18.8	2044	5.1	2.35×10 ⁻⁹	17.37
#150821-4	Ti	18.8	1972	5.0	2.52×10 ⁻⁹	16.84
160720-1	Ti	18.8	5181	6.7	3.65×10 ⁻¹⁰	22.68
#150309-3	ZrO ₂	26.5	3864	6.9	6.56×10 ⁻¹⁰	20.79
#150311-7	ZrO ₂	25.7	2004	5.7	2.44×10 ⁻⁹	17.31
*,#150820-4	ZrO ₂	25.4	3934	6.7	6.33×10 ⁻¹⁰	20.42
#160719-3	ZrO ₂	26.9	2153	5.4	2.11×10 ⁻⁹	16.26
160720-2	ZrO ₂	24.2	5727	7.3	2.99×10 ⁻¹⁰	22.85
#150309-2	SUS	32.4	3943	7.5	6.30×10 ⁻¹⁰	21.14
#150310-3	SUS	32.6	2018	5.7	2.41×10 ⁻⁹	16.01
#150610-2	SUS	32.6	1841	5.7	2.89×10 ⁻⁹	16.12
150611-4	SUS	32.6	4936	8.4	4.02×10 ⁻¹⁰	23.55
#150612-3	SUS	32.6	1848	5.7	2.87×10 ⁻⁹	15.98
#150612-4	SUS	32.8	1824	5.8	2.95×10 ⁻⁹	16.18
*,#150820-6	SUS	32.6	3805	7.1	6.77×10 ⁻¹⁰	20.07
#150310-6	Cu	37.8	3962	7.6	6.24×10 ⁻¹⁰	20.32
*150820-5	Cu	38.1	3626	7.2	7.45×10 ⁻¹⁰	19.33
#151105-1	Cu	37.8	2114	6.0	2.19×10 ⁻⁹	16.01
160719-1	Cu	38.2	2238	6.2	1.96×10 ⁻⁹	16.60

160720-4	Cu	37.9	5348	7.9	3.43×10 ⁻¹⁰	21.10
160721-3	Cu	37.9	2306	6.3	1.84×10 ⁻⁹	16.75
150309-4	WC	62.1	3726	9.1	7.06×10^{-10}	20.63
150312-2	WC	61.9	1786	6.8	3.07×10 ⁻⁹	15.47
*150819-4	WC	62.5	3422	8.3	8.37×10 ⁻¹⁰	18.82

- The asterisk (*) indicates the example of the crater profile shown on Fig. 3.
- The sharp (#) indicates the data calculating by Eq. (5) shown on Fig. 5.
- m_p : projectile mass
- v_i : impact velocity
- R_{rim} : crater rim radius
- π_2 : a scaling parameter defined by Eq. (2)
- π_R : a scaling parameter defined by Eq. (1)

Table 3. Experimental conditions and results of a PC projectile at low impact velocity.

Run number	v_i , m s ⁻¹	$E_{\mathbf{k}}$, J	$R_{\rm rim}$, cm	x, cm	$x/R_{\rm rim}$	$t_{ m max}$, ms	$g_{ m max}$, m s ⁻²	$T_{ m half}$, ms
140710-2	221	1.7	3.0	5.1	1.6	0.42	30.0	1.13
				7.5	2.5	0.87	7.1	0.63
140715-2	222	1.67	3.2	6.0	1.8	0.39	35.5	0.79
140718-2	208	1.5	3.1	5.7	1.7	0.87	18.1	1.52
				6.6	2.0	0.70	12.4	1.02
				7.2	2.2	1.01	6.75	1.66
140718-3	204	1.4	3.2	8.8	2.7	1.32	7.13	1.59
140718-4	217	1.6	3.3	9.4	2.8	1.31	9.6	0.78
140720-1	173	1.0	2.9	6.4	2.1	0.66	10.5	1.02

1002 v_i : Impact velocity

1003 E_k : Kinetic energy of the projectile

 R_{rim} : Crater rim radius

1005 x: Distance from the impact point to the accelerometer

 t_{max} : Traveling time from the impact time to the time when the acceleration becomes the maximum

1007 g_{max} : Maximum acceleration

 T_{half} , Duration of the first upward acceleration

Table 4. Experimental conditions and results of a PC projectile at high impact velocity.

Run number	$v_{\rm i}$, m s ⁻¹	<i>E</i> _k , J	$R_{\rm rim}$, cm	x, cm	$x/R_{\rm rim}$	$t_{ m max},{ m ms}$	$g_{ m max},{ m m~s}^{-2}$	$T_{ m half}$, ms
140530-1	2279	176.6	7.6	10.3	1.4	0.93	67.3	1.25
				16.2	2.1	1.74	17.0	1.44
				25.1	3.3	3.07	2.8	1.66
140530-2	3748	477.6	8.7	12.3	1.4	0.94	116.4	1.02
				17.2	2.0	1.83	27.6	1.90
				25.1	2.9	2.81	8.2	1.50
140530-3	5155	903.5	10.2	15.2	1.5	1.26	116.2	1.24
				20.2	2.0	2.03	22.0	1.28
				27.1	2.7	3.49	15.3	2.33
140530-4	3381	388.7	8.9	15.2	1.7	1.64	57.2	0.89
				20.2	2.3	2.41	21.9	1.55
				30.1	3.4	3.59	9.7	1.32
141001-1	1739	102.8	6.6	10.7	1.6	0.97	45.6	1.77
				14.0	2.1	1.56	25.1	1.64
141002-1	1506	77.1	6.5	8.9	1.4	_	35.0	1.17
				12.0	1.9	_	30.7	1.81
				15.2	2.3	_	9.7	1.52
141002-2	2495	211.7	7.6	12.1	1.6	1.05	56.9	2.48
				14.8	1.9	1.72	30.1	2.46
141002-3	3268	363.1	8.6	18.9	2.2	2.30	28.4	2.04

141002-4	4417	663.3	9.3	16.6	1.8	1.77	44.0	2.53
				20.6	2.2	2.56	13.9	2.57
141002-5	5754	1125.7	10.3	15.4	1.5	1.58	69.5	1.94
				20.2	2.0	2.38	13.8	1.68
141003-1	6711	1531.3	11.1	15.0	1.3	1.27	77.7	2.46
				20.6	1.9	2.16	34.3	2.66
141003-2	-2 6887 1612.7	1612.7	11.4	13.3	1.2	0.97	118.2	1.23
				17.2	1.5	1.59	79.7	2.05
				23.8	2.1	3.10	21.6	2.24
141003-3	1003-3 4003 544.8	544.8	9.5	13.6	1.4	1.25	63.7	1.87
				17.6	1.8	2.10	25.6	1.88
				23.6	2.5	3.09	5.6	2.28
141003-4	1689	97.0	6.8	7.9	1.2	0.42	46.4	1.78
				15.3	2.2	1.99	28.9	2.10
				23.1	3.4	3.07	5.6	2.18
141003-5	3016	309.3	8.4	10.1	1.2	0.81	53.0	1.75
				15.7	1.9	2.12	28.8	2.09
				19.9	2.4	2.98	11.4	2.48
141202-1	6158	1289.3	10.8	20.1	1.9	1.97	30.0	1.94
				23.1	2.1	2.42	14.2	2.15
141202-2	6596	1479.3	10.8	15.3	1.4	1.67	47.8	2.61
				17.6	1.6	1.97	34.5	2.60

				24.6	2.3	3.59	14.2	2.40
141202-3	6402	1393.5	10.9	14.6	1.3	1.85	35.9	2.27
				19.8	1.8	2.46	33.5	2.16
				24.2	2.2	3.78	15.7	2.15
141203-1	5192	916.5	10.2	14.3	1.4	1.84	53.5	1.99
				17.1	1.7	2.12	27.9	2.04
				23.4	2.3	3.51	16.1	1.55
141203-2	5330	965.9	9.9	12.1	1.2	1.13	77.8	2.17
				16.7	1.7	2.20	26.0	2.52
				19.6	2.0	2.91	20.9	2.29
				19.9	2.0	2.91	10.5	2.60
150819-1	4322	635.1	9.7	11.6	1.2	0.88	86.6	1.91
				17.8	1.8	2.03	27.6	1.95
				23.6	2.4	2.88	14.8	1.78
				26.0	2.7	3.38	9.1	1.61
150819-3	4230	608.4	8.9	12.9	1.5	1.27	68.0	2.13
				15.7	1.8	1.86	36.0	2.28
				20.6	2.3	2.62	19.7	2.44
				26.2	3.0	3.75	11.2	2.21
151104-1	2208	165.8	6.9	12.4	1.8	1.51	47.3	1.48
				13.9	2.0	1.89	27.3	1.89
				15.2	2.2	2.24	19.3	1.52

_	-	-	-					
				19.2	2.8	2.74	9.7	1.58
				23.3	3.4	3.27	3.7	2.14
151104-3	1125	43.0	5.6	12.1	2.2	1.53	7.9	1.66
				15.6	2.8	2.13	3.6	2.66
				19.2	3.4	2.78	2.8	1.97
151105-2	1206	49.5	5.8	10.3	1.8	0.96	6.9	2.94
				13.0	2.2	1.48	6.5	2.05
				16.5	2.8	2.13	5.3	1.72
				18.3	3.1	2.52	5.2	1.74
				21.5	3.7	3.06	4.0	1.54

Table 5. Experimental conditions and results of 2-mm projectiles with different densities.

Run number	Projectile	v_i , m s ⁻¹	$E_{\mathbf{k}}, \mathbf{J}$	$R_{\rm rim}$, cm	x, cm	$x/R_{\rm rim}$	$t_{ m max}$, ms	$g_{ m max}$, m s ⁻²	$T_{ m half}$, ms
150611-5	Gl	5318	150.3	5.8	11.1	1.9	1.45	14.7	2.39
					15.1	2.6	2.29	5.6	2.19
					21.0	3.6	3.42	1.5	2.07
					23.0	3.9	3.36	1.6	1.82
150611-6	Gl	2281	28.0	4.4	13.1	3.0	2.16	3.1	2.12
					18.2	4.1	2.37	0.5	1.49
					23.3	5.2	3.59	0.5	1.77
					23.9	5.4	4.17	0.4	1.58
150612-5	Gl	2023	23.1	4.2	9.2	2.2	1.45	8.9	2.13
					17.3	4.1	2.39	0.5	1.11
150820-2	Gl	4023	88.2	5.0	14.1	2.8	2.07	7.1	2.10
					17.9	3.6	2.77	5.0	1.94
					21.9	4.4	3.43	2.4	1.85
150821-1	Gl	3427	62.5	5.0	14.2	2.8	1.96	5.0	1.96
					16.2	3.2	2.49	5.1	1.93
					23.2	4.6	3.58	0.9	1.87
150821-2	Gl	4013	85.5	5.1	13.2	2.6	1.93	7.4	2.81
					16.7	3.3	2.66	4.0	2.29
					20.4	4.0	3.33	2.3	2.15
150309-1	Al	4124	98.8	5.7	9.6	1.7	0.99	36.2	1.82

					12.7	2.2	1.97	12.4	1.40
					12.8	2.2	1.49	11.1	1.18
					14.8	2.6	2.01	8.0	1.13
					19.3	3.4	2.48	5.4	1.29
150310-2	Al	2059	24.7	4.4	10.6	2.4	1.86	7.9	1.40
					21.9	4.9	3.85	0.9	1.60
150610-1	Al	1994	23.1	4.2	7.9	1.9	0.86	12.7	0.99
					9.9	2.4	1.13	11.3	1.07
					13.9	3.3	1.87	3.8	1.36
					17.3	4.2	2.16	2.5	1.45
					24.2	5.8	3.61	1.4	1.42
150611-2	Al	4936	142.3	6.1	13.2	2.2	1.78	7.5	1.88
					17.7	2.9	2.49	4.1	1.74
					20.1	3.3	2.65	2.1	1.81
					25.6	4.2	3.48	1.1	1.64
150611-3	Al	5015	146.9	6.3	13.5	2.2	1.70	8.2	1.72
					18.3	2.9	2.53	4.2	1.25
					20.5	3.3	2.72	3.1	1.54
150612-1	Al	1815	19.6	4.0	16.0	4.0	2.25	1.5	1.40
					22.2	5.6	3.00	0.7	1.33
150612-2	Al	2062	24.5	4.4	21.9	5.0	2.69	0.3	1.52
150820-1	Al	2920	49.9	4.6	12.0	2.6	1.33	14.5	1.83
					15.3	3.3	1.92	5.8	1.66

					18.5	4.0	2.68	5.6	1.79
150820-7	Al	3953	92.0	5.2	12.2	2.3	1.76	8.0	1.88
					17.5	3.3	2.68	2.7	2.14
151105-3	Al	4143	100.3	5.4	10.7	2.0	1.30	13.6	2.34
					13.0	2.4	2.01	6.2	1.72
					15.4	2.8	2.46	3.5	1.68
					18.1	3.3	2.82	3.8	1.73
					22.4	4.1	3.33	2.6	1.53
150311-1	Ti	4097	159.0	6.5	13.4	2.1	1.88	10.7	2.38
					15.9	2.4	2.32	5.1	2.51
					18.1	2.8	2.84	6.5	2.08
					20.5	3.1	3.50	1.8	1.66
150820-3	Ti	3834	138.8	5.8	12.1	2.1	_	16.5	2.53
					18.4	3.2	_	9.8	2.06
					24.6	4.2	_	3.5	1.73
150821-5	Ti	1972	36.6	4.8	14.0	2.9	2.04	4.0	1.48
					15.1	3.2	2.39	4.1	1.92
					19.8	4.2	3.22	1.8	1.89
150309-3	ZrO ₂	3864	197.8	6.9	10.0	1.4	0.94	24.0	2.43
					15.3	2.2	2.21	15.8	2.31
					18.2	2.6	2.67	4.4	2.03
					19.1	2.8	2.82	6.2	1.82
					22.6	3.3	3.61	3.2	1.67

150311-7	ZrO ₂	2004	51.5	5.7	12.5	2.2	1.83	12.7	1.80
150820-4	ZrO ₂	3934	196.5	6.6	11.7	1.8	1.23	15.4	2.76
					15.0	2.3	1.94	7.5	2.53
					17.2	2.6	2.51	12.2	2.09
					22.1	3.3	3.21	5.3	1.77
150309-2	SUS	3943	252.1	7.5	14.9	2.0	2.36	12.9	2.28
					16.4	2.2	2.73	8.7	2.43
					17.7	2.4	2.67	8.8	2.06
					21.1	2.8	3.56	3.5	1.98
150310-3	SUS	2018	66.4	5.7	17.4	3.1	3.19	3.3	2.10
150610-2	SUS	1841	55.2	5.7	11.6	2.0	1.78	11.7	1.87
					15.9	2.8	2.53	3.3	2.05
					21.2	3.7	3.10	1.2	1.60
					25.3	4.4	3.80	1.1	1.71
150611-4	SUS	4936	397.5	8.4	15.6	1.9	1.85	14.9	2.21
					18.1	2.2	2.77	7.7	2.09
					21.6	2.6	3.32	4.6	2.52
					24.8	3.0	3.92	7.6	2.28
150612-3	SUS	1848	55.6	5.7	15.6	2.7	2.25	6.6	1.94
					21.6	3.8	3.27	2.0	1.82
150612-4	SUS	1824	54.5	5.8	11.8	2.1	1.64	9.1	1.95
150820-6	SUS	3805	235.6	7.0	12.7	1.8	1.44	26.0	2.71
					17.3	2.5	2.49	22.1	2.17

					21.0	3.0	3.25	8.2	1.82
					24.0	3.4	3.51	6.5	1.55
150310-6	Cu	3962	296.4	7.6	14.8	2.0	1.98	15.4	2.00
					17.1	2.3	2.55	14.1	2.36
					20.0	2.6	2.88	7.7	2.00
					23.4	3.1	3.61	6.0	1.50
150820-5	Cu	3626	250.1	7.1	13.9	2.0	1.45	20.0	2.20
					18.4	2.6	2.76	14.2	1.96
					25.0	3.5	3.38	7.5	1.67
151105-1	Cu	2114	84.5	6.0	9.9	1.6	0.87	28.5	1.09
					14.1	2.3	1.54	10.0	1.49
					17.2	2.9	2.31	5.8	1.15
					19.7	3.3	2.75	7.1	1.20
					24.8	4.1	3.41	3.1	1.44
150309-4	WC	3726	430.7	9.1	14.0	1.5	1.42	40.9	2.06
					15.8	1.7	1.87	37.0	1.99
					20.6	2.3	2.79	13.0	1.99
					21.2	2.3	2.81	12.9	2.02
					24.4	2.7	3.76	8.4	2.19
150312-2	WC	1786	98.8	6.4	12.2	1.9	2.13	12.4	2.70
					15.8	2.5	3.39	8.5	2.80
150819-4	WC	3422	366.2	8.2	12.7	1.6	1.16	46.3	2.89
					15.6	1.9	1.68	16.5	2.95

	18.5	2.3	2.46	20.8	2.65
	26.2	3.2	3.75	6.8	2.05

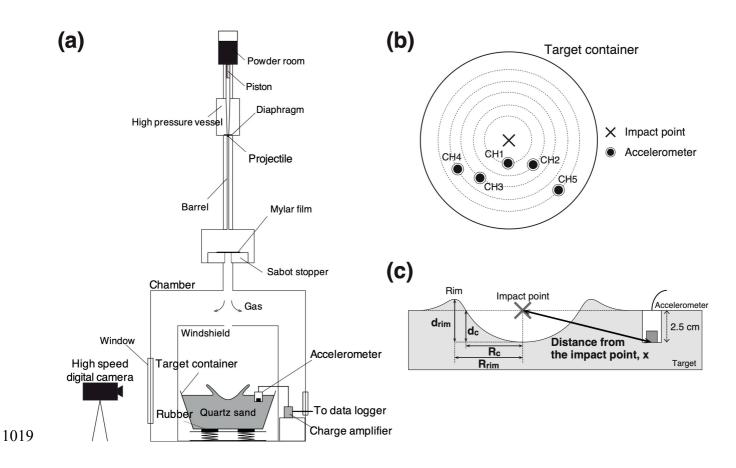
Impact velocity v_i ,	Propagation velocity V_{prop} ,	Average crater rim radius	Eq. (7) related to maximum acceleration g_{max}	
$km s^{-1}$	$m s^{-1}$	_		_
KIII S	111 5	$R_{\rm rim_ave}$, cm	$a_{\rm r}$	$b_{ m r}$
0.2	37.4 (7.3)	3.1	-1.68 (0.88)	2.38 (0.76)
1	53.4 (2.9)	5.7	-0.18 (0.30)	1.08 (0.36)
1.5	56.4 (3.9)	6.6	-0.35 (0.41)	1.92 (0.46)
2	57.2 (7.1)	7.6	-1.77 (0.16)	3.78 (0.20)
3	65.0 (10.2)	8.9	0.09 (0.37)	1.75 (0.48)
4	56.5 (3.0)	9.0	-0.92 (0.19)	3.14 (0.25)
5	47.9 (8.1)	10.1	-0.46 (0.49)	2.59 (0.66)
6	40.7 (12.9)	10.7	-0.39 (0.60)	2.52 (0.84)
7	45.3 (4.5)	11.1	-0.54 (0.38)	2.93 (0.51)

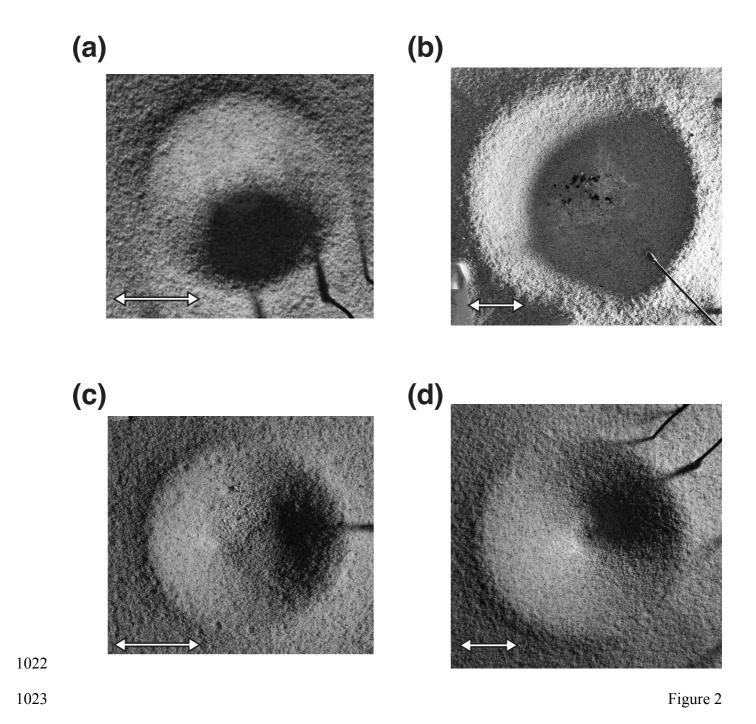
1016 The number in the parentheses indicates the standard error.

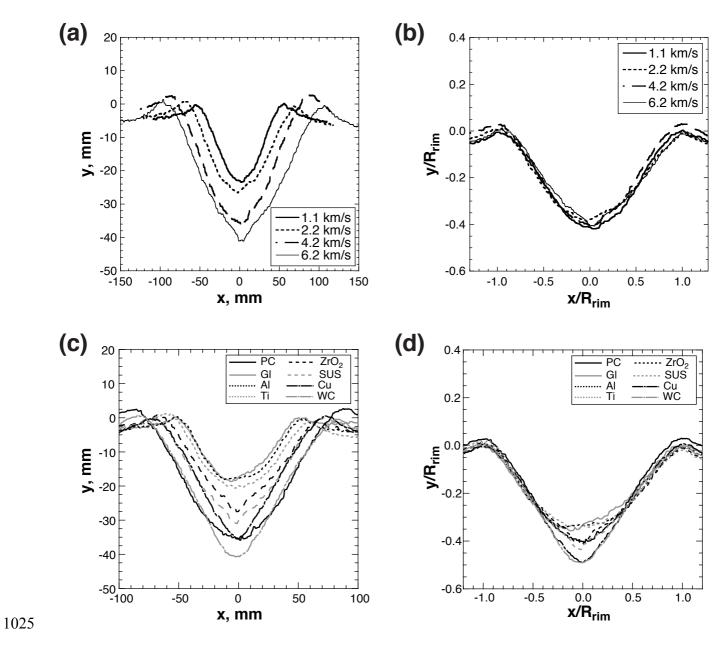
Table 7. Propagation velocity, parameters related to maximum acceleration on Eq. (7) for 2-mm projectiles with different densities.

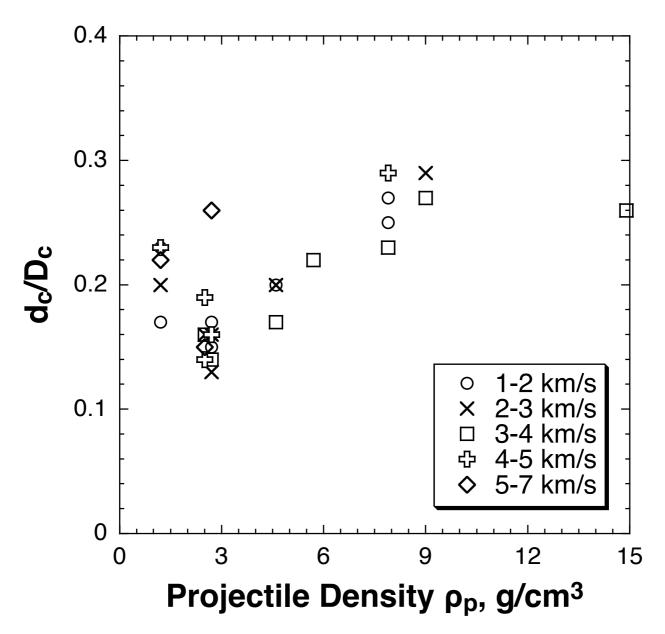
	Impact	Propagation	Average crater	Eq. (7) related to maximum	
Projectile	velocity v_i ,	velocity V_{prop} ,	rim radius	acceleration g_{\max}	
	$km s^{-1}$	$\mathrm{m}\;\mathrm{s}^{-1}$	$R_{\rm rim_ave}$, cm	$a_{\rm r}$	$b_{ m r}$
Al	2	56.1 (7.5)	4.2	-2.08 (0.39)	3.00 (0.48)
	3	47.5 (4.0)	4.6	-0.98 (0.86)	2.27 (1.03)
	4	53.4 (5.6)	5.5	-1.42 (0.38)	2.70 (0.45)
	5	70.6 (3.9)	6.2	-1.56 (0.25)	2.84 (0.33)
Gl	2	54.0 (9.1)	4.3	-2.64 (0.45)	3.41 (0.57)
	3	56.6 (6.7)	5.0	-2.33 (0.77)	3.68 (1.01)
	4	55.0 (3.0)	5.1	-1.21 (0.32)	2.40 (0.41)
	5	56.8 (7.6)	5.8	-1.95 (0.25)	3.26 (0.31)
Ti	2	50.3 (6.0)	4.8	-1.43 (0.43)	2.43 (0.54)
	4	43.0 (3.3)	7.3	-1.01 (0.61)	2.35 (0.78)
ZrO ₂	4	48.9 (2.4)	6.8	-0.76 (0.35)	2.15 (0.44)
SUS	2	60.4 (7.7)	5.7	-1.81 (0.28)	3.04 (0.36)
	4	49.8 (7.3)	7.3	-0.81 (0.62)	2.41 (0.83)
	5	45.2 (6.0)	8.4	-0.27 (0.84)	1.66 (1.19)
Cu	2	56.4 (3.8)	6.0	-0.87 (0.26)	2.27 (0.33)
	4	52.4 (7.7)	7.3	-0.34 (0.26)	1.90 (0.35)
WC	3	51.5 (3.1)	8.2	-0.54 (0.51)	2.39 (0.66)
	4	46.1 (2.9)	9.1	-0.96 (0.20)	3.06 (0.27)

¹⁰¹⁸ The number in the parentheses indic1ates the standard error.

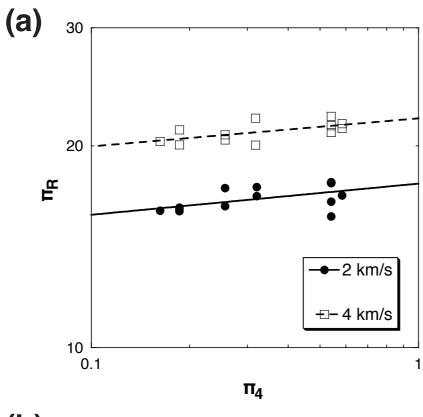








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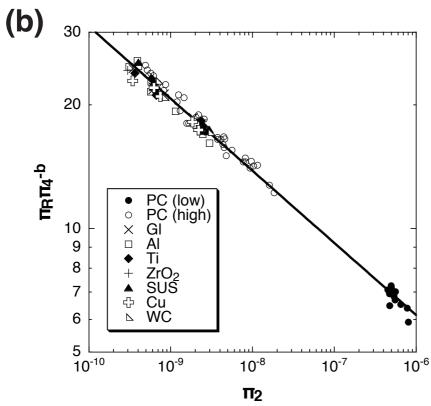
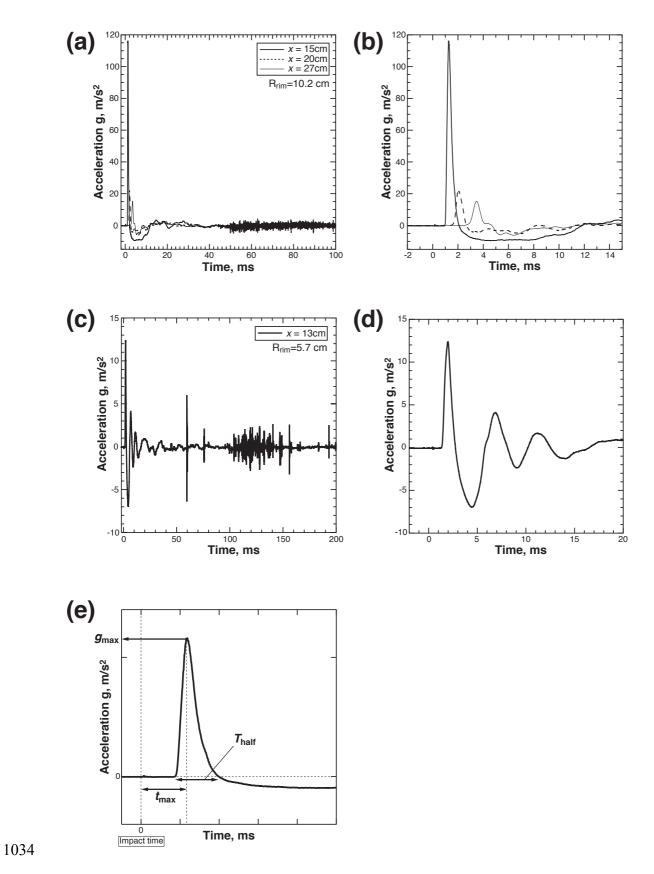
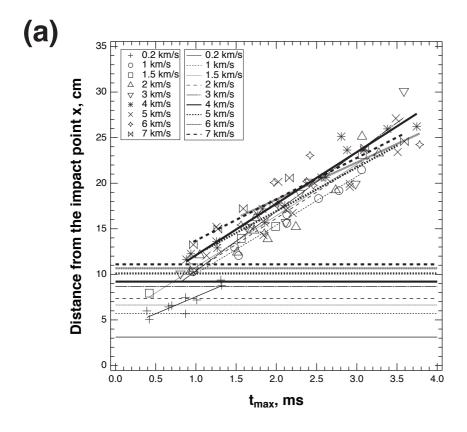


Figure 5



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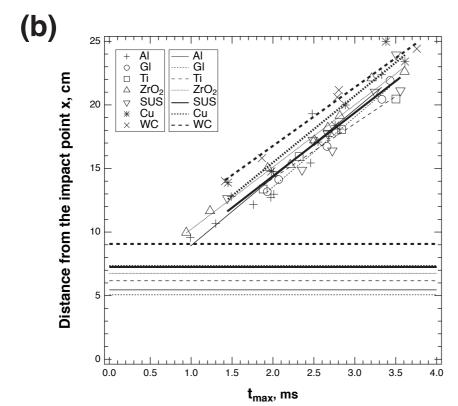


Figure 7

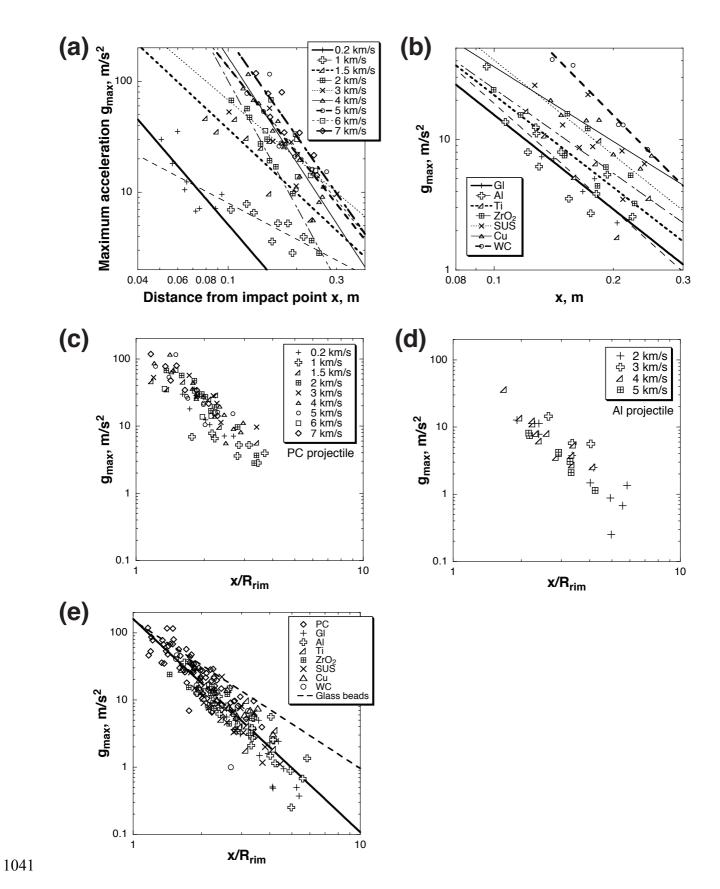
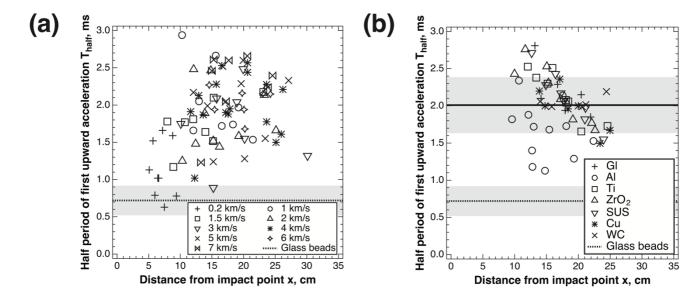


Figure 8



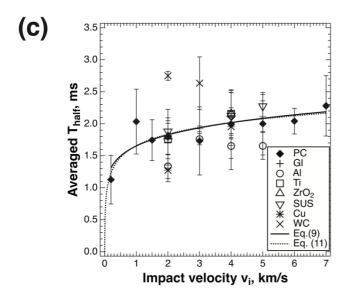
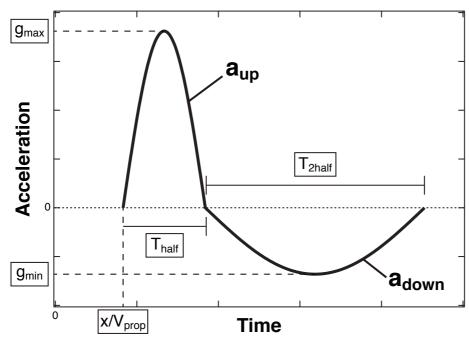
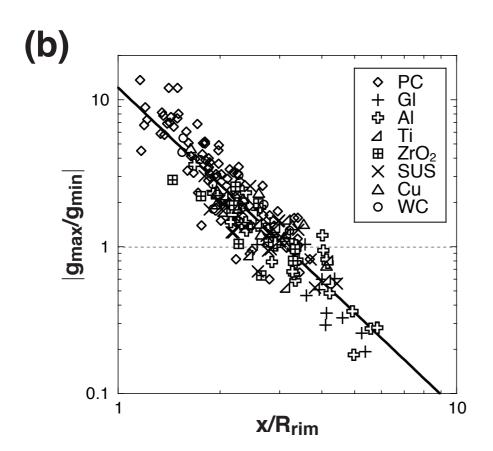


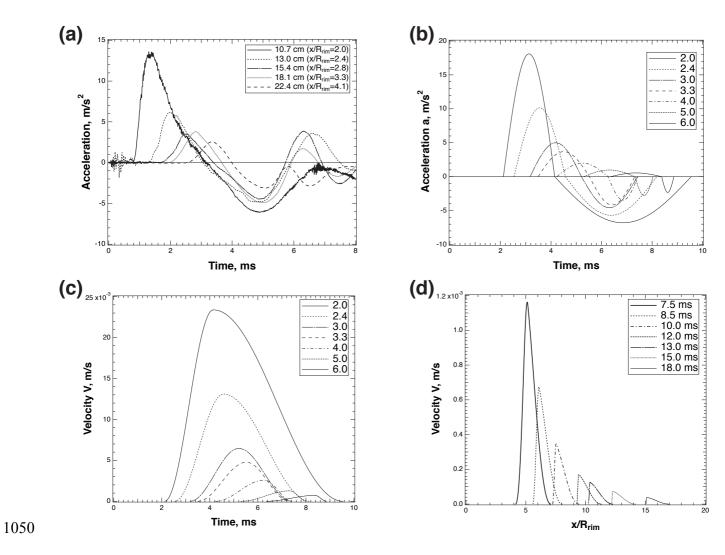
Figure 9

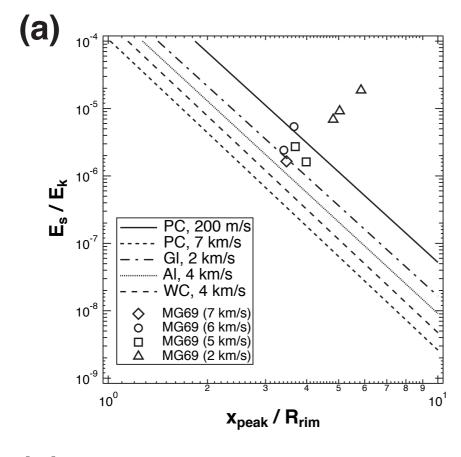


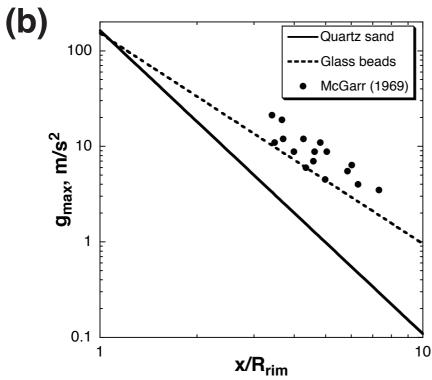




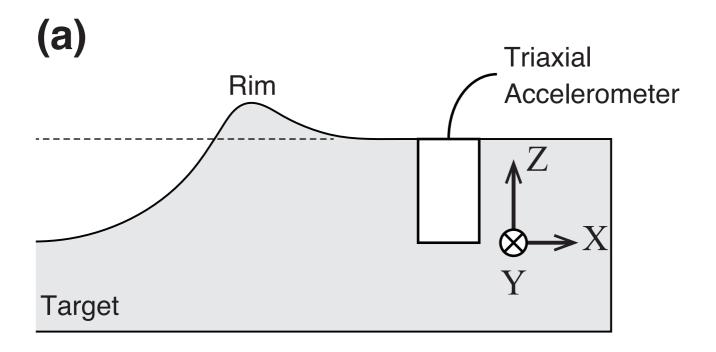
1048 Figure 10

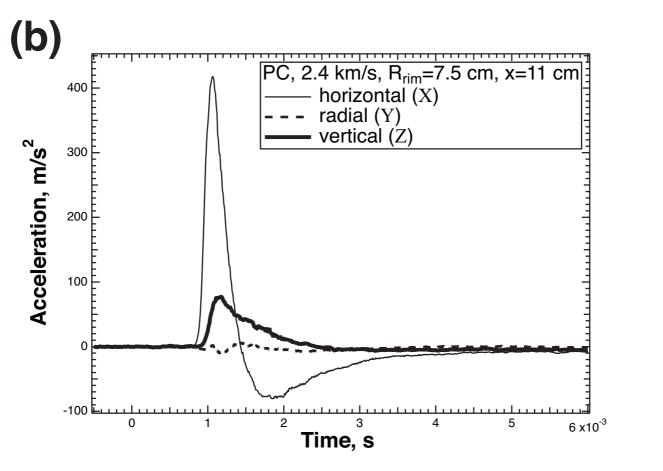




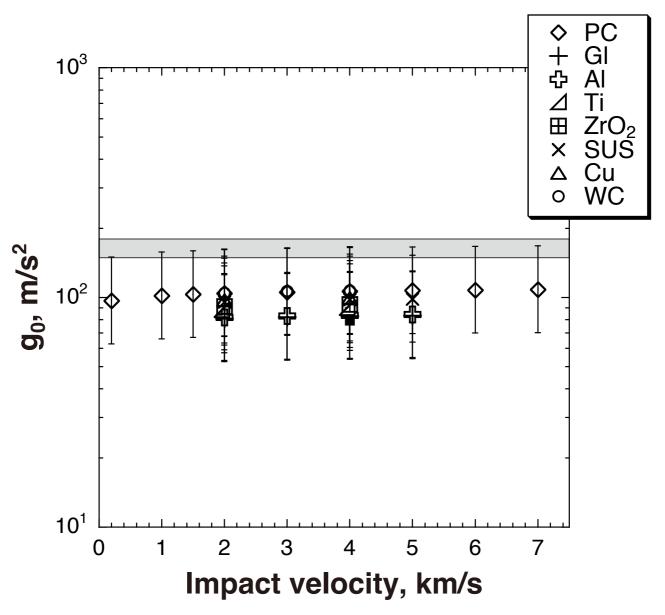


1054 Figure 12

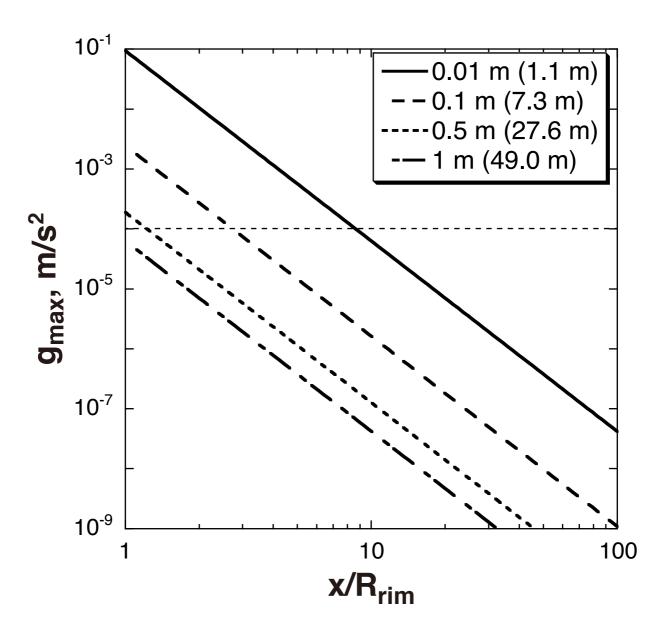




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